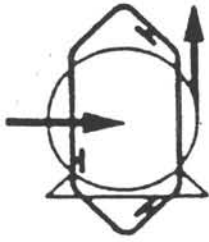


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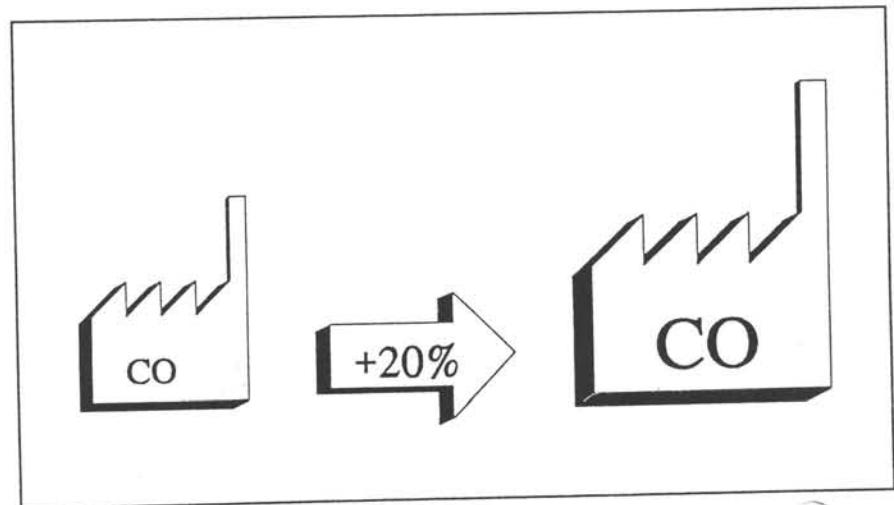
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Capacity increase of a CO-plant

Study of the capacity increase of an existing CO-plant

G-groep, february 1993



CAPACITY INCREASE OF A CO-PLANT

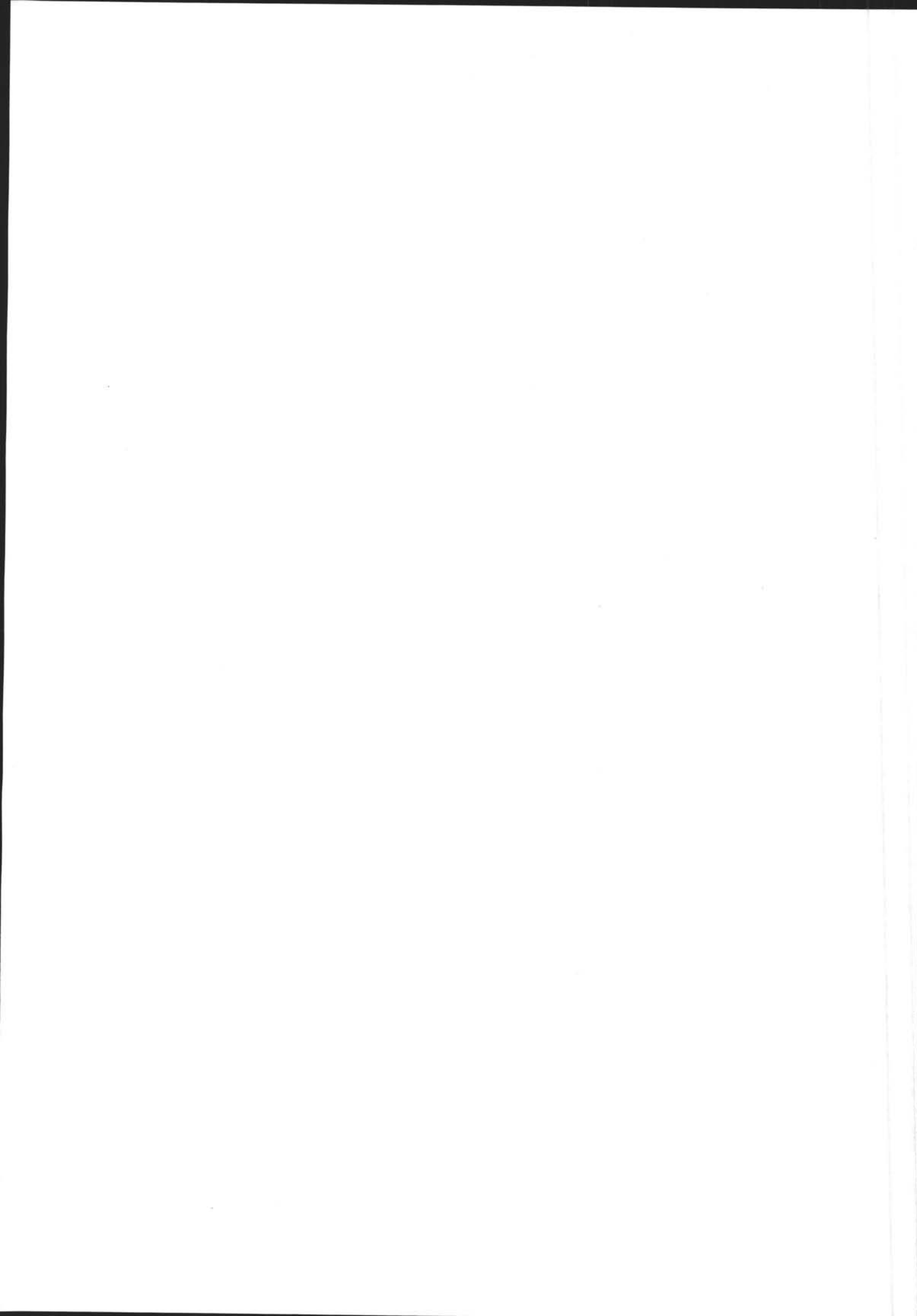
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PREFACE

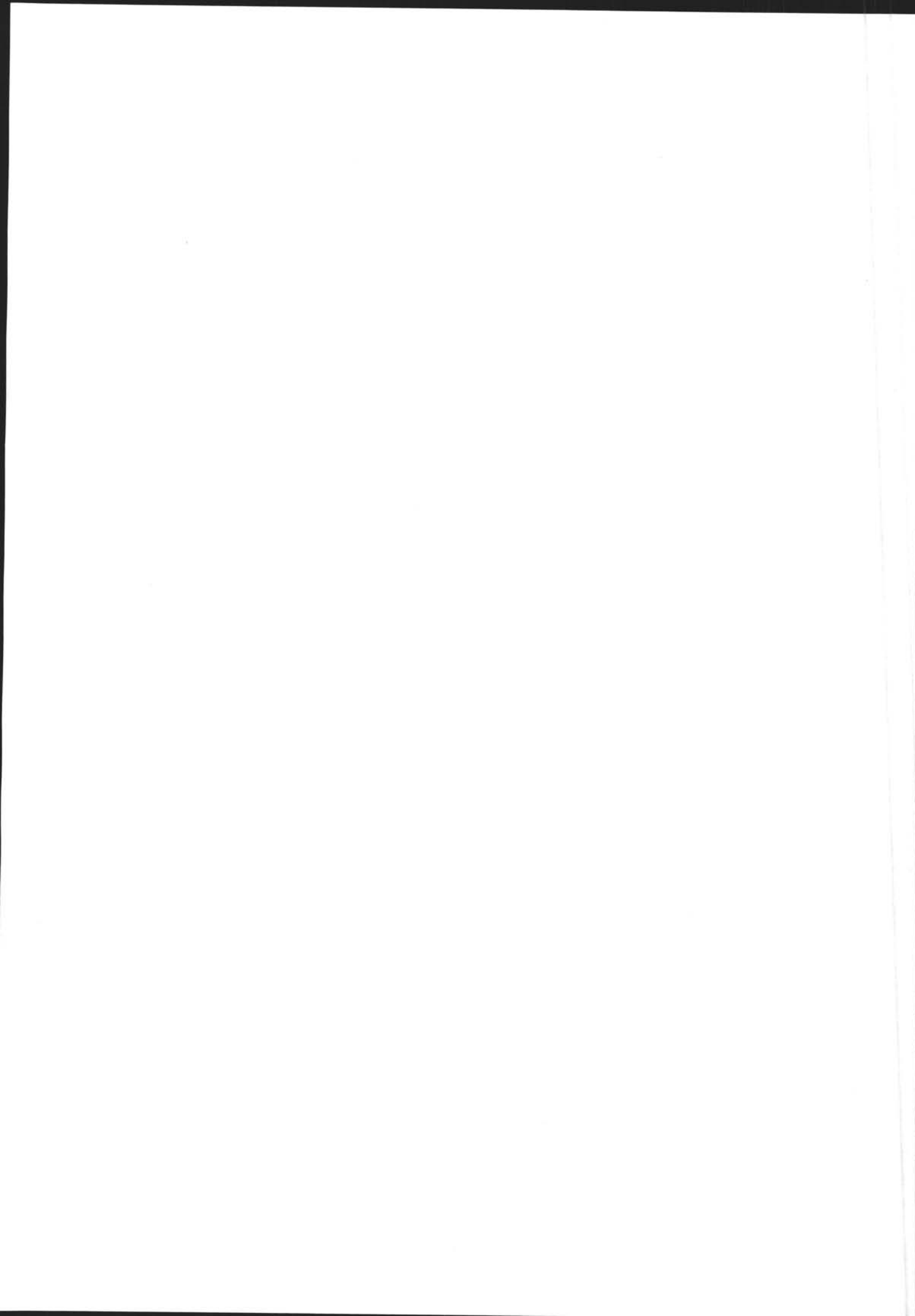
This report is the result of the G-group study at the Delft University of Technology in spring 1993. The G-group study, which lasted 12 weeks, is carried out by students mechanical- and chemical engineering and is done before graduating.

We would like to thank everybody who has helped us with the completion of this report, in particular our supervisors ing. A. de Bruin, prof. dr. ir. J. de Graauw, prof. ir. J. Grievink, ir. V. Mattheij, drs. F.A. Meijer and dr. Ž. Olujić.

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SUMMARY

This report is the result of a study of the G-Groep of february 1993. In Bergen op Zoom GE Plastics (GEP) has two plants for producing CO. This CO is an intermediate in the production of different plastics.

The purpose of this study is to accomplish a capacity increase of 20 % of the actual CO production in CO plant II. The different options to upscale the plant were handled in a pre-study. The main subjects for investigation are:

- * removing N₂ from the natural gas feed
- * changing the reformer, eventually adding a pre-reformer
- * changing the feed composition
- * investigate gas separation processes after the reformer
- * producing a simulation by using, on request of GEP, the ASPEN PLUS program
- * look for alternative CO production systems

In order to have an economically feasible method, the total group of thirteen students was divided in four groups and one ASPEN PLUS expert. After a few weeks some groups finished most of their work and some students went to other groups. The occupation of the groups was as follows:

- 1) investigate the reformer to get 20 % more CO output and give data to simulate with ASPEN.
- 2) investigate the gas separation processes after the reformer with the new streams of group one and give data to ASPEN in order to simulate the gas separation units.
- 3) investigate a possible way to remove the N₂ from the natural gas feed and analyse the advantages of this removal on the production capacity.
- 4) investigate the other ways to produce and separate CO with a totally new plant based on different technology.

The conclusion of group 1 is that an increase of the reformer capacity is possible by using an extra feedstream consisting of CO₂, eventually in combination with a pre-reformer. With this extra CO₂ an increase of CO production of approximately 20 % is possible. If the feed is for 80 % free of N₂ the increase in CO production or heatduty is neglectible.

Group 2 has calculated that a greater gasflow coming out of the reformer, in the case of only adding extra CO₂ to the reformer feed, can be treated in the stripping section of the MEA-system with addition of several heatexchangers, and that the regeneration of the MEA solution has a limit of 17 % capacity increase when operating at 90 % flooding. This 17 % capacity increase is sufficient for a total plant scale up of 20%, because the CO₂ concentration is lower in the new plant. When using a pre-reformer the equipment in the MEA section does not need to be changed. The columns in the coldbox all have sufficient overcapacity for 20% capacity increase in both cases. Pumps and compressors also handle the capacity increase in both cases.

Group 3 has simulated a plant which has a N₂ removal of 80 % out of the natural gas stream needed for 20 % more CO production. The N₂-removal is done by cryogenic

separation. The costs of such a plant are not only so high that a commercial use is not recommendable, but also the use of N₂ free natural gas as feedgas has no use for higher CO production, according the conclusion of group 1.

Group 4 has made a study for a total new plant with the same CO production as the scaled up existing plant and this new plant consists of a CO₂ reformer with a CO Vacuum Swing Adsorption (VSA) system for the CO recovery. The new plant makes no use of the unit for N₂ free natural gas feed, because of cost calculations. The CO in this plant is produced for \$ 0.60/kg. Also the costs for a COSORB (CO-complex adsorption system) and a H₂ fuel cell system, for electricity generation out of the H₂-rich stream of the H₂-PSA, are calculated.

Coal gasification would only be economical producing more then 6.5 ton/hr, which is far more then 1.22 ton/hr and therefore not an option.

Working with ASPEN gave a lot of difficulties but most of them were problems with our computernetwork. The program itself has been very usefull to calculate the new streams and all physical properties of the stream.

The upscaled plant will have a CO price of ± \$0.88 which is less expensive than the current CO price.

The costs for one ton CO with the total new process designed by group 4 is \$ 0.60. So the conclusion is when a total new plant will be made a good alternative will be the process of group 4.

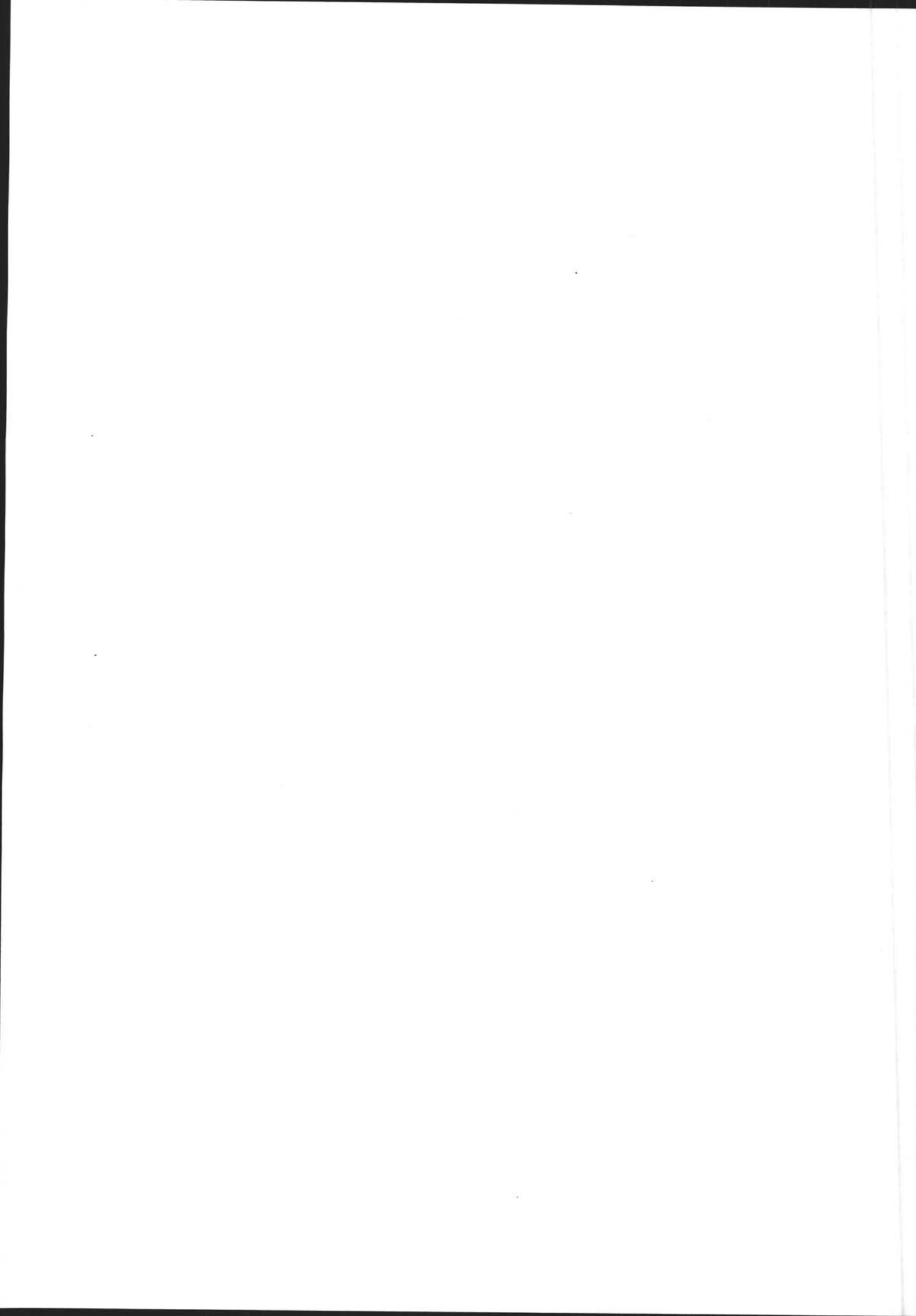
Some further attention can be given to cost calculations, which are only an indication of the new CO production costs.

1. INTRODUCTION

In Bergen op Zoom General Electric Plastics produces carbon monoxide in CO-plant I and CO-plant II. The carbon monoxide is produced by steam reforming of natural gas over a nickel based catalyst. After reforming the process gas contains many components: CO, CO₂, CH₄, H₂, H₂O and N₂, so the next step in the production of carbon monoxide is purification of the process gas. The purification steps are carbon dioxide removal, drying and cryogenic separation of methane, carbon monoxide and hydrogen.

The goal of this G-group is to investigate the possibility to increase the capacity of CO plant II by twenty percent, using Aspen for the simulations. Hence the capacity has to be increased from 1,26 tons per hour to 1.51 tons pure CO per hour.

A short description of the current proces is given in chapter two. In chapter three the reformer, which is the main bottle neck for capacity increase, is investigated. The way the plant is modelled in Aspen is described in chapter four. After the modelling, we have used Aspen to investigate the different influences on the production capacity, hence chapter five deals with the simulation and calculation of the streams in the CO-plant. In chapter six the removal and recovery of CO₂ with MEA is discussed. After the CO₂ separation the process gas is purified to CO and N₂ by means of cryogenic separation; this is described in chapter seven. Because natural gas is used as feed in the current process, we have to deal with N₂ in the process gas. So in chapter eight advantages and methods for N₂ removal are studied. In chapter nine the total costs are estimated. Alternatives for the production of CO are discussed in chapter ten. Recommendations and conclusions for the capacity increase of CO-plant II are given in chapter eleven. Finally the experiences with ASPEN are described in chapter twelve.



2. THE CURRENT PROCESS

In this chapter CO plant II will briefly be discussed, in order to understand the production process of CO. The information of the plant was released by General Electric Plastics [2.1]. If we summarize the information we get a block scheme with design main stream-numbers, which can be found in appendix 2. With the aid of this block scheme the process is described. Appendix 2 is completed with the KTI process flow diagrams of CO plant II.

2.1 Feed of the plant

The feed which is used for the production of CO is Dutch natural gas. The properties of this gas can be found in a book published by the Nederlandse Gasunie [2.2]. The composition of this gas is given in table 2.1.

table 2.1: Average composition of natural gas

Component	Formula	% (mol)	% (mass)
methane	CH ₄	81.29	69.97
ethane	C ₂ H ₆	2.87	4.63
propane	C ₃ H ₈	0.38	0.90
butane	C ₄ H ₁₀	0.15	0.47
pentane	C ₅ H ₈	0.04	0.16
hexane	C ₆ H ₁₄	0.05	0.23
nitrogen	N ₂	14.32	21.52
oxygen	O ₂	0.01	0.02
carbon dioxide	CO ₂	0.89	2.10
		-----	-----
		100	100

2.2 Desulphurisation of the natural gas

Since sulphur poisons the reformer catalyst, it is necessary to remove the sulphur from the natural gas. If no recycled hydrogen is added to the natural gas feedstream then the gas stream goes to a carbon bed desulphurizer.

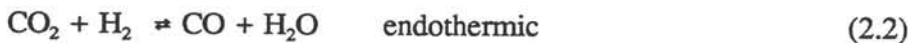
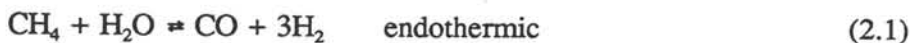
At normal production conditions however recycled hydrogen is used. Then the gas stream follows another desulphurisation route; the feed is transported through a cobalt-molybdenum catalyst bed (COMOX). In this bed the sulphur compounds are converted in H₂S. In a zinc oxide reactor H₂S is adsorbed.

2.3 Reforming of natural gas

The feedstream to the reformer exists of four separate streams:

- Desulphurized natural gas with or without recycled hydrogen.
- Recycled CO₂ from the CO₂ removal section.
- Recycled tail gas from the CO recovery section.
- Process steam.

The feedstream enters at a temperature of 487 °C and is sent into tubes which are packed with a catalyst. The chemical reactions which occur in the reformer can be described by two main reactions:



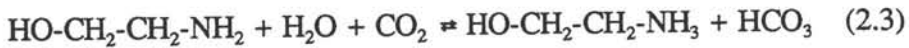
The overall reaction is endothermic and favoured at high temperature. The energy needed for the reactions is provided by side wall mounted burners. The heat is transferred by radiation and convection to the process gas in the catalyst tubes. The flue gasses are used for pre-heating of the feed and steam production. The burners are started up with natural gas and when H₂-rich gas from the CO recovery unit is available natural gas is replaced by the H₂-rich gas.

At the reformer outlet (temperature 860 °C) the methane equilibrium is approached to 20 °C lower, so the equilibrium constant of 840 °C is taken. The CO equilibrium is approached to 0 °C , so the equilibrium constant of 860 °C is taken. Hence with known outlet pressure, temperature and feed composition it is possible to calculate the reformer outlet composition.

In the reformer the steam to carbon ratio should stay above a certain minimum value because of the carbon deposition tendencies. This is the reason why an excess of steam is added. The reformer outlet temperature is decreased in two cooling steps to a temperature of 40 °C. The condensated steam is almost completely separated from the process gas.

2.4 Process gas purification

In the MEA section the CO₂ is removed from the process gas. The process gas is counter currently transported through an absorption column with a monoethanolamine (MEA) stream at a temperature of 40 °C. The CO₂ reacts with MEA according to the next reaction:



The CO₂ rich MEA solution is transported to the MEA stripper. The CO₂ can be released from the solution at low pressure and a temperature of 120 °C. The CO₂ stream flows to the compressor where it is compressed and recycled to the reformer.

The process gas is compressed in a compressor and transported to the process gas drying unit where the temperature is reduced to 35 °C. The formed condensate is separated by a gas liquid separator. In the last step of the drying unit the process gas is transported to a set of mole sieve beds where the last bit of H₂O is removed. The process gas leaving the drying unit contains H₂O and CO₂ at a level of 0.1 ppm volume. This ratio is necessary to avoid clogging problems in the cold box.

The process gas coming from the drying unit contains H₂, CO, N₂ and CH₄. To separate this gas, cryogenic separation is used (Linde Coldbox). This separation method is based on the difference in boiling points. The process gas is cooled to -200 °C by heat exchangers. For this process, the required refrigeration can be obtained by H₂ expansion and a CO recycle loop. A disadvantage of the cryogenic process is that the nitrogen present in the process gas can not be completely separated from the carbon monoxide, because of the close physical properties (boiling point and molecular weight). At the end the process gas contains CO and N₂ in a 9:1 ratio.

Literature

- [2.1] Design and operating manual carbon monoxide plant II, General Electric Plastic, KTI.
- [2.2] Geerssen, T.M., Physical properties of natural gases. Groningen, 1988.

3.THE REFORMER

3.1.Theory

3.1.1.The reformer

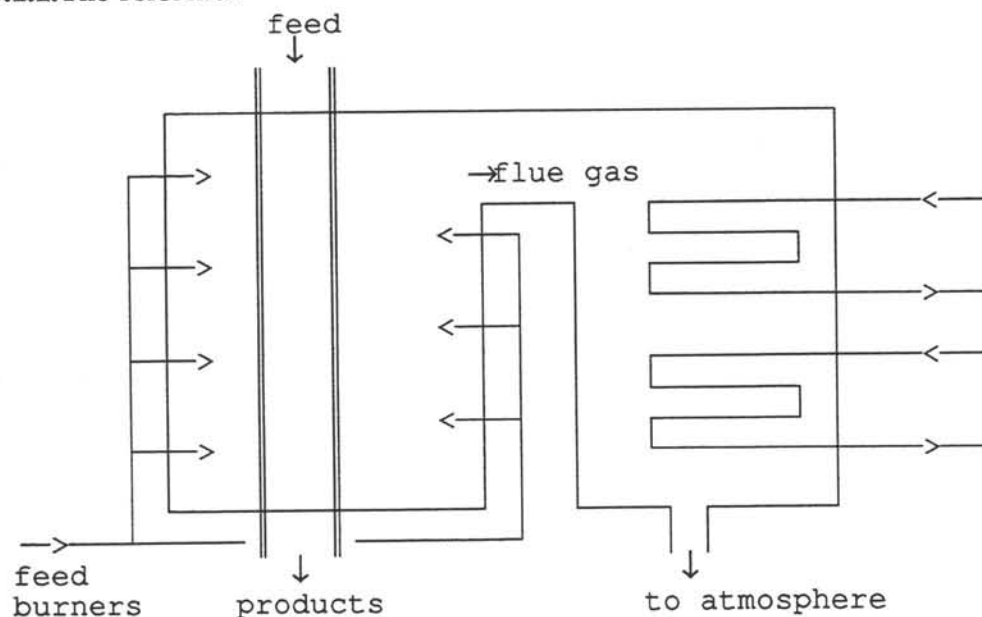


Figure 3.1. A schematic representation of the reformer.

The reformer consists of 20 tubes, which are filled with a catalyst. The tubes are suspended in a furnace which are heated with side wall mounted burners. The hot flue gas leaving the furnace is used to heat other streams.

When the CO plant is started up the burners are fed with methane. When the plant is in operation enough hydrogen is produced to feed the burners. Methane and hydrogen are fed to the burners with an excess of oxygen. The reactions taking place are:



Table 3.1 lists 11 reactions [3.1] which in theory may occur between CH_4 , CO_2 , CO , H_2 , H_2O and carbon. Whether the reactions occur depends on the composition of the gas by a given pressure and temperature. Reactions 3.3 to 3.7 are mutually depended. Therefore only 2 of these reactions are needed to calculate the conversion of methane and carbon dioxide. In literature usually reactions 3.3 and 3.4 are used to calculate the conversion, because these reactions give the best understanding in what happens in the reformer. Reactions 3.8 to 3.13 are carbon deposition reactions which are unfavourable in the reformer.

Table 3.1. Possible reactions in methane steam reforming.

number	reactions	ΔH_{298} KJ/mol
3.3	$\text{CH}_4 + \text{H}_2\text{O} \rightleftharpoons \text{CO} + 3\text{H}_2$	+ 206.1
3.4	$\text{CO} + \text{H}_2\text{O} \rightleftharpoons \text{CO}_2 + \text{H}_2$	- 41.15
3.5	$\text{CH}_4 + 2\text{H}_2\text{O} \rightleftharpoons \text{CO}_2 + 4\text{H}_2$	+ 165.0
3.6	$\text{CH}_4 + \text{CO}_2 \rightleftharpoons 2\text{CO} + 2\text{H}_2$	+ 247.3
3.7	$\text{CH}_4 + 3\text{CO}_2 \rightleftharpoons 4\text{CO} + 2\text{H}_2\text{O}$	+ 330.0
3.8	$\text{CH}_4 \rightleftharpoons \text{C} + 2\text{H}_2$	+ 74.82
3.9	$2\text{CO} \rightleftharpoons \text{C} + \text{CO}_2$	- 173.3
3.10	$\text{CO} + \text{H}_2 \rightleftharpoons \text{C} + \text{H}_2\text{O}$	- 131.3
3.11	$\text{CO}_2 + 2\text{H}_2 \rightleftharpoons \text{C} + 2\text{H}_2\text{O}$	- 90.13
3.12	$\text{CH}_4 + 2\text{CO} \rightleftharpoons 3\text{C} + 2\text{H}_2\text{O}$	- 187.6
3.13	$\text{CH}_4 + \text{CO}_2 \rightleftharpoons 2\text{C} + 2\text{H}_2\text{O}$	- 15.3

When natural gas is used as a feed higher alkanes are also present. The reactions of these alkanes are listed in Table 3.2. The most important reaction of Table 3.2 is the carbon deposition reaction (3.16); higher alkanes have a higher potential for carbon deposition than lower alkanes.

Table 3.2. Possible reactions in methane steam reforming when higher alkanes are present.

number	reactions	ΔH_{298} KJ/mol
3.14	$\text{C}_n\text{H}_{2n+2} + (n-1)\text{H}_2 \rightleftharpoons n\text{CH}_4$	-
3.15	$\text{C}_n\text{H}_{2n+2} + n\text{H}_2\text{O} \rightleftharpoons n\text{CO} + (2n+1)\text{H}_2$	+
3.16	$\text{C}_n\text{H}_{2n+2} \rightleftharpoons n\text{C} + (n+1)\text{H}_2$	+

Equilibrium constants

All reactions in Table 3.1 and 3.2 are equilibrium reactions. To predict the composition of the process gas at a given temperature and pressure, the equilibrium constants must be calculated. Therefore the equations have to be solved simultaneously. The equilibrium constants of the most important reactions, reactions 3.3 and 3.4, can be calculated with the following equations [3.2]:

for reaction 3.3

$$K_3 = \exp\left[-\frac{27464}{T} + 30.707\right] \quad (3.1)$$

and for reaction 3.4 below 773 K

$$K_4 = \exp\left[\frac{4578}{T} - 4.33\right] \quad (3.2)$$

and for reaction 3.4 above 773 K

$$K_4 = \exp\left[\frac{4084}{T} - 3.765\right] \quad (3.3)$$

with

T : the equilibrium temperature (K)

K₃: the equilibrium constant for reaction 3.3 (atm²)

K₄: the equilibrium constant for reaction 3.4 (-).

With the K-values now calculated and the feed composition known, the composition can now be calculated with the following equations, assuming that no carbon deposits and that all higher alkanes are converted into methane.

$$K_3 = \frac{P_{CO} * P_{H_2}^3}{P_{CH_4} * P_{H_2O}} \quad (3.4)$$

and

$$K_4 = \frac{P_{H_2} * P_{CO_2}}{P_{CO} * P_{H_2O}} \quad (3.5)$$

where P_i is the partial pressure of the components which can be calculated with the following equation.

$$P_i = \frac{n_i}{n_{tot}} * P \quad (3.6)$$

with

n_i : number of moles of component i

n_{tot} : total number of moles

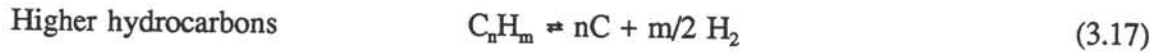
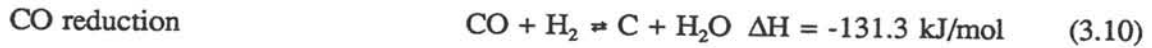
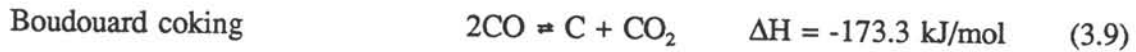
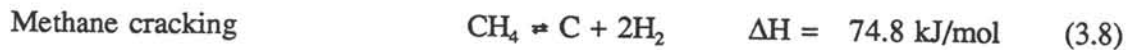
P : pressure (Pa)

3.1.2. Carbon-free operation of the reformer

Carbon formation in a steam reformer will reduce the activity of the catalyst. As a result of the carbon deposition the rate of heat transfer decreases, resulting in higher temperatures of the reformer tubes. These higher temperatures will eventually lead to overheating. This overheating will increase the carbon formation, resulting in a self-accelerating situation and breakdown of the catalyst.

In the plant the steam to carbon ratio must be optimized. Steam favours the methane conversion, prevents carbon formation but too much steam is not economical.

Carbon formation can take place by the following reactions [3.3]:



sub reaction (3.8). The methane cracking reaction is endothermic and exhibits a volumetric increase. Hence it is favoured by high temperatures and low pressures. Because methane is consumed and hydrogen is produced in the reformer, the methane cracking reaction will most likely take place near the reactor inlet, where methane is still present. Temperatures at the reformer outlet will thermodynamically favour cracking, but then there is so much hydrogen and so little methane, that there is no affinity for the coking reaction.

sub reaction (3.9) and (3.10). Both the Boudouard and CO reduction reactions are exothermic and result in a volumetric decrease. Hence these reactions are thermodynamically favoured by low temperatures and high pressures. Since there is no CO at the inlet of the reformer, both the Boudouard and the CO reduction reactions can not take place even though they are thermodynamically favoured. A region near the reformer inlet can however exist, where temperatures are low enough and concentrations of CO and H₂ high enough to enable these reactions to thermodynamically take place.

sub reaction (3.17). Higher hydrocarbons are prone to carbon formation. If not reformed in the upper part of the tubes, they will crack to carbon further down. Since the amount of higher hydrocarbons in natural gas is low and due to the saturated character of these hydrocarbons, we have not evaluated them separately. The hydrocarbons will crack due to the activity of the catalyst, so they will be considered to be CH₄. The critical steam to carbon ratio will be taken more conservative in the beginning of the reformer to cope with the carbon-forming potential of the higher hydrocarbons.

From the reactions 3.9 and 3.10 it can be seen that H₂O and CO₂ will remove the deposited carbon. To optimize the flow in the reformer it is necessary to consider the risk of carbon formation.

The risk of carbon formation is evaluated by means of " The principle of equilibrated gas ". [3.4]. This principle states: " Carbon formation is to be expected if the gas shows affinity for

carbon after the establishment of the methane reforming and the shift equilibria ". The principle is no law of nature, it's merely a rule of thumb which results in a steam to carbon ratio that is rather conservative. Since the principle is a rule of thumb, a total analysis of the reformer is necessary for complete safety. This analysis should contain kinetic factors of carbon formation and gasification, radial temperature profiles, catalyst activity etc.. Since such an analysis is very complex it will not be carried out in this report.

With known feed, K-values, temperatures and pressures in the reformer it is possible to calculate the equilibrated gas composition. With this calculated composition it is possible to calculate the potential for coke formation.

Potential for carbon formation:

$$-\Delta G_e = R * T * \ln\left(\frac{K_{3,i}}{Q_e}\right) \quad (3.7)$$

with

$K_{3,i}$: equilibrium constant for graphite formation for reaction i (i = 8 or 9)

Q_e : $P_{H_2}^2/P_{CH_4}$ or P_{CO_2}/P_{CO}^2 (K values of reactions 3.8 and 3.9 of the equilibrated gas)

no potential for carbon formation results in $\Delta G_e > 0 \rightarrow Q_e > K$

hence:

$$P_{H_2}^2/P_{CH_4} > K_{3,8}$$

$$P_{CO_2}/P_{CO}^2 > K_{3,9}$$

The CO reduction reaction does not have to be considered, because the shift reaction is at equilibrium inside the pellets and therefore its affinity for coking will be the same as the Boudouard reaction. $K_{3,8}$ and $K_{3,9}$ are given in figure 3.2 [3.5].

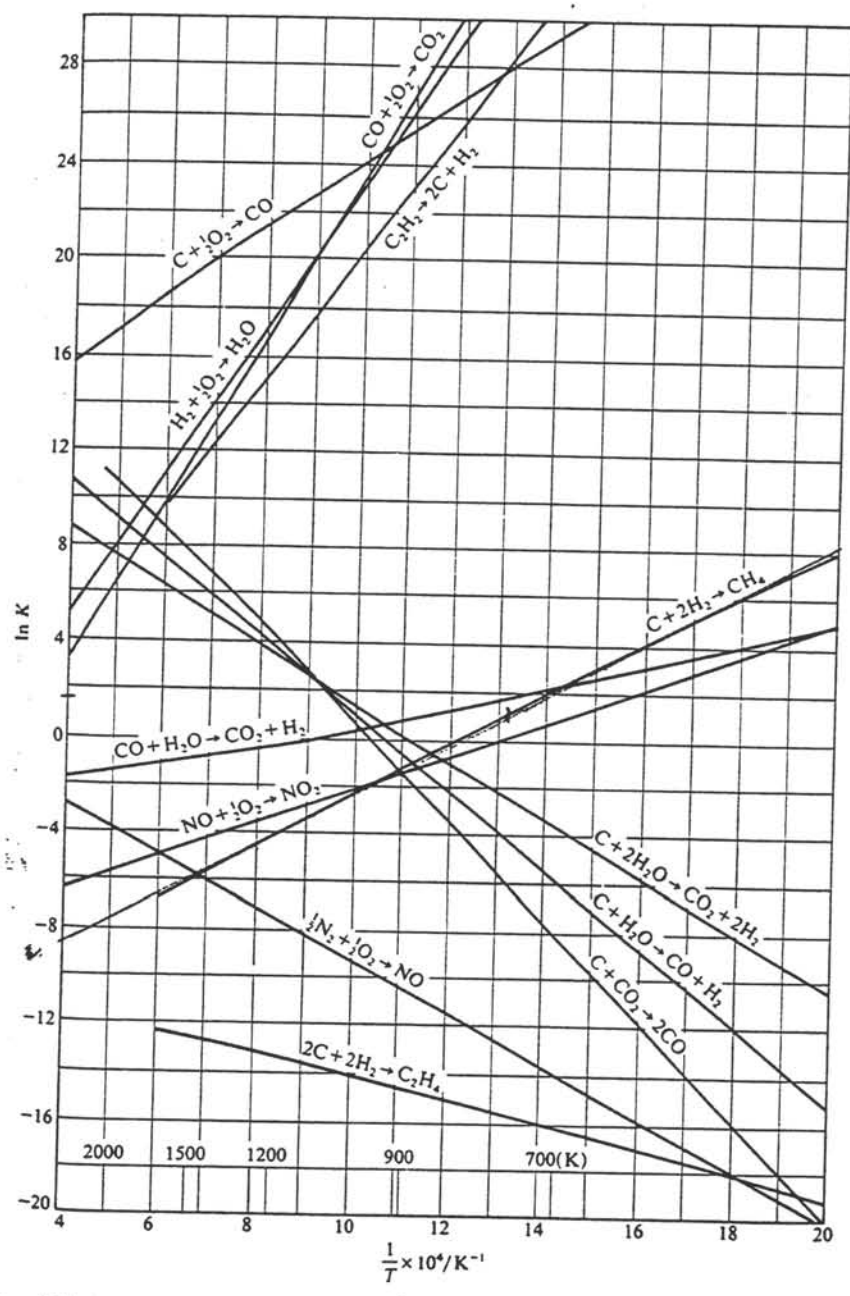


Figure 3.2. Equilibrium constants as a function of temperature for some common reactions.

3.1.4. Necessary energy input in the reformer tubes.

Introduction.

Heat transfer in the reformer is important. Therefore we want to know how much energy is approximately needed to heat the gas and to let the reactions take place. The extent of conversion of the two most important reactions can be calculated with the Mercury program, given in appendix 3.2. In this program the higher alkanes are assumed to be immediately converted into methane before participating in the process. After the higher alkanes being converted, the total energy needed for the (real) conversions and heating can be calculated.

Hydrogen and air are burned to supply the energy, needed for the reactions in the reformer. The feed is preheated to 760.15 K. It then enters the reformer, where the higher alkanes are converted into CH_4 , according to reaction (3.14). These reactions are endothermic. We have to devise a path, for purposes of calculation, to proceed from reactants (here the higher alkanes) at 760.15 K to product (here CH_4), also at 760.15 K. Since data are available for the standard heats of reaction at 298.15 K, the most convenient path is the one which includes the reactions at 298.15 K.

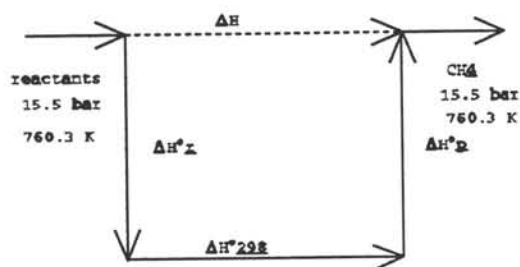


Figure 3.3. Path for calculation first reactions

The dashed line represents the actual path, with enthalpy change ΔH . Since this enthalpy change is independent of the path chosen, equation (3.8) is valid.

$$\Delta H = \Delta H^{\circ}_R + \Delta H^{\circ}_{298} + \Delta H^{\circ}_P \quad (3.8)$$

For the calculation of ΔH°_{298} , reaction (3.14) has to be considered. The calculation of ΔH°_P means that first of all we have to convert a real gas into an ideal gas, see equation 3.14. This enthalpy change is H^R . Secondly we have to "cool" the now ideal gas at 760.15 K to 298.15 K. For ΔH°_P it is the other way around, which means that after the reaction, the product CH_4 is "warmed" as an ideal gas, and then made a real gas at 15.5 bar and 760.15 K.

Approximate results of H^R for mixtures can often be obtained with critical parameters for the mixture and a simple linear mixing rule for the acentric factor. Since values for the actual critical properties T_c and p_c for mixtures are rarely known, the pseudoparameters T_{pc} and p_{pc} are used, determined again by a simple linear mixing rule. Thus by definition,

$$T_{pc} = \sum y_i * T_{ci} \quad (3.9)$$

$$P_{pc} = \sum y_i * P_{ci} \quad (3.10)$$

$$\omega = \sum y_i * \omega_i \quad (3.11)$$

The pseudoreduced temperature and pseudoreduced pressure, which replace T_r and p_r , are determined by

$$T_{pr} = \frac{T}{T_{pc}} \quad (3.12)$$

and

$$P_{pr} = \frac{P}{P_{pc}} \quad (3.13)$$

Thus, for a mixture at a given T_{pr} and p_{pr} we may determine a value of H^R using the following equation:

$$\frac{H^R}{R * T_{pc}} = P_{pr} * [B^0 - T_{pr} * \frac{dB^0}{dT_{pr}} + \omega * (B^1 - T_{pr} * \frac{dB^1}{dT_{pr}})] \quad (3.14)$$

$$B^0 = 0.083 - \frac{0.422}{T_{pr}^{1.6}}$$

$$B^1 = 0.139 - \frac{0.172}{T_{pr}^{4.2}}$$

$$\frac{dB^0}{dT_{pr}} = \frac{0.675}{T_{pr}^{2.6}}$$

$$\frac{dB^1}{dT_{pr}} = \frac{0.722}{T_{pr}^{5.2}}$$

After this the enthalpy change has to be calculated after "cooling" the stream from 760.15 K to 298.15 K via the ideal path. This can be done with equation (3.15):

$$\Delta H^{ig} = (\sum n_i * c_{p_{m,i}}) * (T_2 - T_1) \quad (3.15)$$

Here n_i is the number of moles of each higher alkane that enters the reformer in one hour. This because the flows are given in moles per hour, and for the SI unit of enthalpy only mole-numbers are of interest here.

$$c_{p_{m,i}} = R * (A_i + B_i * T_{am} + \frac{C_i}{3} * (4 * T_{am}^2 - T_1 * T_2) + \frac{D_i}{T_1 * T_2}) \quad (3.16)$$

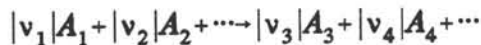
with

$$T_{am} = \frac{T_1 + T_2}{2}$$

So now ΔH°_R is known:

$$\Delta H^\circ_R = \Delta H^R + \Delta H^{ig}$$

In general a chemical reaction can be written as



where $|v_i|$ are stoichiometric coefficients and A_i chemical formulas. The species on the left are reactants; those on the right are products. We adopt a sign convention for v_i that makes it

positive (+) for products

negative (-) for reactants

The v_i with their accompanying signs are called stoichiometric numbers. So in reaction (3.3)

$$v_{CH_4} = -1 \quad v_{H_2O} = -1 \quad v_{CO_2} = 1 \quad v_{H_2} = 3$$

This sign convention allows the definition of a standard heat of reaction to be expressed mathematically by the equation:

$$\Delta H_{298}^\circ = \sum v_i * H_{298}^\circ_i \quad (3.17)$$

where H°_i is the enthalpy of species i in its standard state and the summation is over all products and reactants. The standard-state enthalpy of a chemical compound is equal to its heat of formation plus the standard-state enthalpies of its constituent elements. If we arbitrarily set the standard-state enthalpies of all elements equal to zero as the basis of calculation, then the standard-state enthalpy of each compound is its heat of formation. In this event, $H_{298,i}^\circ = \Delta H_{f,i}^\circ$ and (equation 3.18) becomes

$$\Delta H^{\circ}_{298} = \sum v_i \cdot \Delta H^{\circ}_{f,i} \quad (3.18)$$

where the summation is over all products and reactants. We here assume that the higher alkanes are all converted in CH_4 at a temperature of 760.15 K.

After this the enthalpy change for the product CH_4 , can be calculated. When this is "heated" and "made" a real gas, the same formulas can be used as for the reactants. Only T_1 and T_2 are switched. Now also ΔH^P can be calculated.

$$\Delta H^{\circ}_P = \Delta H^P + \Delta H^{ig} \quad (3.19)$$

Because the flow of these higher alkanes is relatively small, it can be expected that the total heat needed so far is not very much, but for significant reasons we must look at this as well.

The CH_4 formed by these reactions can be added to the amount of CH_4 already available in the feed. At this point we are able to look at the main reactions occurring in the reformer and which are mainly of interest for the heat balance of the reformer. Figure 3.4 represents the way we modelled the problem, that is, how to calculate the required enthalpy change ΔH between entering gas and leaving gas.

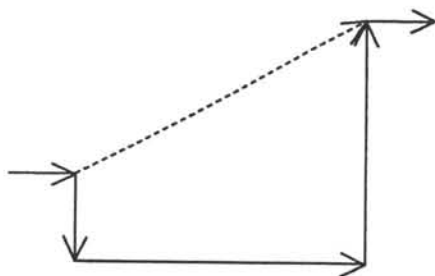


Figure 3.2. Path of main reactions

The calculations are about the same as those for the higher alkanes, that is why the formulas mentioned above can be used. This means that we also have to convert the real gas into an ideal gas, then "cool" to 298.15 K, let the reactions take place, and "warm" the products formed here to 1133.15 K. The calculations are performed with the program *mathcad*. This can be found in appendix 3.1.

3.1.4. The overall heat transfer coefficient.

The heat transfer in the reformer from the burner gas through the tubes to the process gas is limiting. In order to produce more carbon monoxide there are, in case of the reformer, three possible ways to do so. The first and most obvious option is a change in the reformer feed. The question then is: when will the heat transfer become limiting. The second option is to change the material the tubes are made of. The better the tube material conducts heat or the higher the service temperature of the tube, the more carbon monoxide can be produced. The third option is a combination of the two options mentioned above.

For the calculation of the transferred heat, equation (3.20) is used.

$$\Phi_w = U * A * \Delta T_{\ln} \quad (3.20)$$

with

- Φ_w : Transferred heat (= heat or energy flow) (W)
- U : Overall heat transfer coefficient (W/m²/K)
- A : Outside tube wall area for heat transfer (m²)
- (ΔT)_{ln} : Logarithmic mean temperature difference (K)

The overall heat transfer coefficient expresses the combined effect of the whole series of resistances through which the transferred heat must flow. Generally there are three resistances. First of all the resistance for heat transfer from the outside fluid to the outside tube wall. Secondly the resistance for heat transfer through the tubes. Finally the third resistance will be from the inside tube wall to the inside fluid, see equation (3.21).

$$\frac{1}{U} = \frac{D_i}{D_u * h_u} + \frac{D_i}{2 * k_w} * \ln \frac{D_u}{D_i} + \frac{1}{h_i} \quad (3.21)$$

with

- U : Overall heat transfer coefficient
- D_u : External tube diameter
- D_i : Internal tube diameter
- h_i : Internal heat transfer coefficient
- h_u : External heat transfer coefficient
- k_w : Conductivity of tube material

In this particular case the outside tube wall temperature is known. This means that the outside resistance can be omitted.

In literature an empirical relation for the resulting equation for the calculation of the overall heat transfer coefficient was found, see equation (3.22) [3.6].

$$\frac{1}{U} = \frac{D_i}{2 * \lambda_{st}} * \ln \frac{D_{te}}{D_i} + \frac{1}{a_w} + \frac{D_i}{8 * \lambda_{er}} \quad (3.22)$$

with

$$a_w = \frac{8.694 * \lambda_{er}^{\circ} + \frac{0.512 * \lambda_g * D_i * Re_p * Pr^{\frac{1}{3}}}{d_p}}{D_i^{\frac{4}{3}}} \quad (3.23)$$

$$\lambda_{er} = \lambda_{er}^{\circ} + \frac{0.111 * \lambda_g * Re_p * Pr^{\frac{1}{3}}}{1 + 46 * \frac{d_p^2}{D_i^2}} \quad (3.24)$$

As can be seen from equation (3.22), the overall heat transfer coefficient only deals with the resistances to heat transfer of the tube material, the process gas and the catalyst particles. This means that the "black box" considerably reduces. On the other hand, this implies that the outside tube wall temperature as well as the outside tube area have to be known. Indeed this is the case. Both parameters can be found or be calculated from the data given by KTI [3.7]. The logarithmic mean temperature difference, better known as the driving force for heat transfer, is given by equation (3.25).

$$\Delta T_{\ln} = \frac{(T1 - t1) - (T2 - t2)}{\ln \frac{T1 - t1}{T2 - t2}} \quad (3.25)$$

3.1.5. Possible ways to increase the CO-plant capacity

The heat transfer in the reformer tubes is the limiting factor. Therefore other feeds or other pieces of equipment must be used to increase the CO production.

3.1.5.1. Less N_2 in the feed

One possible way to increase the CO production is to reduce the amount of N_2 in the feed. Because N_2 is an inert gas, the energy needed to heat the nitrogen in the reformer tubes are pure wastes of energy. By removing the nitrogen this energy can be used for reaction energy, needed for the production of CO.

3.1.5.2. Using a pre-reformer

A possible way to increase the CO production is to put a pre-reformer ahead of the reformer. A pre-reformer is an adiabatic reactor [3.9], which converts all the higher hydrocarbons into a mixture of carbon oxides and hydrogen. The methane reforming and shift reactions are taking place to chemical equilibrium. A pre-reformer operates with a inlet temperature of 380-520 °C. Since the pre-reformer is an adiabatic reactor the temperature will decrease about 50 °C. After the pre-reformer the gas is heated again and sent to the reformer. A pre-reformer is packed with several types of catalysts. In top, where there is no risk of carbon formation, an alkali free catalyst is used (15%), then an intermediate alkali promoted catalyst (25%) and at the bottom again an alkali free catalyst is used (65%). A schematic arrangement of a pre-reformer in combination with a reformer is given in figure 3.5.

Case "C"
Installation of
adiabatic Pre-
Reformer with
heat of the
process gas both
before and after
the Pre-Reformer

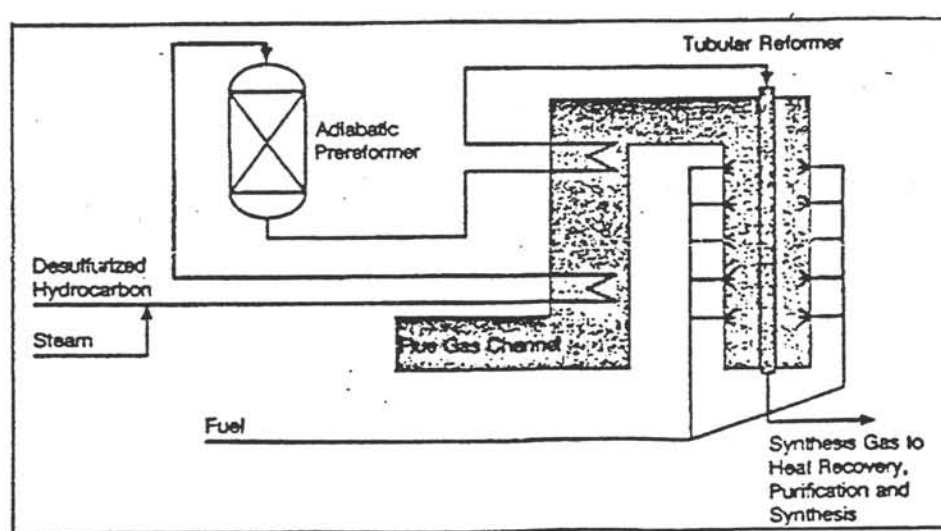


Figure 3.5. A schematic representation of a pre-reformer in a plant.

The advantages and disadvantages resulting from installation of a pre-reformer are summarized below:

- Low risk of carbon formation in the reformer even with higher hydrocarbons in the feed due to their conversion in the pre-reformer.
- All sulphur is removed from the feed in the pre-reformer.
- The steam to carbon ratio can be lowered in the reformer.
- The capacity of the plant can be increased, because there is an extra heat input after the pre-reformer.

Disadvantages of a pre-reformer:

- The pre-reformer will result in a larger pressure drop.
- The pre-reformer will increase the investment cost.

3.1.5.3. Adding CO₂ to the feed

A possible way to increase the CO production is to add CO₂ to the feed. While the reaction reaction enthalpy of the reaction of CO₂ into CO is 1/5th the reaction enthalpy of the reaction of CH₄ to CO (see Table 3.1), more CO can be produced with the same energy consumption in the reformer tubes. An other advantage is that CO₂ reacts with deposited coke to form CO (reaction 3.9), so the steam to carbon ratio can probably be lowered. The steam to carbon ratio can also be lowered because during the reaction of CO₂ to CO also water is formed (reaction 3.4). A disadvantage of this method is the price of CO₂ which is about twice that (per mole) of CH₄. Following the patent of Green [3.10] the H₂ to CO₂ ratio must be close to one.

3.2. Calculations and results

3.2.1. Design

Simulation with Aspen

The reformer was first simulated with the design concentrations and specifications (appendix 5). These data correspond to a production of 1220 kg/h CO. The results of this simulation can be found in appendix 5 and in Table 3.3.

Table 3.3. The heat duties of the heat exchangers, calculated with the design feed.

equipment	heat duty (10^5 W) design
H-2101	41.7807
H-2102	4.3005
H-2103	8.0790
H-2104	2.7157
H-2105A	4.4251
H-2105B	5.4867
H-2106	6.5278
H-2107	18.3583
V-2101	-0.4166
V-2103	4.3241

Heat transfer in the reformer tubes

The temperature gradient inside and outside (tube wall) the tubes can be considered to be linear. The composition half-way is considered to be the average composition in the tubes from inlet to outlet. The conductivity, specific heat and the viscosity of the process gas are based on this composition, the average temperature and the average pressure. The composition half-way the tubes was determined with the mercury (appendix 3.2) program, with the assumption that K_3 is -10 °C, see also Table 3.4 for the composition of the gas half-way the tubes.

Table 3.4. Composition of the process gas half-way the tubes.

Comp.	Φ_{mol} kmol/hr	Percentage (mol%)	Φ_{m} kg/hr	Percentage (weight%)
H ₂	78.0	0.263	155.2	0.029
CO ₂	41.9	0.142	1840.8	0.344
CO	15.8	0.053	444.2	0.083
H ₂ O	119.9	0.405	2156.6	0.403
N ₂	8.8	0.030	246.1	0.046
CH ₄	31.7	0.107	508.4	0.095
	296.1	1.000	5351.3	1.000

In order to determine the energy flow, three parameters are needed, as can be seen from equation (3.20).

A (Outside tube area)

Information given by KTI [3.7].

$$A = 75 \text{ m}^2$$

(ΔT)_{ln} (Logarithmic mean temperature difference)

Can be determined with equation (3.25) and Figure (3.6) [3.6].

$$(\Delta T)_{\text{ln}} = \frac{(720 - 487) - (925 - 860)}{\ln \frac{720 - 487}{925 - 860}} = 132 \text{ K} \tag{3.25}$$

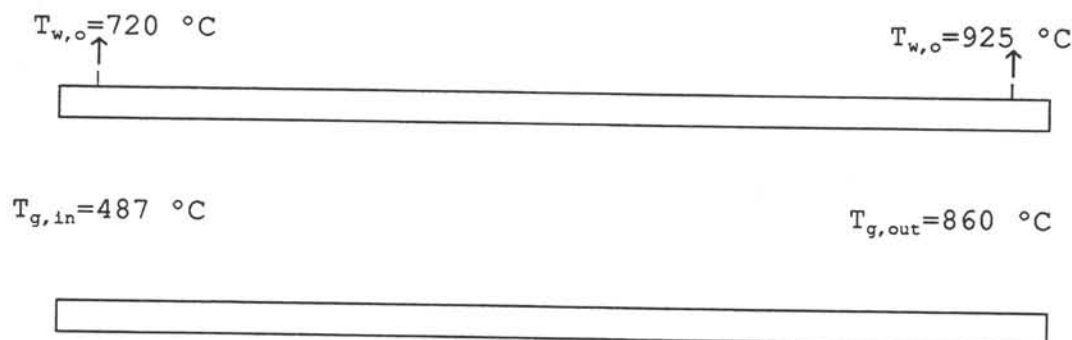


Figure 3.6. Temperature profile of the reformer tubes.

U (Overall heat transfer coefficient)

Can be determined with equation (3.21). See for the results of the calculation of the coefficient appendix 3.3.

$$U = 448 \text{ W/m}^2/\text{K}$$

This results in the following energy flow.

$$\phi_w = U * A * (\Delta T)_m = 448 * 75 * 132 = 4.44 \text{ MW}$$

The overall heat transfer coefficient can also be calculated, or rather be verified, by using equation (3.20) with the data given by KTI [3.7].

$$\phi_w = U * A * (\Delta T)_m$$

$$4.17 \text{ MW} = U * 75 * 132$$

$$\implies U = 421 \text{ W/m}^2/\text{K}$$

This means a difference of only 6 percent. Besides the accuracy of the equations used, the difference between the two overall heat transfer coefficients is most likely caused by depositions, mainly of carbon on the tube wall and the catalyst particles. The depositions have a lower thermal conductivity resulting in a lower energy flow. The calculated heat transfer coefficient can be corrected for the depositions with a so called fouling factor.

Coke formation

With the iteration program given in appendix 3.2 the design (KTI) equilibrium constants are calculated and with figure 3.2 the thermodynamic equilibrium constants are determined, see Table 3.5. The calculated equilibrium constants are larger than the thermodynamic equilibrium constants, because with the equations used, no allowance is made for the higher alkanes and the higher temperatures near the tube wall.

Table 3.5. Thermodynamic and design (KTI) equilibrium constants for the coke deposition reactions.

T (°C)	P (bar)	Therm. K _{3,8} (atm ²)	Therm. K _{3,9} (-)	Design (KTI) K _{3,8} (atm ²)	Design (KTI) K _{3,9} (-)
487	15.00	0.30	365	0.498	618
500	14.88	0.41	221	0.637	390
550	14.41	0.86	40.4	1.53	75.3
600	13.94	1.92	9.97	3.37	17.8
650	13.47	4.06	4.06	7.17	5.35
700	13.00	6.36	1.00	15.1	1.95
750	12.53	13.5	0.30	32.9	0.874
800	12.06	24.5	0.135	77.3	0.484
850	11.59	34.8	0.052	198	0.336
860	11.50	40.4	0.045	240	0.318

Appendix 3.1. Reformer heat duty calculations.

Appendix 3.2. Prediction of the streams and calculation of the equilibrium constants.

Appendix 3.3. Calculation of the design (KTI) overall heat transfer coefficient.

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4.2 CO₂ removal section

The CO₂ removal section involves the CO₂ removal including the water removal, the CO₂ compression and the desulphurisation (see appendix 2, flowsheet 200-FP-41/45). We will look at each of these different functions, including their operation units, separately.

The first step is to determine which units are indispensable with regard to the model. We therefore have to look at the main process streams. It is not necessary to sum up every unit in ASPEN+, because the operation units given by ASPEN can perform several functions in one block. One of these functions is the pressure of the process streams. It is usually provided by pumps or compressors. These types of equipment have been left out as much as possible in order to simplify the flowsheet.

The CO₂ removal section consists of several main operation units. Besides the two packed columns there are some auxiliary units like a reboiler, heat exchangers and knock out drums (K.O. drums). The water cooled heat exchangers (H2108, H2111, H2112) are modelled as HEATER blocks. For these blocks one can specify output temperature and pressure. The other heat exchangers (H2109A, H2109B) are modelled as one block. The block we used is the HEATX block. This block has two in- and outlet streams for the cold and hot streams. For the reboiler (type kettle reboiler) we had to use a separate block. This because ASPEN simulates a reboiler as a part of the column, where the heat stream is given. In order to get this heat stream we placed a HEATER block where the process stream entered the reboiler. Then we connected this heat stream to the reboiler.

The K.O. drums are ordinary vessels where the mixed phases are separated into a gas stream and a liquid stream. All these types of equipment are modelled by a FLASH2 block. This is a block where normally the input is flashed, by pressure drop or temperature difference. This becomes a K.O. drum when these variables are kept constant. A special approach is needed for the MEA vent tank (V2110). It does officially not contribute to the separation process but because it has some losses which are leaving the entire process we had to simulate it as a separate block. For this block the SEP block was used in which the components that were leaving the process stream could be specified.

The two packed columns (V2105, V2106), which are the driving forces behind this section, are modelled as two RADFRAC blocks. These blocks use rigorous fractionation calculations and can also be used for absorbers and strippers. In case of the stripper it is necessary to specify a reboiler duty in order to let ASPEN know a reboiler is connected.

All the streams which are bypassing the main process, like the streams to and from the mixer (M2104) and the aeration/inhibitor injection tank (V2111), are left out because they are not taking part in the main process. This means that the equipment connected to these streams are not modelled either.

In the CO₂ compression section the CO₂ rich gas is compressed by several two stage reciprocating compressors (K2103). These are presented in the model as two COMPR blocks. The only problem is that ASPEN is not equipped with this type of compressor. Therefore the most flexible one is used which is the polytropic centrifugal compressor. The interstage and final cooling, provided by two water cooled heat exchangers (H2113,

H2114), are, like in the CO₂ removal section, modelled as HEATER blocks. For the K.O. drums (V2113, V2114) the same blocks were used as the blocks applied in the CO₂ removal section, serving the same purpose. All other auxiliary equipment, like suction dampers and filters, were left out because of the earlier mentioned reason.

The desulphurisation of the process stream is achieved by a catalyst which is packed in a vessel (V2115). Because the stream has only traces of sulphur and the feed (in ASPEN) has no sulphur components what so ever, the unit is totally neglected even though it causes a small pressure drop.

For a total view of the modelled equipment and flows in this section see appendix 4.

4.3 Drying section

This section has two main functions, namely the compression of the process gas and the removal of the last traces of water and CO₂ (see appendix 2, flowsheet 200-FP-42). This is necessary for usage of the process gas in the cryogenic separation section. These functions, with their operation units, will be discussed also separately.

The compression of the process gas is achieved by a reciprocating compressor (K2104). This is presented in the model as a COMPR block. Because of the earlier mentioned problem the polytropic centrifugal compressor is chosen.

The final cooling, provided by a water cooled heat exchanger (H2115), is again modelled as a HEATER block. For the K.O. drums (V2116, V2117) the FLASH2 blocks were used. All other auxiliary equipment, like suction dampers and filters were again left out because of the reason mentioned above.

The actual drying of the process stream occurs in three columns filled with a zeolite mole sieve. Two of these columns are actually working while the third one is being regenerated by a H₂ rich stream. Even though the process is complicated because of the different residence times, valves and controlling sequences, this part of the process is modelled by just one block. Because of the simplicity of the drying process itself and the preference of the mole sieve a SEP block is chosen in which the several amounts of the components in the different streams are specified.

For a total view of the modelled equipment and flows in this section see appendix 4.

4.4 Cryogenic separation section (coldbox)

The goal of this section is to separate the CO from the other components in the process stream. This is done by cryogenic separation which occurs in a special unit called the coldbox. It is then compressed and ready for further usage in other processes on the site (see appendix 2, flowsheet 200-FP-43/44). Again the section is discussed in these two above mentioned parts.

The coldbox is probably the most complicated unit of the plant. Not only because of the fact that the separation occurs at extremely low temperatures but also because of the complexity of the equipment, the recycle streams and the heat transfer in this unit. It is therefore important how the different streams are modelled. The model is primarily based on the compositions of the different streams. The heat transfer between the streams and units are modelled in such way it only serves a controlling purpose. The coldbox consists of five operation units, namely one heat exchanger and four columns with reboilers and/or condensers. One of the columns is rather complicated. This because of the huge condenser part in the top.

The feed gas cooler (E21) is modelled as a MHEATX block. This block has several in- and outlets and therefore very suitable for this purpose. All other heat transfer units, like reboilers and condensers (E22, E23U/L, E24, E25, E27), are, for the same reason mentioned in paragraph 4.2, presented as HEATER blocks. These different blocks however are not connected to the corresponding reboilers or condensers because of the complexity of the heat balance.

The CO-wash column (T21) consists of two parts. The upper part where the vaporized liquid is condensed and washed and the lower part where the actual mass transfer takes place. Because of this obvious difference in duty the column is modelled in three separate parts. The lower part is modelled as a RADFRAC block and the upper part as two FLASH2 blocks (there are two different temperatures present). The way the mass flows in the upper part however is difficult to predict. Therefore a simple construction is made by returning the liquid flow of the second FLASH2 block to the first FLASH2 block.

The other columns (T22, T23, T24) are also entered as RADFRAC blocks with different reboiler and/or condenser specifications. The CO/CH₄ splitter (T22) has a reboiler and the feed flash column (T23) has a condenser. Therefore the condenser (T22) and the reboiler duties (T23) are specified. The reflux flash column (T24) has both.

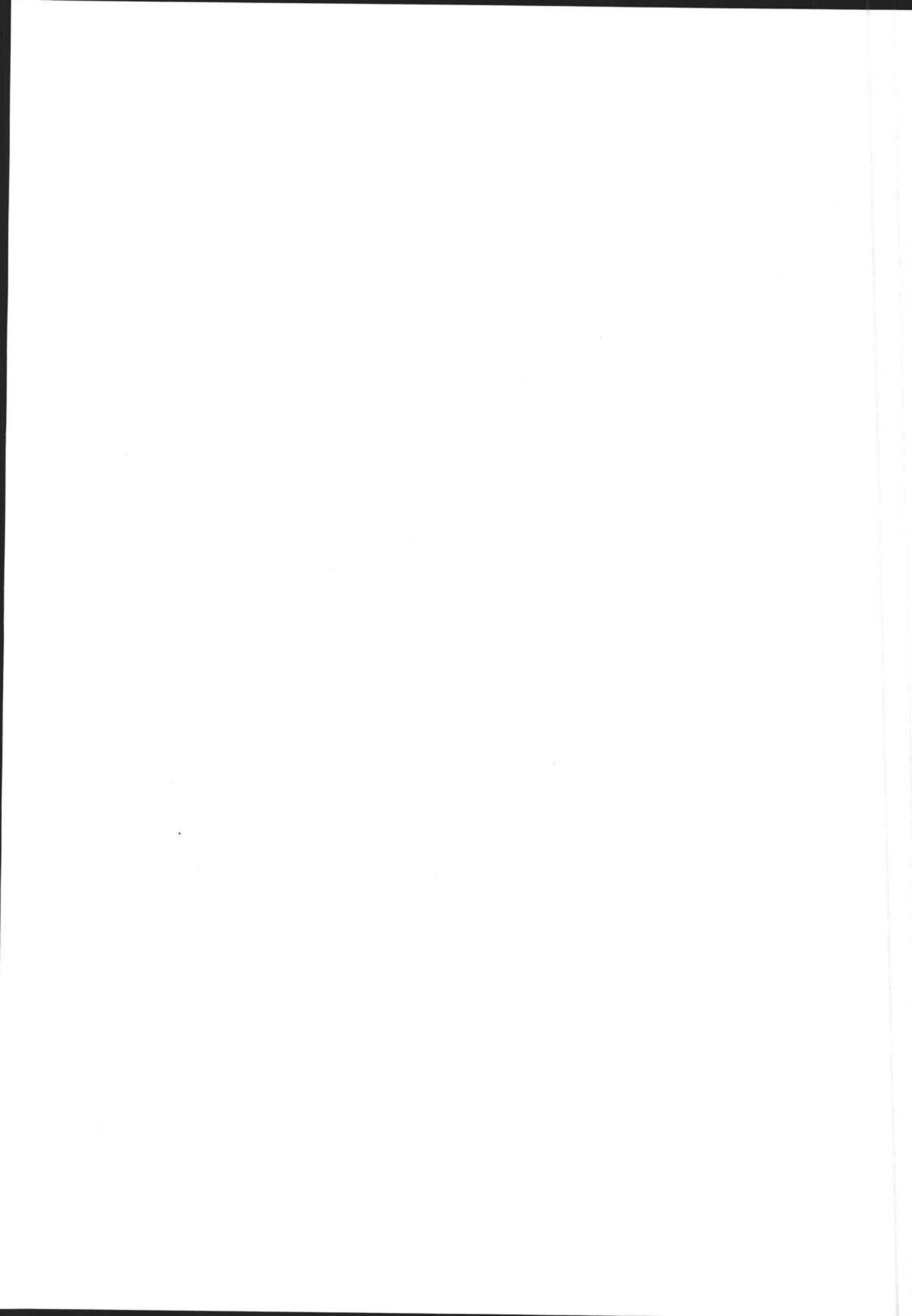
Other equipment, like (controlling) valves, were modelled as HEATER blocks. Also the decision was made to leave out the expansion machines because they do not contribute to the separation process but only to the state in which the H₂ rich gas is leaving the CO-wash column (T21).

The CO rich product gas is compressed by several two stage reciprocating compressors (K2105). These are modelled as two COMPR blocks. The interstage- and final cooling, also provided by two water cooled heat exchangers (H2118, H2119), are, like in the CO₂ compression section, modelled as HEATER blocks. All other auxiliary equipment like suction dampers and filters were again left out.

For a total view of the modelled equipment and flows in this section see appendix 4.

Literature

- 4.1 Design and operating manual carbon monoxide plant II, General Electric Plastic, KTI.
- 4.2 Aspen Plus User Guide, 1988, Aspen Technology, Inc.



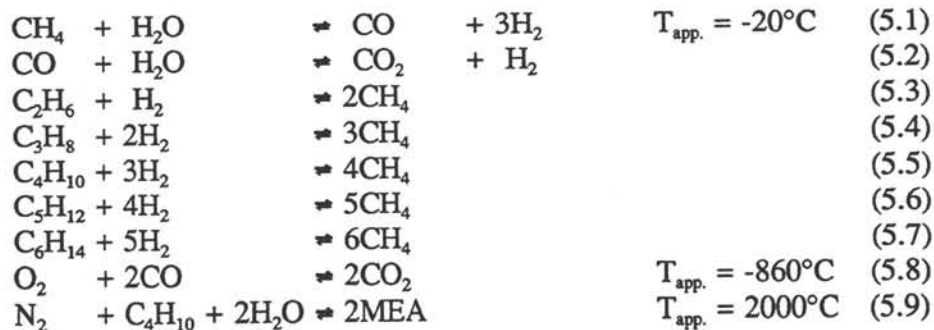
5.SIMULATION AND CALCULATION OF THE STREAMS IN THE PLANT

In this chapter the simulation of the different options as described in chapter 3 will be discussed. First the design values are being simulated in order to get good settings for the different blocks, to simulate the different possibilities to increase the capacity. Furthermore the coke deposition and the heat transfer coefficient are investigated.

5.1 Design

For simulation of the process the design data of the KTI and Linde flowsheets were used (see appendix 2). The pressure, temperature and composition of the streams were pursued as much as possible. The different blocks are specified with a temperature (difference) and a pressure (drop) as much as possible. The blocks with other specifications are discussed separately.

In the reformer section the reformer tubes, simulated by a Gibbs reactor, contain one or more reactions that do not reach equilibrium, a restricted chemical equilibrium calculation must be performed and a set of linearly independent reactions involving all components present in the system must be supplied. In this case reaction 5.1 has its equilibrium at 840 °C as calculated in the Design and Operating Manual [5.1]. All other reactions reach equilibrium at the reactor temperature (design temperature 860 °C). In order to give a temperature approach for reaction 3.3 the following nine reactions were supplied to Aspen:



Although the last two reactions will not occur in the reformer these reactions are given, because the components O₂ and MEA appear further on in the process. To make sure that none of these reactions will take place a large temperature approach is given.

For the heat exchangers in the reformer section, where steam is generated, the vapor fraction is given in stead of the temperature. The flash2 block is simulated with a temperature and a vapor fraction in the outlet stream. For the splitters the mass fractions and a pressure drop are specified.

All attempts to split the steam stream failed, therefore it was necessary to disconnect these streams. The H₂ rich stream to the burners is also not connected to simplify the simulation process. This could be done, because the stream does not contribute to the actual process.

In the CO₂ removal section the separation block is specified with the component fractions, calculated from the design flowsheets, and a pressure drop of zero. For the absorber (V2105) the number of stages and the reboiler and condenser duty are specified (both zero). The number of stages required are defined by pursuing the design top stream composition. The stripper (V2106) is simulated with the number of stages, the condenser duty (zero) and an estimation of the reboiler duty. The final value of the reboiler duty is found by simulating the column with a spec on the mass flow of CO₂ leaving the top.

Between the two columns (MEA) is, in the original flowsheet, a loop in which the MEA circulates. Aspen, however, does not converge if this loop is closed. This because a little bit of MEA and water is transferred to other streams. This can be solved by entering a MEA buffer or by disconnecting the loop. The last option is chosen.

An other difficulty occurs when the heater duty (H2110), which provides the duty for the reboiler, is connected to the column (V2106). The duties differ so much from each other that the desired separation is not possible. Therefore the heat stream is disconnected and the duty of the heater serves only a controlling purpose.

In the drying section only the separator has diviant specifications. The separation block is specified with component fractions, calculated from the design flowsheets, and a pressure drop of zero.

For the cryogenic separation section a heat duty of zero and a pressure drop is given for all modelled valves. The recycle H₂ rich stream to the reformer is given by a mass flow of H₂ in the splitter B1. For the columns first the condenser and reboiler duties, which are zero are given. Then the mass flow of the bottom is given to calculate the boilup or reflux ratios. With these ratios it is possible to replace several duties. This is done to keep the section as flexible as possible. Except for the T23, where a condenser duty is specified because of the fact that at least one (condenser or reboiler) duty has to be specified.

Difficulties in this section occured mainly with the heat balance and the connection of the heaters with the reboilers and condensers of the columns. Therefore these heat streams are not connected to each other. The main goal was to pursue the composition of the mass flows as they are given on the Linde design flow sheets. The duties of the heaters therefore serve only a controlling purpose. Another problem was that the Linde data, first used for the coldbox, did not correspond with the data on the KTI design flowsheets. Hence it was necessary to increase the flows until they matched the KTI design flows.

The results of the design simulation are given in appendix 5. The simulation data correspond with the KTI data. There is only a difference between the CO product stream no. 226 (± 1 kmol N₂ too much) and in the H₂ rich stream no. 228 (too little N₂) coming out of the coldbox.

The heat flow going into the reformer tubes given by Aspen (4.19 MW), does also agree with the KTI value. Because the streams are almost the same as the given design streams by KTI, the already calculated heat transfer coefficient (448.3 W/(m²K)) is maintained (paragraph 3.1.4). The calculated heat transfer coefficients will be compared with each other and be used to make a prediction of the heat flow by other feed streams. Assuming a constant temperature of the outside tube wall (constant ΔT_m).

The CO design production rate, calculated by Aspen is 1225 kg/h.

5.2 Actual

5.2.1 Simulation

The actual amount of CO produced is 1.26 ton/hr, this is about 3.2 % higher than the design production (1.22 ton/hr). With the iteration program described in appendix 3 the necessary increase in the amount of feed is calculated (using the multiplication factor). This calculation leads to a feed increase of ± 2.9 %. Therefore all design feed streams are increased with 2.9 %. Also some block specifications need to be increased, because of the increased feed. Namely one condensor duty, one mass flow over a top of a column and the amount of H₂ recycled (change in splitter). The other block specifications are given in such a way that no change in the specifications is needed when all the feed streams are increased with the same fraction. The most important in and out going streams are given in appendix 5.

The streams that are increased with 2.9 % are : 201, 205, 253, 260, 281 and BIN (stream 251 is not increased with 2.9 %, resulting in a too high temperature of stream 7).

The block specifications that are increased with 2.9 % are : the condensor duty of column T23, the mass flow of CO₂ over the top of column V2106 and the mass flow of H₂ recycled (change in splitter B1).

5.2.2 Calculations using simulated streams

For calculation of the heat transfer coefficient, the stream which enters the reformer is used. The composition of this stream is given in table 5.1 With this stream the composition of the stream half way the reformer is calculated using the iteration program of appendix 3 and by assuming a linear temperature and pressure profile in the reformer. With this composition the heat transfer coefficient is calculated in the same way as carried out in paragraph 3.1.4. This results in a heat transfer coefficient of 458.6 W/(m²K). The heat flow, calculated by Aspen, in the reformer tubes (H2101) is 4.32 MW. The actual heat transfer coefficient will be compared to the heat transfer coefficient of the reformer tubes using other feed streams in order to make an estimation of the new heat flow.

Table 5.1 Stream composition in the reformer tubes

STREAM Comp.	207 stream entering the tubes (kmol/hr)	stream half way the tubes (kmol/hr)	208 stream going out the tubes (kmol/hr)
H ₂	9.09	80.55	153.49
CO	7.11	16.19	55.11
CO ₂	31.19	42.91	31.90
N ₂	8.92	8.92	8.92
O ₂	0.00	0.00	0.00
CH ₄	49.13	32.63	4.72
C ₂ H ₆	1.56	0.00	0.00
C ₃ H ₈	0.20	0.00	0.00
C ₄ H ₁₀	0.08	0.00	0.00
C ₅ H ₁₂	0.02	0.00	0.00
C ₆ H ₁₄	0.03	0.00	0.00
MEA	0.00	0.00	0.00
H ₂ O	155.86	123.35	106.45
AR	0.00	0.00	0.00
T (°C)	487.0 0	673.50	860.00
P (bar)	15.00	13.25	11.50

The equilibrium coefficients of the coke deposition reactions 3.8 and 3.9 are calculated using stream 207, the iteration program of appendix 3 (calculating the composition) and the equations mentioned in chapter 3.1.2, assuming a linear temperature and pressure profile in the reformer tubes. In table 5.2 the thermodynamical equilibrium constants are given for the coke deposition and the calculated ones of the equilibrated gas. To prevent coke deposition the actual values need to be higher than the thermodynamic values. Since the theory used (see 3.1.2) is a rule of thumb, the actual values should be taken as safety reference when the reformer feed is changed.

As can be seen in table 5.2 the current process operates in a safe region since the actual k-values are at least 1.3 times bigger than the thermodynamic k-values.

Table 5.2 Thermodynamic and actual equilibrium constants for the coke deposition reactions.

T (°C)	P (bar)	Therm. K _{3,8} (atm ²)	Therm. K _{3,9} (-)	Actual K _{3,8} (atm ²)	Actual K _{3,9} (-)
487	15.00	0.30	365.00	0.50	620.00
500	14.88	0.41	221.00	0.64	391.00
550	14.41	0.86	40.40	1.53	75.50
600	13.94	1.92	9.97	3.38	17.90
650	13.47	4.06	4.06	7.18	5.35
700	13.00	6.36	1.00	15.10	1.95
750	12.53	13.50	0.30	32.90	0.87
800	12.06	24.50	0.13	77.20	0.48
850	11.59	34.80	0.05	197.00	0.33
860	11.50	40.40	0.04	240.00	0.31

Aspen calculates an actual CO production rate of 1260 kg/hr, which is the same as the given actual production rate.

5.3 Less N₂ in the feed

5.3.1 Simulation

For the simulation a natural gas stream with less N₂ is used. The N₂ is removed using a simulation of a Linde patent (paragraph 8.2.2.2). Using this patent ± 80 % of the N₂ is removed. Because the mole fraction of N₂ isn't large in most of the streams, the other streams and block specifications are kept the same as with the actual simulation. The most important in and out going streams are given in appendix 5.

5.3.2 Calculations using less N₂ in the feed

For the feed with less N₂ also the heat transfer coefficient is calculated using the stream half way given in table 5.3 (see 5.3.1)

This results in a heat transfer coefficient of 450.8 W/(m²K). The heat flow, calculated by Aspen, in the reformer tubes (H2101) is 4.29 MW. Compared with the actual value, the heat transfer coefficient has decreased ± 4 %. This will also decrease the heat flow with ± 4 %, resulting in a heat flow of : 0.96*4.32=4.15 MW. So the calculated heat flow by Aspen will not be reached, resulting in a lower temperature of the stream coming out of the reformer tubes. Which will cause a lower production of CO.

Table 5.3 Stream composition in the reformer tubes

STREAM Comp.	207 stream entering the tubes (kmol/hr)	stream half way the tubes (kmol/hr)	208 stream going out the tubes (kmol/hr)
H ₂	8.61	79.90	152.76
CO	6.05	15.78	54.29
CO ₂	31.19	42.32	31.64
N ₂	1.78	1.78	1.78
O ₂	0.00	0.00	0.00
CH ₄	49.21	32.64	4.80
C ₂ H ₆	1.56	0.00	0.00
C ₃ H ₈	0.20	0.00	0.00
C ₄ H ₁₀	0.08	0.00	0.00
C ₅ H ₁₂	0.02	0.00	0.00
C ₆ H ₁₄	0.03	0.00	0.00
MEA	0.00	0.00	0.00
H ₂ O	155.85	123.87	106.69
AR	0.00	0.00	0.00
T (°C)	487.00	673.50	860.00
P (bar)	15.00	13.25	11.50

The equilibrium coefficients of the coke deposition reactions 3.8 and 3.9 are calculated as in chapter 5.2.2 and given in table 5.4.

As can be seen in table 5.4 the process, with less N₂ in the feed, operates in a safe region with k-values, which are almost the same as the actual k-values.

Aspen calculates a CO production rate of 1253 kg/hr. This is lower than the actual production rate (1260 kg/hr) and will decrease further due to the too high heat flow. So the removal of N₂ out of the feed gas is no good solution for increasing the production capacity. An advantage of the N₂ removal is the big increase in purity of the CO product stream.

Table 5.4 Thermodynamic and equilibrium constants for the coke deposition reactions with less N₂.

T (°C)	P (bar)	Therm. K _{3,8} (atm ²)	Therm. K _{3,9} (-)	Less N ₂ K _{3,8} (atm ²)	Less N ₂ K _{3,9} (-)
487	15.00	0.30	365.00	0.50	629.00
500	14.88	0.41	221.00	0.65	397.00
550	14.41	0.86	40.40	1.56	76.60
600	13.94	1.92	9.97	3.43	18.10
650	13.47	4.06	4.06	7.27	5.41
700	13.00	6.36	1.00	15.20	1.96
750	12.53	13.50	0.30	33.10	0.87
800	12.06	24.50	0.13	77.20	0.48
850	11.59	34.80	0.05	197.00	0.33
860	11.50	40.40	0.04	239.00	0.32

5.4 Capacity increase of 20 % by adding CO₂ to the feed.

5.4.1 Simulation

Using the iteration and the heat duty calculation program of appendix 3 an estimation is made of the feed streams necessary to increase the CO production by 20 %. This is done by assuming a constant heat flow (actual value). Using these streams an estimation is made of the needed changes in the actual streams and block specifications. Extra pressure drops due to the increased mass flow are neglected.

This results in changing the following streams : 201, 205, 251, 253, 260 and 281 (appendix 5).

The changes in block specifications are:

- mass flow of H₂ recycled, splitter B1 (more H₂ needed in the feed)
- condenser duty of block T23 (higher CO flow)
- mass flow of CO₂ over the top of column V2106 (more CO₂ in the stream coming out of the reformer)
- the mass fractions of water (coming out of the steam drum) specified in block SPLIT (stream 208 which must be cooled down by making steam has become bigger)

The bigger H₂ recycle stream is let to the H₂-mix block, which in practise gives a stream to the desulphurizer with a too high fraction of H₂ (more than 10 mole %). An other assumption is the input of CO₂ into the natural gas stream. In practise the CO₂ can better be mixed with the CO₂ recycle stream.

The new block specifications are given in table 5.5 and all the streams are given in appendix 5.

Table 5.5 Changed block specifications.

Block	block type	changed specification	stream	new value
B1	splitter	mass flow	202	97.0 kg/hr
SPLIT	splitter	mass fraction	A C E G	0.13586 0.16845 0.63000
T23	radfrac	condensor duty		-5923 W
V2106	radfrac	mass flow of CO ₂ over the top	213	1662 kg/hr

5.4.2 Calculations using a CO₂ rich feed stream.

Also for the CO₂ rich feed the heat transfer coefficient is calculated (see 5.3.1) using the stream half way the reformer tubes given in table 5.6.

This results in a heat transfer coefficient of 500.2 W/(m²K) due to the higher Reynolds number. The heat flow, calculated by Aspen, in the reformer tubes (H2101) is 4.37 MW. Compared with the actual value, the heat transfer coefficient has increased $\pm 9\%$. This will also increase the heat flow with $\pm 9\%$, resulting in a heat flow of : $1.09 \cdot 4.32 = 4.71$ MW. So the calculated heat flow by Aspen is lower than the calculated one, resulting in a higher temperature of the stream coming out of the reformer tubes. Which will give a higher production of CO. But it must be considered that no temperature approach is used for the CO₂ conversion. With the new CO₂ rich feed streams a temperature approach may be needed, because more CO₂ must be converted in the reformer tubes. When a temperature approach is needed the CO₂ recycle stream will be higher.

The velocity in the tubes has increased compared to the actual velocity, because the number of moles has increased. This can result in a higher temperature approach for both reactions 3.3 and 3.4. This results in a higher recycle stream and a lower CO production rate.

Table 5.6 Stream composition in the reformer tubes

STREAM Comp.	207 stream entering the tubes (kmol/hr)	stream half way the tubes (kmol/hr)	208 stream going out the tubes (kmol/hr)
H ₂	52.40	79.90	170.53
CO	19.80	15.78	67.01
CO ₂	51.65	42.32	37.07
N ₂	8.49	1.78	8.46
O ₂	0.00	0.00	0.00
CH ₄	46.64	32.64	5.96
C ₂ H ₆	1.43	0.00	0.00
C ₃ H ₈	0.18	0.00	0.00
C ₄ H ₁₀	0.07	0.00	0.00
C ₅ H ₁₂	0.02	0.00	0.00
C ₆ H ₁₄	0.02	0.00	0.00
MEA	0.00	0.00	0.00
H ₂ O	131.17	123.87	113.04
AR	0.00	0.00	0.00
T (°C)	487.00	673.50	860.00
P (bar)	15.00	13.25	11.50

As can be seen in table 5.6 the stream coming out of the reformer is bigger compared with the actual stream. This may lead to some capacity increase in the MEA section (see paragraph 6.5).

The equilibrium coefficients of the coke deposition reactions 3.8 and 3.9 are calculated as in chapter 5.2.2 and given in table 5.7.

As can be seen in table 5.4 the current process operates in a somewhat less safer region with k-values, which are some smaller than the actual k-values. But with more CO₂ present in the reformer tubes, CO₂ is able to remove coke, so that the somewhat lower k-values can be accepted.

Aspen calculates a CO production rate of 1551 kg/hr, which is an increase of 23 %.

The mentioned influences on the CO production are hard to predict, but taking everything into account, the addition of CO₂ to the feed will be a good solution to increase the capacity of the plant.

Table 5.7 Thermodynamic and equilibrium constants for the coke deposition reactions for the simulation with CO₂ added to the feed.

T (°C)	P (bar)	Theory K _{3,8} (atm ²)	Theory K _{3,9} (-)	20% more CO K _{3,8} (atm ²)	20% more CO K _{3,9} (-)
487	15.00	0.30	365.00	0.43	545.00
500	14.88	0.41	221.00	0.56	344.00
550	14.41	0.86	40.40	1.36	66.70
600	13.94	1.92	9.97	3.00	15.90
650	13.47	4.06	4.06	6.39	4.76
700	13.00	6.36	1.00	13.50	1.73
750	12.53	13.50	0.30	29.20	0.77
800	12.06	24.50	0.13	67.90	0.43
850	11.59	34.80	0.05	172.00	0.29
860	11.50	40.40	0.04	229.00	0.28

5.5 Capacity increase of 20 % by adding CO₂ to the feed and using a pre-reformer.

5.5.1 Simulation

A pre-reformer converts all higher alkanes, with the advantage that in the reformer the k-values can be lowered and that an extra heat flow can be added to the stream after the pre-reformer. A steam to carbon ratio of 0.4 can be taken in a pre-reformer feeded with natural gas [3.9], the ratio used in the simulation is 0.5. In the reformer the k-values are lowered by adding less steam to the feed. This results in a lower CO₂ recycle, a higher CH₄ recycle and a lower mass flow coming out of the reformer. With a pre-reformer the feed can be taken higher, resulting in a ± 5 % higher CO production.

With the capacity increase using the CO₂ rich feed stream, the MEA section has to be changed. This is because of the higher mass flow coming out of the reformer. Using a feed stream with added CO₂ and a pre-reformer it can be possible to have an increased CO production without the need to change the MEA section.

After the desulphurizer block a mixer is placed to add steam to the pre-reformer feed stream. After the mixer a heater is placed to get the wanted temperature (500 °C) at the inlet of the pre-reformer. For the simulation the pre-reformer is put after the heater and is simulated with a Gibbs reactor. The heater can be left out when the right temperatures are chosen in the heat exchangers, heated by the reformer flue gasses. The extra CO₂ is added to stream 205. In practise the extra CO₂ can better be added to the CO₂ recycle stream (temperature of stream 205 is too high).

The following streams are changed compared with the actual simulation : 201, 205, 251, 253, 260 and 281 (appendix 5).

Also some new streams are entered to connect the new blocks namely stream : 142, 143, 144, 146 and 147 (appendix 5).

The block specification which are changed are :

- mass flow and streams of the recycled H₂, splitter B1 (more H₂ needed in the feed)
- condenser duty of block T23 (higher CO flow)
- mass flow of CO₂ over the top of column V2106 (less CO₂ in the stream coming out of the reformer)
- the mass fractions of water (coming out of the steam drum) specified in block SPLIT (other streams to warm up)
- boilup ratio of block T22 (to get the CH₄ out of the feed, the CH₄ recycle is bigger)
- the streams coming in block H₂O-mix (extra H₂ recycle stream is entered)

The added blocks are : B2, B3 and B5 (see appendix 4).

The block specifications are given in table 5.8 and 5.9 and all the streams are given in appendix 5. In the product stream there is still too much CH₄ present (0.01 kmol), which mean that the boilup ratio of column T22 is still too high.

Table 5.8 Changed block specifications.

Block	block type	changed specification	stream	new value
B1	splitter	mass flow out	150 202 140	80.0 kg/hr 10.7 kg/hr
SPLIT	splitter	mass fraction	A C E G	0.08 0.10 0.704
T23	radfrac	condensor duty		- 6000 W
V2106	radfrac	mass flow of CO ₂ over the top	213	1200 kg/hr
T22	radfrac	boilup ratio		6.0
H ₂ O-mix	mixer	extra stream coming in	140	

Table 5.9 Added block with their specifications

Block	block type	stream in	stream out	specification
B2	mixer	142 144	143	$P_{out} = 18.5$
B3	rgibbs	147	146	heat duty = 0 $P_{out} = 15$ bar
B5	heater	143	147	$P_{out} = 18.2$ $T_{out} = 500$ °C

5.5.2 Calculations

Also for the CO₂ rich feed combined with a pre-reformer the heat transfer coefficient is calculated (see 5.3.1) using the stream half way the reformer tubes given in table 5.10. This results in a heat transfer coefficient of 462.8 W/(m²K). The heat flow, calculated by Aspen, in the reformer tubes (H2101) is 4.21 MW. Compared with the actual value, the heat transfer coefficient is almost the same. The calculated heat flow by Aspen is somewhat lower than actual heat flow. A temperature approach for the CO₂ reaction maybe needed (see paragraph 5.4.2). But taking everything into account the conversion calculated by Aspen should give a good estimation for the CO production.

As can be seen in table 5.10 the stream coming out of the reformer is looking to the total mass and mole flow, about the same compared to the actual stream. So no capacity increase is needed in the MEA section (see paragraph 6.5).

Table 5.10 Stream composition in the reformer tubes

STREAM Comp.	207 stream entering the tubes (kmol/hr)	stream half way the tubes (kmol/hr)	208 stream going out the tubes (kmol/hr)
H ₂	52.67	82.60	170.53
CO	6.92	17.38	63.40
CO ₂	41.29	40.92	28.39
N ₂	8.59	8.59	8.59
O ₂	0.00	0.00	0.00
CH ₄	51.79	41.69	8.21
C ₂ H ₆	0.00	0.00	0.00
C ₃ H ₈	0.00	0.00	0.00
C ₄ H ₁₀	0.00	0.00	0.00
C ₅ H ₁₂	0.00	0.00	0.00
C ₆ H ₁₄	0.00	0.00	0.00
MEA	0.00	0.00	0.00
H ₂ O	122.15	112.42	91.46
AR	0.00	0.00	0.00
T (°C)	487.00	673.50	860.00
P (bar)	15.00	13.25	11.50

The equilibrium coefficients of the coke deposition reactions 3.8 and 3.9 are calculated as in chapter 5.2.2 and given in table 5.11.

As can be seen in table 5.11 the process operates in a less safer region comparing the k -values with the actual k -values. But since the higher alkanes are converted in the pre-reformer, it must be possible to accept the lower k -values.

Aspen calculates a CO production rate of 1507 kg/hr, which is an increase of 20 %.

Table 5.11 Thermodynamic and equilibrium constants for the coke deposition reactions for the simulation using a pre-reformer and CO₂ added to the feed

T (°C)	P (bar)	Theory K _{3,8} (atm ²)	Theory K _{3,9} (-)	20% more CO K _{3,8} (atm ²)	20% more CO K _{3,9} (-)
487	15.00	0.30	365.00	0.43	537.59
500	14.88	0.41	221.00	0.55	337.97
550	14.41	0.86	40.40	1.31	64.70
600	13.94	1.92	9.97	2.86	15.11
650	13.47	4.06	4.06	5.96	4.44
700	13.00	6.36	1.00	12.20	1.57
750	12.53	13.50	0.30	25.48	0.68
800	12.06	24.50	0.13	56.13	0.35
850	11.59	34.80	0.05	134.10	0.23
860	11.50	40.40	0.04	161.02	0.21

The adding of CO₂ to the feed in combination with a pre-reformer seems to be a good solution. Comparing to the solution with only extra CO₂ added to the feed, the advantage of using a pre-reformer is that the capacity of the MEA section need not be changed.

5.6 Conclusions

The capacity of the plant can be increased in two ways :

1. adding CO₂ to the feed
2. adding CO₂ to the feed in combination with a pre-reformer.

For the first option the MEA section must be changed. For the second option a pre-reformer must be installed. Which option is the best will depend on the costs (chapter 9).

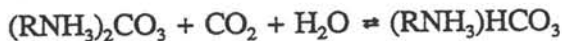
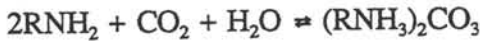
6 MEA section

6.1 Introduction

In the MEA-section the CO_2 gas is removed from the process gas. The process gas is counter-currently transported through an absorption column with a monoethanolamine (MEA) stream at a temperature of 40°C , where the CO_2 is absorbed into the MEA-solution. The CO_2 rich MEA solution is transported to the MEA-stripper. The CO_2 is released from the solution at low pressure and a temperature of 120°C . The CO_2 stream flows to the compressor where it is compressed and recycled to the reformer. The process gas is compressed and flows to the drying unit.

6.2 MEA

Monoethanolamine used for CO_2 -removal works as follows: The hydroxyl group reduces the vapor pressure and increases the water solubility. The amino group provides the necessary alkalinity in water solutions to absorb CO_2 .



Under normal conditions the composition of the equilibrium varies with the partial pressure of the CO_2 over it. But the vapor pressures of these products increase rapidly with temperature, which makes it possible to strip the absorbed gas from the MEA-water solution by the application of heat. As mentioned above monoethanolamine can react directly with CO_2 and has a greater alkalinity which gives it a greater effectiveness than other amines.

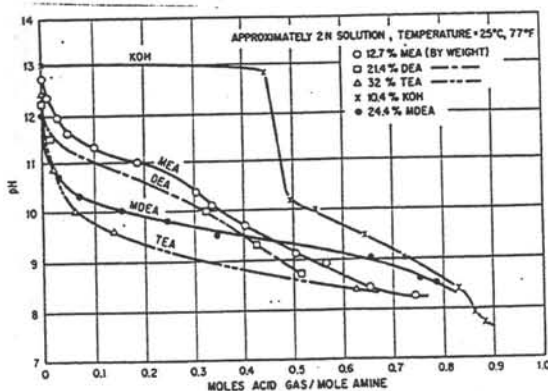
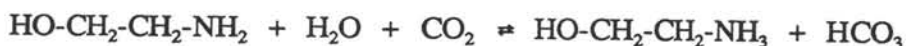


fig. 6.1

6.3 Absorber

The process gas enters the base of the MEA absorber at a temperature of 40°C and a pressure of 10.5 bar. The process gas flows through three packed beds of 2 inch diameter stainless steel Cascade Mini Rings (CMR) with a packing factor of 22. The CO₂ is absorbed in the MEA solution which flows downwards through the packing and absorbs essentially all the CO₂.

The CO₂ reacts with MEA according to the next reaction :



The MEA solution leaves the base of the absorber at a temperature of 72°C due to the heat of absorption. In heat exchanger H-2109 it cools the lean MEA solution from the stripper and heats up to 103 °C.

In ASPEN the actual vapour flows have been calculated together with densities and viscosities of vapour and liquid. This is presented in appendix 6. The maximum possible vapour flow has been determined also using the generalized pressure drop correlation, also presented in appendix 6.

This results in a maximum possible capacity increase, of this absorber, of 109% at 60% flooding with the same packing. Which is more than sufficient for the 20% capacity increase.

6.4 Stripper

The rich MEA solution enters near the top of the MEA-stripper via a flashing liquid distributor and flows down over two packed beds of stainless steel Cascade Mini Rings (CMR) with a packing factor of 29. The solution then flows via a liquid draw off tray into the MEA-reboiler. In this reboiler vapour is generated which flows to the stripper column through the chimney of the liquid draw-off tray and up the tower stripping the CO₂ from the solution. In the top of the stripper the vapour is washed with reflux condensate. The hot lean solution at 120°C flows through the MEA cross exchangers back to the absorber.

In ASPEN the actual flows, densities and viscosities have been calculated based on the Linde data. With these data the maximum possible vapour flow has been determined using the generalized pressure drop correlation, presented in appendix 6.

This results in a maximum possible capacity increase of 17% at 90% flooding, which is very high, with the actual packing. This is just enough for the 20% capacity increase of the plant when using only extra CO₂ in the reformer feed.

6.5 Heat exchangers

When using only extra CO₂ in the reformer feed, the increased flows in the MEA-section make new heat exchanger dimensioning necessary. The new flows are from ASPEN, other data are based on KTI values. The best choice for reasons of efficiency and costs has turned out to be the use of parallel heat exchangers. In appendix 6 this is illustrated with an estimation of the new temperatures using the old heat exchangers with new flows.

The following heat exchangers and condensors should be installed parallel to the existing ones:

heat exchanger parallel to:	H-2108	H-2109 A/B	H-2110	H-2111	H-2112
flow shell side (kg/hr)	3350.4	3350	3848	23775	10152
flow tube side (kg/hr)	13060	3594	1324	1040	403.2
shell outside diameter (mm)	152	135	495	95	181
number of tubes	24	18	40	44	28
temperatures tube side in (°C)	25	72	180	25	102
temperatures tube side out (°C)	35	103	132	35	40
temperatures shell side in (°C)	88	120	120	133	25
temperatures shell side out (°C)	40	88	120	40	35
pressure drop shell (bar)	0.26	0.26	-	0.25	0.4
pressure drop shell/tube (bar)	0.4	0.14	0.24	0.15	0.08
overall heat transfer coefficient (kJ/m ² °C hr)	2928.8	2573.16	3757.2	2263	483
length (m)	5.5	6	6	3.9	3.0
total surface (m ²)	7.63	12.42	14.2	8	4.9

6.6 Pumps

Introduction

The pumps used in the MEA section are designed for a certain operating point. When more fluid must be pumped, a couple of ways to achieve so are discussed here, with the possibilities :

1. Another pump in parallel with the original one.
2. Another pump in series with the original one.
3. Parallel pipelines.
4. change the RPM's of the pump.

Before discussing these four possibilities, it is wise to give a brief resume of the most important aspects of a centrifugal pump.

Characteristic of the pump.

A characteristic is known as a functional relationship between two or more for the process relevant variables. The two best-known variables are the suction head H and the volume flow ϕ_v (fig. 6.2). The suction head is the pressure increase in the pump, converted into the height of a column of the pumped fluid:

$$H = \frac{1}{\rho g} (P_2 - P_1) = h_{geom} \quad (6.1)$$

The pressure difference itself is called the manometric head. H is expressed in lengthunits, so in SI units it is in meters. When working with centrifugal pumps, the "characteristic" is meant to be H as a function of ϕ_v .

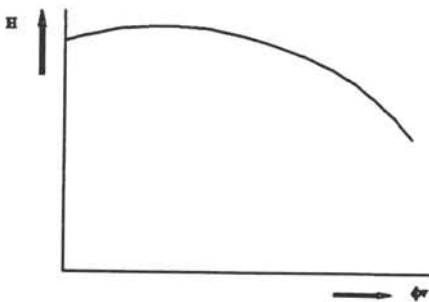


fig 6.2 pumpcharacteristic

The form of the characteristic of a centrifugal pump can be explained with simple physicle models. The simplest model is Euler's, by which the moments of inertia theorem is used for the fan under the assumption that the flow of the fluid is homogeneous and takes along with the rotating movement (fig 6.3). This model results in a linear characteristic, of

which the tangent depends on the fanblade configuration. The requirements aren't coming up in a real fan completely, through which the characteristics descend. On top of that, the fan is situated in a pump, so there will be flowresistance, of which the effect roughly is changing with the square of the volume flow ϕ_v . Flows less and more than the optimal operating point, results in oblique entrance in the fan. This all means that Euler's characteristic will never achieved (fig 6.4).

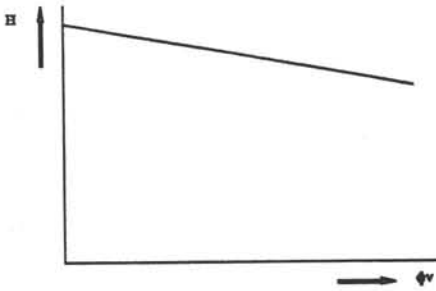


Fig 6.3 Euler's characteristic

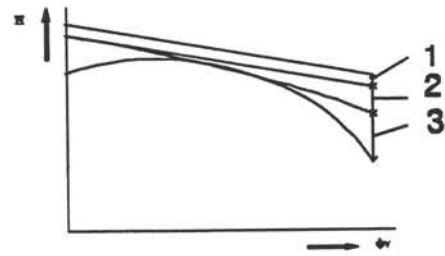


fig 6.4 real characteristic

- 1 Effect of not homogeneous distribution
- 2 Effect of flow resistances
- 3 Effect of oblique entrance
(with respect to fan direction.)

Systems characteristics

Often there is a functional relationship between pressure drop and volume flow for the system a pump cooperates with. When this relationship is drawn in the $H-\phi_v$ diagram, the result is an operating point of the pump.

Example 1

Systems with only height difference, without flow resistance in the pipes (fig 6.5 and 6.6).

Here the geometric height h_{geom} is the same as the height H (formula 6.1)

If the height difference doesn't depend on the flow ϕ_v , then the system curve is a horizontal.

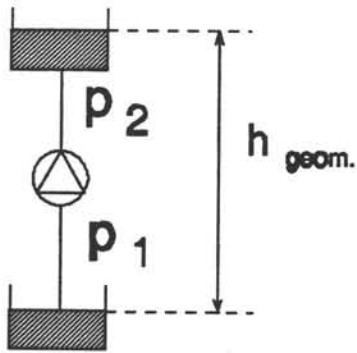


fig 6.5

system with heightdifference

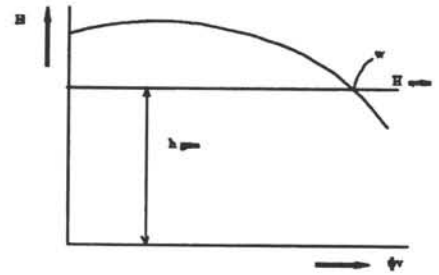


fig 6.6

Example 2

Only flow resistances, no height differences.

Assume the flow resistance can be described by a square resistance law (valid in tubes above $Re \approx 2000$):

$$\Delta p = \frac{L}{D} \xi \frac{1}{2} \rho c^2 = \text{constant} \cdot \varphi_v^2 \quad (6.2)$$

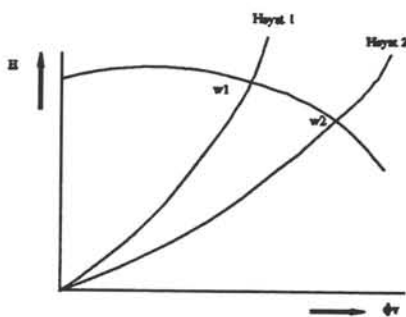


fig 6.7 system with only piperesistance

With this model the pipe characteristic is a parabola with its top in the origin of the axiscross. In fig 6.7 two system curves with different values for the constants are drawn.

Example 3

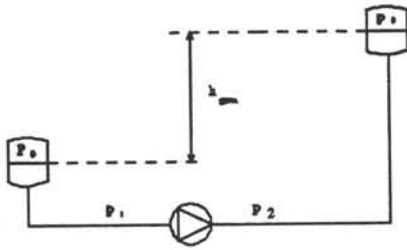


fig 6.8

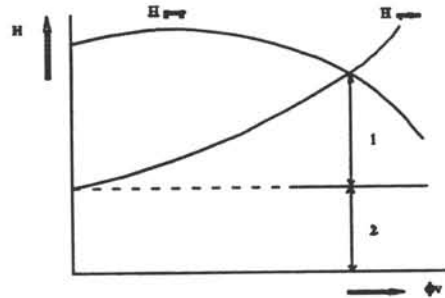


fig 6.9

System curves for the more common case

More common case: height difference, pressure difference and flowresistance (fig 6.8 and 6.9). This is a combination of example 1 and 2, whereby also different gas pressures p_0 and p_3 above the levels are assumed. So now we have:

$$\frac{p_2 - p_1}{\rho g} = H_{pump} = \frac{p_3 - p_0}{\rho g} + h_{geom} + \frac{\Delta p_{pipe}}{\rho g} \quad (6.3)$$

Unstable behaviour of a centrifugal pump.

It is possible that a characteristic has a rising and descending part as shown in fig. 6.10. In that case the pump might operate unstable, depending on the coupled system and the position of the operating point. Assume that with the as example drawn system, the system curve exists of a static height (H_{01}) and a resistance height (H_{pipe1}). The resulting operating point is W_1 . Here, the value of H_0 depends on the use: The pump must deliver as much as the user can take. Assume, this is not the case and the deliverance of the pump becomes to large. The operating point W moves across the pump characteristic to a higher value of H , causing ϕ_v to decrease. An equilibrium can be found that way, for example at W_2 . Decreases the consumption by the user, the pump will search for a new equilibrium, whereby H_0 increases again and ϕ_v decreases. The utmost stable operating point that can be achieved that way, is W^* . when the deliverance of the pump still is too large, H_0 wants to rise further; the pump can deliver less, but at a lower H_0 ; The pump stops at that point and when no retraction valve is used, even backflowing may occur. When because of this backflowing, H_0 decends the pump will be able to work again. The present operating point lies on the rightside of the characteristic. The value of ϕ_v is however much to big again,

and the operating point will again move towards W^* , after which the cycle repeats. At this way a pulsing behaviour shows up, not only in the fluid flow, but also in the traction of the pump. The frequency depends on the volume flow and the reservoir dimensions. When the pump load only exists of static height (no pipe resistance), then the ultimate between the stable and instable part of the characteristic lies not near W^* , but in the top T of the curve. When the load exists only out of pipe resistance (pump working on a smore valve for example), then the pipe curve always has a point of intersection with the pump characteristic.

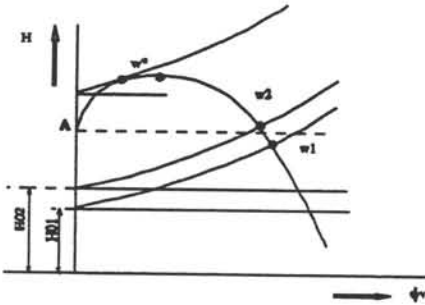


fig 6.10 pump characteristic with chance for instability.

ad 1) Pump operating in parallel

Where it is necessary to increase the amount of liquid to be pumped, two or more pumps are often installed beside one another. This condition has been sketched in fig 6.11.

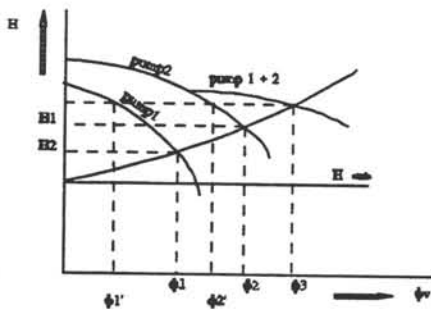


fig 6.11 two pumps in parallel

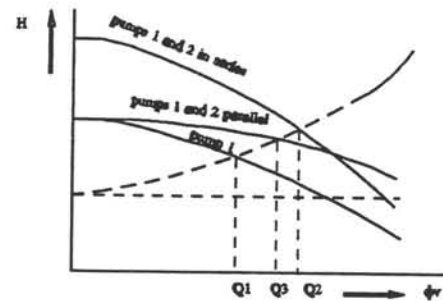


fig 6.12 different configurations

With either pump 1 or 2 in operation, capacities would be Q_1 and Q_2 respectively at manometric heads H_1 and H_2 . If pumps 1 and 2 are operating in parallel, the capacities of pumps 1 and 2 at equal head can be added. The new curve intersects the pipe characteristic in point Q_3-H_3 . Pump 1 capacity is now Q_1' and pump 2 capacity Q_2' . The above mentioned pipe characteristic refers only to the common discharge line of the pumps and hence the friction of the individual suction and discharge lines should be subtracted from

the $Q-H_{tot}$ curve by the method described under 3.1 before determining operating points for pumps 1 and 2.

ad 2) Pumps operating in series

If the manometric head of one pump appears to be too low to displace a certain quantity of liquid through a given pipeline, more pumps can be installed in the same pipe line to increase the pressure. The discharge of the first pump is connected to the suction of a second pump. Depending on local conditions, these pumps can either be placed close to each other (in the same room) or far apart. It is unusual to install pumps in series in new installations since a multi-stage pump is usual preferred. If two pumps, 1 and 2 with identical $Q-H_{tot}$ curve are installed, capacity will amount to Q_1 when either pump 1 or 2 is operated separately. When both pumps are operated in series, the joint $Q-H_{tot}$ curve is found by adding the manometric head at equal capacities. The intersection with the pipe characteristic now shows operating point Q_2 with series-operation (fig 6.12)

ad 3) Parallel pipelines

If the capacity of one pipeline becomes too low - e.g. because of plant enlargement - a second pipe can be placed parallel to the existing one.

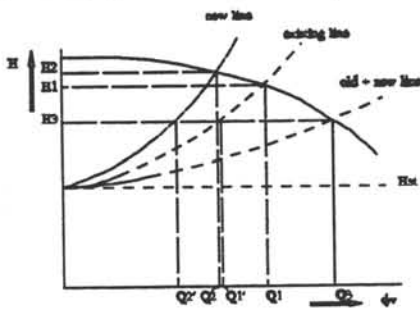


fig 6.13 different piperesistances

An increase in capacity is provided that can be calculated as follows: in graph (6.13) a pump in the existing pipeline is operating at capacity Q_1 and H_{tot} H_1 . If the pump operates on the new line only, capacity is Q_2 at H_{tot} H_2 . Pipe characteristic of the total system can be found by adding capacities of the old and new pipeline (at equal friction).

$Q_1' + Q_2' = Q_3$. The intersection of this characteristic with pump $Q-H_{tot}$ curve gives operating point $Q_3 - H_3$. Capacity per pipeline can be found by reading same for the particular pipe at head H_3 (Q_1' and Q_2').

ad 4) Change the RPM's of the pump

For centrifugal pumps, there is a relation between manometric head H and the rotationfrequency n . At different rotationfrequencies the same centrifugal pump has different pump

characteristics. The one belonging to the higher rotation frequency lies above, the one belonging to the lower rotation frequency lies under the normal $Q-H_{tot}$ curve (fig 6.14).

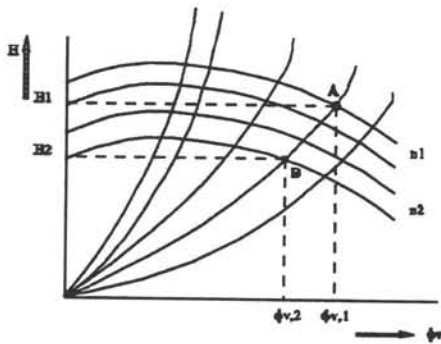


fig 6.14 different rpm's

These characteristics are determined by a congruence law, that is why the shapes are alike. For fluids with low viscosity the following relations exist:

$$\frac{\varphi_{v,2}}{\varphi_{v,1}} = \frac{n_2}{n_1} \quad (6.4)$$

$$\frac{H_{tot,2}}{H_{tot,1}} = \frac{n_2^2}{n_1^2} \quad (6.5)$$

$$\frac{P_2}{P_1} = \frac{n_2^3}{n_1^3} \quad (6.6)$$

Here P is the power needed to pump. The index 1 relates to the original situation. This congruence law can be tested when the new rotation frequency does not deviate from the original rotation frequency.

Power for the pumping

The power the electro engine has to deliver can be calculated as follow:

$$-P_{axis} = \frac{v (P_2 - P_1) \phi_m}{\eta} = \frac{\rho g H \phi_v}{\eta} \quad (6.7)$$

Here η is the total efficiency of the pump, ϕ_m is the massflow.

The massflow will rise 15% so this means that the power increase per pump will also be around 15%. The cheapest and easiest way to increase the flow will be the latter described way, so with a higher rotation frequency. Questions remaining now are: can the electro engine handle the power, and doesn't the flow resistance become too huge in the pipes?

Calculations have been made for the pumps in the MEA section, with two massflows: ϕ_{m1} and $\phi_{m2}(=1.2*\phi_{m1})$. The efficiency was set on $\eta=0.8$. With formula 6.7, the power needed was calculated, by assuming that only the massflow would change; the rest like pressures, viscosities and mixture ratios are constant. We just want to see if the pumps can handle the flowincrease, so we need to have a estimation of the powerconsumption. These were the results:

Table 6.1: calculations results.

	ϕ_1	ϕ_2	P_{in}	P_{out}	v	P_{KTI}	$P_{+20\%}$
	kg/s	kg/s	bar	bar	cP	kW	kW
N2101A/B	6.10	7.32	1.5	13.7	9.8e-4	9.1	10.9
N2102A/B	0.27	0.33	0.5	0.7	1.0e-3	0.02	0.02
N2103	0.63	0.76	0.6	0.7	9.8e-4	0.01	0.01

There will be no problems for the pumps to increase the flow. Only pump N2101 has to pressurize the fluid, the other two pumps have to displace fluid more than pressurize it. Because no extra data then the sheets are known, nothing can be said about the point of instability of the pumps.

6.7 Process gas drying.

The process gas from the MEA absorber contains CO₂ and H₂O, which would freeze in the cold box. The dryer removes these components to a level of 0.1 ppm volume.

The process gas from the MEA section has a temperature of 40 °C and a pressure of 9.4 barg. Any entrained condensate in the gas is separated in a K.O. drum. After this the gas is filtered, compressed, cooled and again sent through an K.O. drum. The gas leaving this K.O. drum (temperature 35 °C, pressure 27.5 barg) is sent to one of the three mole sieve beds, which will further remove the H₂O and the CO₂. One adsorbent bed is in line taking fresh feedgas to adsorb all carbon dioxide and water. The gas from this bed is then routed through a second bed, used as a guard bed, before flowing to the cold box. The third bed is in regeneration. In appendix 6 the cycle sequence is showed.

The process gas dryers are mole sieve beds. Molecular sieves are called zeolites. Commercial molecular sieve zeolites are usually synthetic. They are crystalline framework aluminosilicates containing alkali metal cations. The structure extends in three dimensions by a network of AlO₄ and SiO₂ tetrahedra linked to each other by sharing of oxygen atoms. Molecular sieves possess the high porosity that is characteristic of all adsorbents. In addition, the ordered crystalline structure of the molecular sieve provides pores of a constant size.

The pore size of molecular sieves can be enlarged or diminished by appropriate cation exchange. Therefore many commercial types are available with pore openings ranging from 0.3 nm to about 1.0nm. This property gives molecular sieves unique advantages in certain applications because many gases or liquids can be excluded from the microporous structure. Larger pore molecular sieves can also be used to simultaneously dry and purify streams.

Because of the ordered structure, molecular sieves have excellent capacity at low water concentrations and do not exhibit a capillary condensation pore-filling mechanism at high water concentrations. Molecular sieves are also inert to most fluids and are physically stable when wetted with water. Strong inorganic acids or alkalis should be avoided as well as temperatures above 700 °C.

For the generation of the process gas dryers, the hydrogen rich gas from the cold box is heated to 200 °C by steam and passed across the bed at low pressure to drive off the CO₂ and H₂O. The regeneration contains the following steps:

- Depressurisation step.
The bed is reduced slowly in pressure to the same pressure level as the regeneration gas.

- Heating step.
A flow loop of regeneration gas is established and the heating cycle is started.

- **Cooling step**
The regenerated bed goes on to cooling. The generation gas now bypasses the regeneration heater and the cold gas slowly cools the bed.
- **Purge step**
The regenerated bed goes onto a purge step. This is to slowly purge the bed of its regeneration gas volume so that when it actually goes into line there is no significant change in the quality of gas flowing to the cold box.
- **Repressurisation step.**
The regenerated bed is now repressurized to operating pressure.

If the production capacity of the plant is increased by using only extra CO₂ in the reformer feed, then the CO₂ and H₂O amount sent to the dryers will be somewhat greater. This will result in a shorter adsorption time, so the guard- or the regeneration time must be decreased. At first it is necessary to decrease the regeneration time, which maybe possible by using more H₂ to regenerate the dryer and by speeding up the separate regeneration steps. If it is still necessary to decrease the guard time to cope with the decreased adsorption time the risk of freezing in the cold box will be greater. So the twenty percent capacity increase may lead to a higher risk of freezing in the cold box.

6.8 Conclusion

The capacity increase for 20% more CO production by adding CO₂ to the reformer is possible with the existing MEA equipment, although the 90% flooding in the stripper is high. The heat exchangers in the MEA section need parallel heat exchangers. The dryer section need not to be changed, although the risk of freezing in the coldbox could be greater. Further stretching would require an other stripping column. The columns in the coldbox all have enough overcapacity and also the pumps do not need to be changed.

When using a pre-reformer in combination with adding extra CO₂ there will be no need to change anything because the flows are the same or less and the CO₂ content is about the same.

Literature

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7 CRYOGENIC SEPARATION

7.1 Introduction

The process gas coming from the drying unit containing H_2 , CO, N_2 and CH_4 goes to the coldbox, the cryogenic separation. This separation is based on the difference in boiling points of CO and $H_2 + CH_4$. The process gas is cooled to a temperature of $-203/-193$ °C by heat exchangers. For this process, the required refrigeration can be obtained by H_2 expansion and a CO recycle loop. The cold box consists of condensators, flash drums, CO/ CH_4 splitters and a CO wash column. Because of the close physical properties the N_2 can not be completely separated from the CO.

7.2 Columns

Feed gas coming from the drying unit is precooled against separation products in heat exchanger E-21. In the reboiler E-22 of the column T-22 the feed gas is further cooled down by producing rectification vapour in the column T-22, and then is fed into the lower part of the column T-21.

In this column, a methane rich liquid is produced in the bottom and a methane-free liquid is produced in the top. In the top cooler E-23 L and U, the top gas of T-21 is cooled down to 201 °C by condensing a maximal amount of carbon monoxide. The hydrogen-rich gas is warmed up in E-23 U, enters the first stage of the piston expander EM-21 and is warmed up a third time in E-23 U. After passing E-23 L, it is warmed up in E-21 and leaves the plant. A part of the liquified carbon monoxide is used as reflux for T-21 to wash out the methane from the gas stream.

The remaining part of the carbon monoxide is taken from a chimney tray in the upper part of T-21 and expanded into the H_2 flash column T-24, where the dissolved hydrogen is boiled out.

The methane rich bottom product of T-21 is expanded into the other H_2 flash column T-23, where the dissolved hydrogen is also boiled out. The top products of the two H_2 flash columns, mainly hydrogen are fed to the tail gas. The bottom liquid of T-23 is expanded and used as refrigerant in one pass of the condenser E-23 L.

It is then fed into the middle section of the CO/ CH_4 splitter T-22. The expanded bottom liquid of the other H_2 flash column T-24 is mixed together with the two expanded carbon monoxide streams from the reboilers E-22 of the CO/ CH_4 separation column T-23. Afterwards this stream is used firstly as refrigerant in the condenser E-25 of the column T-24 and enters then the CO/ CH_4 splitter as reflux.

The bottom liquid of the column T-22 consisting mainly of methane represents the major part of the tail gas to which the head product of the H_2 flash columns T-23 and T-24 are added. It is warmed up in E-21 and leaves as tail gas. The top product of T-22 represents the carbon monoxide product.

The carbon monoxide recycle is first cooled down against separation products in the heat exchanger E-21 is split into three streams. Two streams provide heat to the reboiler E-22 of T-22 and reboiler E-24 of T-23. The third stream is injected into the H_2 flash column

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T-24 as stripping gas. The two carbon monoxide streams from the reboilers E-22 and E-24 are expanded, and mixed to the bottom liquid of T-24.

7.3 Conclusion

The columns all have enough capacity for the 20% capacity increase by either using extra CO₂ in the feed or by using a pre-reformer in combination with the extra CO₂ in the feed. Appendix 7 contains the calculations.

8. NITROGEN REMOVAL

In this chapter the advantages of the removal of nitrogen from the natural gas feed are discussed. The different methods to achieve this are described and simulated where possible.

8.1. Advantages of removal of nitrogen from natural gas feed

The feed of the plant, natural gas from Groningen, contains about fourteen volume percent nitrogen. Because the nitrogen is an inert gas it will flow through the plant without being used anywhere. The energy that is needed to pressurize, to heat and to cool the nitrogen during the process are wastes of energy. When we release the flows from nitrogen they can contain more usefull components, which results in more feedstock and probably in a capacity increase. When the nitrogen is totally or partially removed from the feed gas then the product will have a higher purity than the actual one which has a molar ratio carbon monoxide/nitrogen of 9/1.

After separating the nitrogen from the natural gas there will be two flows, a nitrogen-rich flow and a nitrogen-poor flow. The latter will be the feed of the existing CO plant, the nitrogen rich flow can be fed in the natural gas net of the total plant. This is possible because the amount of gas that the CO plant takes from the gas net of the total plant is about ten percent. So when the nitrogen-rich flow, which is very small compared to the total gasnet is mixed with the natural gas of this gasnet the result will be an increase of the nitrogen that is negligible small.

8.2. Available separation technology

This paragraph will study the different methods that can be applied for the nitrogen separation, which in practice means the separation of nitrogen from methane, because methane is the most important component in natural gas. The following methods will be discussed:

- Absorption
- Cryogenic separation
- Membranes
- Pressure Swing Adsorption

8.2.1. Absorption

Generally absorption is a very suitable method for separating one component. The absorbent is a large molecule that binds the specific component that has to be removed. The problem in this case is that such a molecule is not yet available for nitrogen separation from methane. So absorption is not suitable for this proces.

8.2.2. Cryogenic separation

The most applied method for nitrogen removal from a gas which contains methane is cryogenic separation. Cryogenic separation is separation at low temperatures in order to liquefy one or more components or to create low or high solubilities of one component in another. This paragraph describes two methods of cryogenic separation. The first is an installation with one column, the second uses three columns.

8.2.2.1. Method I

The first method is described in two German patents from 1970 of Messer Griesheim GmbH [lit. 8.1, 8.2]. It is based on a nitrogen containing natural gas that is separated in a nitrogen-rich flow and a nitrogen-poor flow, which contains almost all the methane, by one column. It is not known if this installation is operating in practice. The description of this method is the following. Figure 8.1 shows the flowsheet on the next page.

The natural gas enters the plant in pipe 1. In heat exchanger 1 the flow is cooled by the leaving products and then heats the bottom of the column in reboiler 2 in order to create a boil up. Then the flow (3) is flashed in valve 3 to column pressure of about 28 bar and fed into the column. Via pipe 5 the nitrogen-rich flow is withdrawn from the top of the column. This stream can be pure nitrogen, depending on how far the rectification is raised. In this case this is not necessary and also too expensive. The nitrogen-rich flow passes heat exchanger 4, where it is heated, and is then expanded in turbine 5, in which it is cooled deeply. The created coldness is used in heat exchanger 6 in the middle of the column in order to create a reflux, necessary for the distillation. The rest of the coldness is withdrawn from the nitrogen rich fraction in heat exchanger 7 and 1, after which it leaves the plant through pipe 10. The nitrogen poor flow is withdrawn from the bottom of the column (pipe 11) after which it is splitted (8) in two flows. The largest flow (90%) is flashed in valve 9 in order to create coldness and is then heated by the entering gas. Then it leaves the plant through pipe 14 at a pressure of about 7 bar.

A part of the bottom products flows through pipe 15 and is cooled, after which it is flashed to about 2 bar. This flow is used as an open cycle in order to create a reflux in the top of the column. It is flashed in valve 10 to about 2 bar and is then evaporated in heat exchanger 11, which results in the desired reflux. The pressureless nitrogen-poor stream in pipe 19 flows through heat exchangers 4 and 7 in which it is heated in countercurrent flow with the cooling bottom products. In heat exchanger 12 the cycle medium is brought to about 5 °C and is then compressed to column pressure by compressor 13. After flowing through water cooler 14 and heat exchanger 12 the cycle medium is fed into the column as vapour flow 25.

The way this patent is simulated with Aspen Plus is described in Appendix 8, which also contains the results of the simulation.

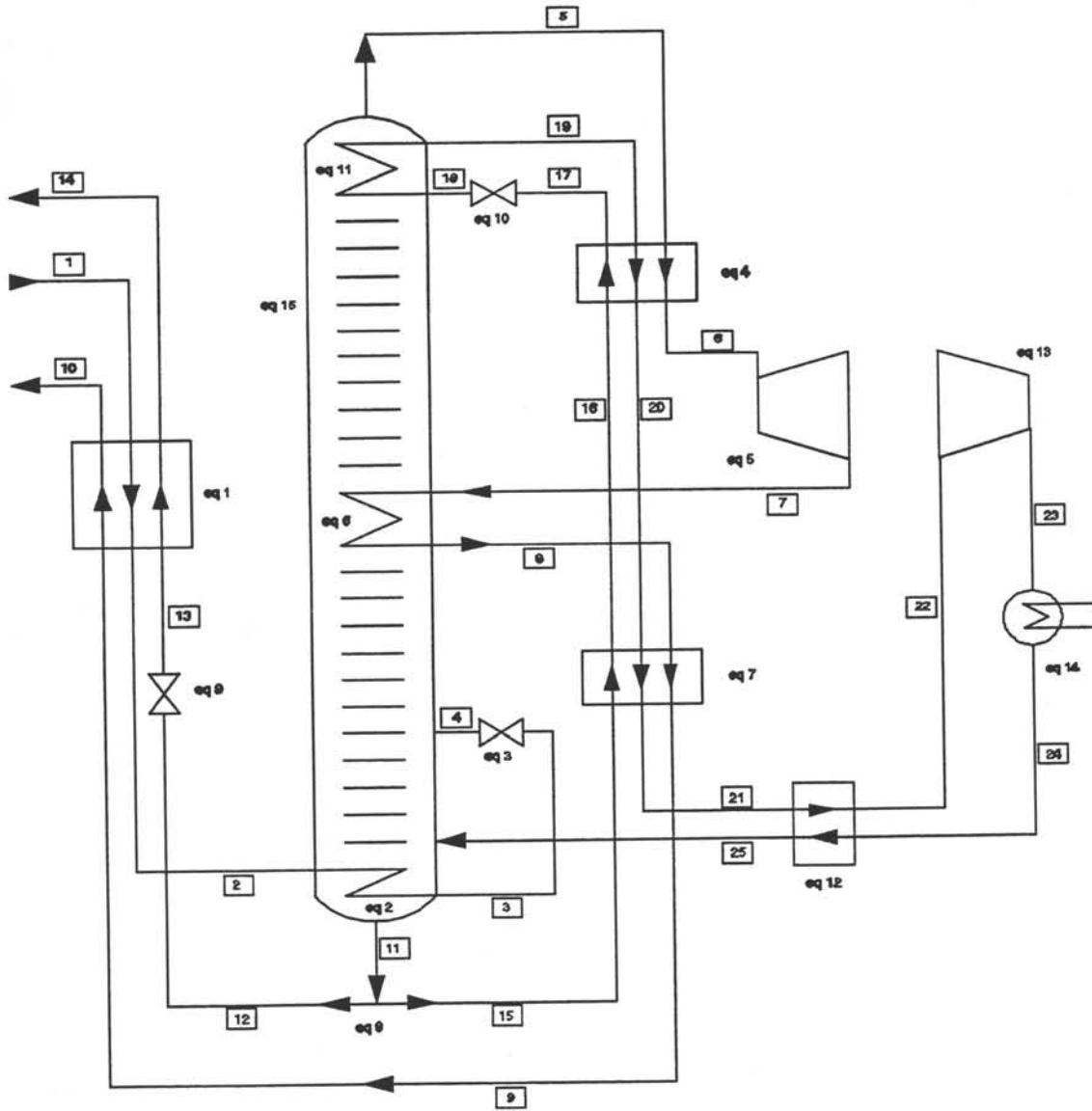


Fig. 8.1. Flowsheet of method I.

8.2.2.2. Method II

The second method is described in a Dutch patent from 1972 of Linde A.G., Germany [lit. 8.3]. The invention is related to a process for separating nitrogen from natural gas by rectification at low temperatures. The carbon dioxide, together with the two or more C-atoms containing hydrocarbons and a part of the methane, is separated by rectification as a high boiling fraction at medium pressure level. The CO₂-free fraction is separated in a nitrogen and a methane fraction by further rectification. After heat exchanging with the hot stream which enters the system, the latter two streams are taken from the system as high boiling fractions. It is not known if this installation is operating in practice. The description of this method is given in the following. Figure 8.2 shows the flowsheet on page 8-6.

In compressor 1 the natural gas is compressed to 48 bars. Assuming that the natural gas contains no water, so the dryers (2) are not needed, the gas mixture is cooled in heat exchanger 3, so that the higher hydrocarbons (C₅+) are almost completely condensed. In separator 4 the liquefied higher hydrocarbons are separated. The vapour fraction is condensed in heat exchangers 5 and 6. After being flashed in valve 7, the mixture enters the medium-pressure column (33 bars) 8.

The bottom product of this column has the following composition: 3 mole-% N₂, 91.5 mole-% CH₄, 3.6 mole-% C₂H₆, 0.5 mole-% C₃H₈, 0.1 mole-% C₄H₁₀ and 1.3 mole-% CO₂. This fraction is a product and carried off at 175 K via pipe 9, vapourized and heated in heat exchangers 5 and 3 and delivered at 32 bars.

The overhead product of column 8, containing 36 mole-% N₂ and 64 mole-% CH₄, is carried off via pipe 10 at a temperature of 164 K. In heat exchanger 11 this vapour fraction is partially condensed at 151 K. The liquid fraction contains 68 mole-% CH₄ and is gathered in separator 12. The liquid partially returns to column 8 via pipe 13 as a CO₂-wash stream. After being flashed, the vapour fraction enters column 15 via pipe 14 at 28 bars. The bottom temperature of this column is 148 K; the percentage nitrogen of the vapour is 30 mole-%.

The remaining part of the vapour from separator 12 is carried off via pipe 16 and together with the bottoms of column 15 cooled in heat exchanger 17 at 112 K and enters via valve 18 the low-pressure column 19 at 1.5 bars. The wash stream needed for this column is gained in column 15. At the top of this column, liquefied nitrogen containing 5 mole-% CH₄ with a temperature of 123 K is carried off via pipe 20 and partially returns via pipe 21 to column 15 as a methane-free wash stream. The remaining part is cooled in heat exchanger 22 to 86 K and after being flashed, enters column 19 as a wash stream.

The liquefied CH₄-fraction at the bottom of column 19 still contains 3 mole-% nitrogen and is almost CO₂ free (50 ppm). This fraction, with a temperature of 109 K, leaves column 19 via pipe 23, compressed with pump 24 to 13 bars and heated in heat exchanger 17 to 146 K. At point 25 almost completely condensed circulation-methane is added. This stream is compressed by compressor 26 to 36 bars, cooled in heat exchangers 3 and 5, condensed in heat exchanger 6, subcooled in heat exchanger 27 to 160 K and flashed in valve 28 to 13 bars, taking a temperature of 151 K. The united CH₄-streams are vapourized in heat exchanger 11 in countercurrent flow with condensed CO₂-free overhead product from column 8, and

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heated to 158 K. After heating in heat exchangers 27, 5 and 3, the stream is taken from the system at ambient temperature. The amount circulation-methane is taken at point 33 and sucked by compressor 26. The CO₂-free methane fraction is mixed with the higher hydrocarbons from separator 4, and taken from the system via pipe 30. The percentage carbon dioxide and higher hydrocarbons in stream 30 is 2 mole-%.

The separated nitrogen leaves the top of column 19 as a vapour fraction at a temperature of 83 K and a percentage of N₂ of 99.5 mole-%. This stream is heated in heat exchangers 22, 17, 11, 27, 5 and 3 to ambient temperature and taken from the system via pipe 32 at atmospheric pressure.

The way this patent is simulated with Aspen Plus is described in Appendix 8, which also contains the results of the simulation.

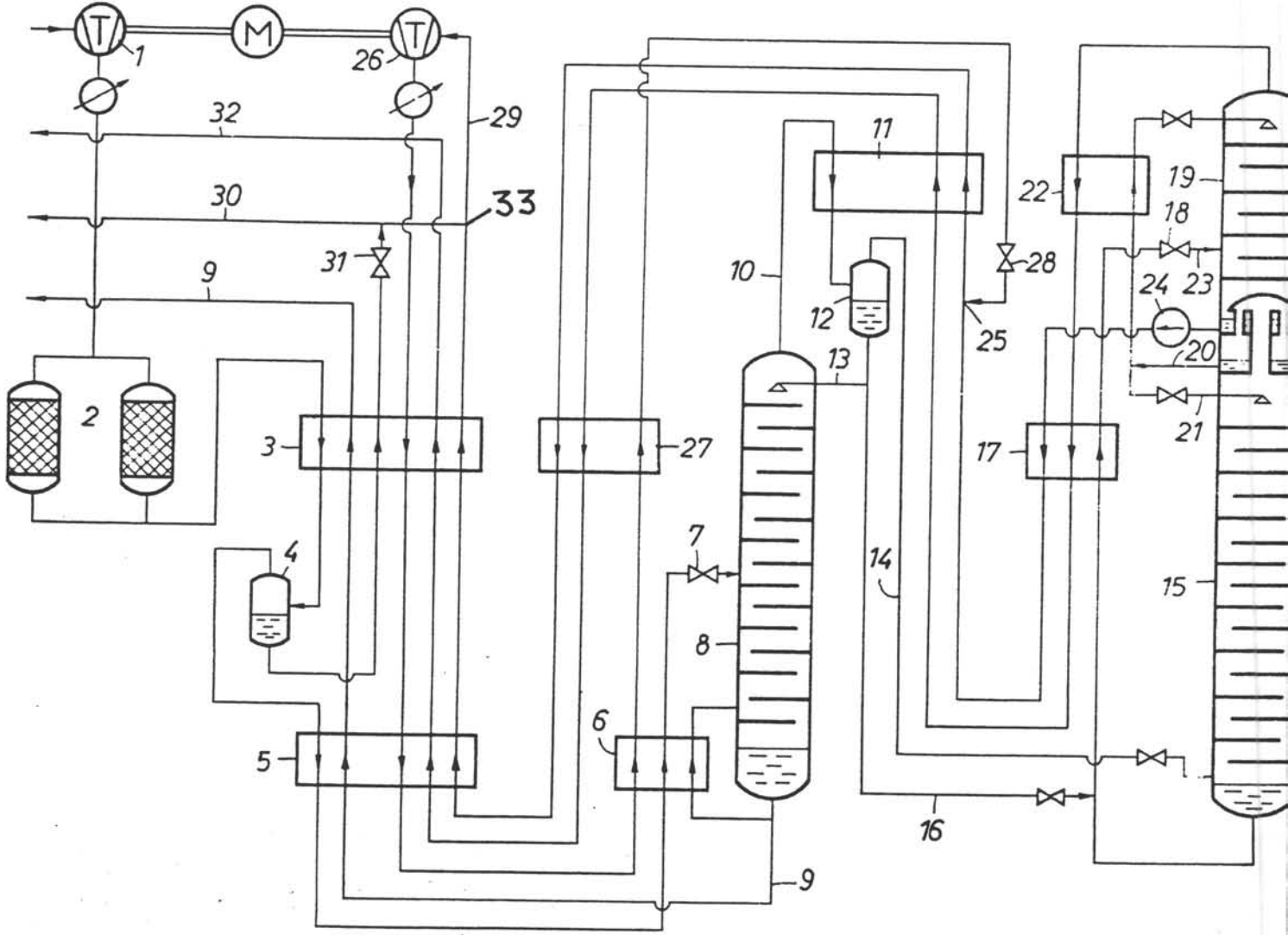


Fig. 8.2. Flowsheet of method II.

8.2.3. Membranes

A very actual way of gas separation is separation by membranes. Membranes are materials with a certain porosity, which gives them the ability of working as a sieve. For the purpose of liquid separation membranes are already being used for a long time, but gas separation by membranes in a commercial way is a recent method.

The principle of gas separation by membranes is simple. When a component has to be removed from a gas mixture in a certain room a membrane can be placed in this room in such a way that two separated rooms are formed. When the pressure in both rooms is equal, no separation will occur. The membrane behaves the same for all gas particles from both sides. Even when one component goes much easier through the membrane than the other components, this counts for both sides and netto nothing happens. Only when a difference in pressure is created between both rooms, for example by making one vacuum, separation will occur. Then there is a stream in one direction and the component that goes easy through the membrane will reach the vacuum room faster than other components, so the concentration of this component in the vacuum will increase. When the vacuum pump is set off, the separation stops and after a while the begin situation is reached again. So only when there is a difference in pressure over the membrane gas separation will occur. The different gasses flow netto to the room with the lowest concentration, but each in it's own tempo. This depends on the properties of the membrane, the type of gas and the interaction between these two factors. The passage of the gas molecules through the solid material of the membrane is attended by forming and breaking up weak physical-chemical connections which vary by the the different components that are used.

In fact two processes are important: the solution of the gas in the solid material and the transport of the gas through it. Essential is the first step: the binding of the gas on the surface of the solid membrane. The next step is breaking up the first connection and forming new one's deeper in the membrane.

If the forces of the connections are small, few gas will dissolve, while the diffusion through the membrane passes relatively fast. When the connections are too strong, then the gas molecules never leave there places and no diffusion will take place. The transport velocity through the membrane depends on the combination of these effects. This explains that a physical-chemical active compound like water easier passes the membrane than an inert gas, like nitrogen. The processes are of course dependent on temperature.

For the amount of gas (F) that passes the membrane per second a simple relation counts:

$$F = \frac{P * O * \Delta p}{d} \quad (8.1)$$

This means that the gas stream is proportional to the surface O of the membrane and to the pressure difference Δp over the membrane and inversely proportional to the thickness (d) of the separating layer. The constant of proportion P contains the structure properties of the membrane and the hidden interaction processes and is called the "constant of permeability". This constant is a function of temperature.

For a given membrane material P varies with different gasses. The ratio of the P-values of

two different gasses is called the selectivity (α) of this couple. For a given gas P varies with different membrane material.

There are two classes of membranes: organic and anorganic membranes. Organic membranes are generally polymers. For gas separation polymeric membranes with pores are not useful because of their bad selectivity. Better useful are the polymeric membranes which are close, but they can not be seen as simple sieves anymore. In fact this is a process where gas dissolves in the polymer and the gas molecules whip from one hole to another.

There are some polymeric materials that are already commercial for gas separation: polydimethylsiloxane (PDMS), polydimethylphenyleneoxide (PPO), cellulose acetate and polysulfone. The constant of permeability and the selectivity are very different for each of this materials.

For practical uses it is necessary to have a great pressure difference over the membrane, that is as thin as possible, in order to get a gas throughput as great as possible. The thickness of the membrane is the crucial factor in this process, because with a thinner membrane the pressure drop decreases, which makes the process cheaper, with regard to the energy demand of pumps or compressors. The problem however is the mechanical strength of the membrane, that of course decreases by making it thinner. This problem was solved by making an asymmetric membrane, that means asymmetric in the structure of the pores. Such a membrane consists of a very compact top layer, with a thickness of a few micrometres or less, that has good selectivity properties, layed on a thicker support layer with a thickness of about a tenth millimetre, that takes care of the mechanical strength. This support layer has a very open pore structure and a resistance for the gas that can be neglected.

Our question was what the possibilities are of a membrane with regard to nitrogen separating from natural gas, or practically the same, with regard to separating nitrogen from methane. This means we looked for a membrane that permeates nitrogen better than methane, thus a membrane with a selectivity $\alpha(N_2/CH_4)$ greater than one. A membrane that permeates methane better than nitrogen is also possible, but since the amount of methane is much greater than the amount of nitrogen in the natural gas (43.3 kmol CH_4 versus 7.5 kmol N_2 per hour) this is not practical, except for a selectivity that is very much smaller than one. For moderate selectivities the most practical way to separate two components is to let the smallest amount of the two permeate the fastest in connection with the process time.

Methane has a higher solubility than nitrogen in most liquids and polymers, so we expect that most membranes have a $\alpha(N_2/CH_4)$ less than one. This proved to be true [lit. 8.4]; cellulose acetate has $\alpha=0.73$ and polysulfone has $\alpha=1.0$. So with cellulose acetate nitrogen and methane can be separated, but this process won't be a very fast one, for the reasons described above. At this moment the technology is not yet able to separate nitrogen and methane in a commercial way, but some companies, like Separations Technology Associates [lit. 8.5], are active in developing such membranes.

8.2.4. PRESSURE SWING ADSORPTION

Introduction

Separation of gasses accounts for the major production costs in chemical, petrochemical and related industries. There has been a growing demand for economical and energy-efficient gas separation processes. Although cryogenics remain the most widely used process for this purpose, the last two decades have seen a tremendous growth in research activities and commercial applications of adsorption-based gas separation processes [lit. 6]. The increasing interest in this area is also reflected in the large number of research articles and patents published in the field of adsorption separation. Such a growing relevance of this area can be attributed to the following factors:

- Development of synthetic and more selective adsorbents.
- An improved theoretical understanding of adsorption from gaseous mixtures.
- A better understanding of multicomponent adsorber dynamics.
- Development of an efficient pressure swing cycle.

The basic principle of PSA

Separation by an adsorptive process is based on the selective accumulation of one or more components of a gas mixture on the surface of a microporous solid. When a gaseous mixture is exposed to an adsorbent for a sufficient time, an equilibrium is established between the adsorbed phase and the gas phase. The adsorbed phase often has a composition different from that of the bulk phase. The gas phase becomes richer in the less selectively adsorbed component, as depicted in Figure 8.3 on the next page.

The attractive forces responsible for adsorption are of the van der Waals type. Desorption can be achieved either by increasing the temperature of the system or by reducing the adsorbate pressure. The desorption step also regenerates the adsorbent. Thus, the adsorptive separation process consists of a cyclic sequence of adsorption and desorption steps. When desorption is achieved by decreasing the pressure, the process is called pressure swing adsorption (PSA).

One of the components is selectively adsorbed at higher partial pressure and desorbed subsequently by lowering the partial pressure. The change in the partial pressure of the component can be caused either by decreasing the total pressure, by changing the composition of the gaseous mixture, or by doing both simultaneously. Figure 8.4 on the next page shows the amount of gas adsorbed as a function of partial pressure at a particular temperature. The amount adsorbed decreases along the curve from point A to point B with a decrease in the partial pressure of the component.

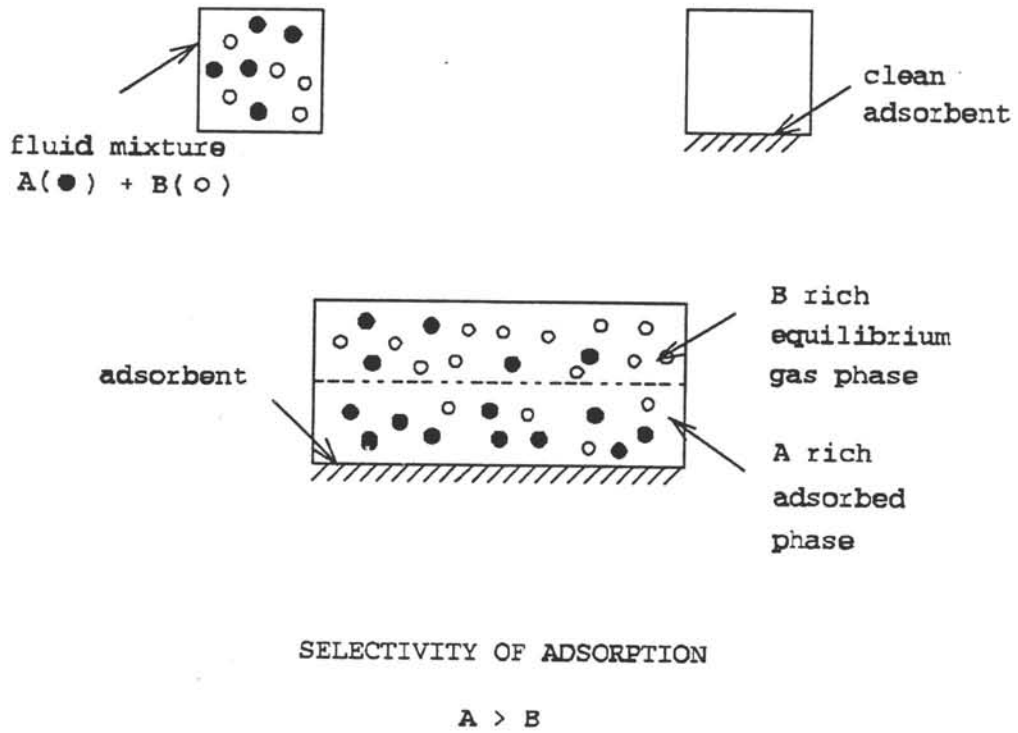


Fig. 8.3. Principle of adsorptive separation.

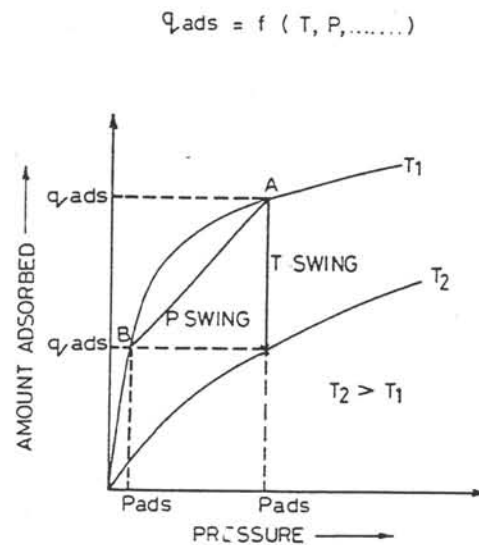


Fig. 8.4. Adsorption isotherms showing pressure and temperature swings.

The adsorption capacity available for separation in PSA is the difference in the capacities for the given component at the two pressures. However, in practice, due to the heat of adsorption, there is an associated change in the temperature of the system and hence the working PSA capacity is slightly different, as depicted in Figure 8.5.

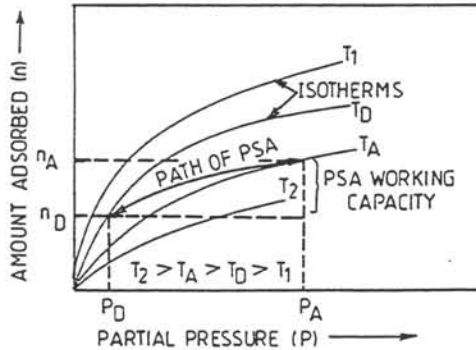


Fig 8.5. Adsorption isotherms showing pressure and temperature swings and PSA working capacity.

The primary steps in PSA

The PSA system in its basic form consists of two beds which are alternately pressurized and depressurized according to a programmed sequence. A two-bed system is shown in Figure 8.6 on the next page. The basic steps of the PSA process are shown in Figure 8.7 on page 8-13 and summarized in the following.

1. Feed pressurization

The feed gas is compressed into an adsorbent-fixed bed from one end, with the other end being closed. The gas freed from the adsorbed components is accumulated at the closed end.

2. Product release

At the upper operating pressure, the adsorbate-free gas, usually the product, is withdrawn from the far end of the bed while feed flow is maintained. Three distinct regions develop inside the adsorbent bed, as shown in Figure 8.7 on page 8-13.

- Near the feed entry the bed is saturated with the adsorbate and the gas phase has the composition of the feed.
- Near the exit the bed is still adsorbate free and the gas phase has the product composition.
- The region lying between the above two ends is called the mass transfer zone. This is where adsorption is occurring, and the gas phase composition changes rapidly along the axial position. The adsorption front moves along the bed as more feed

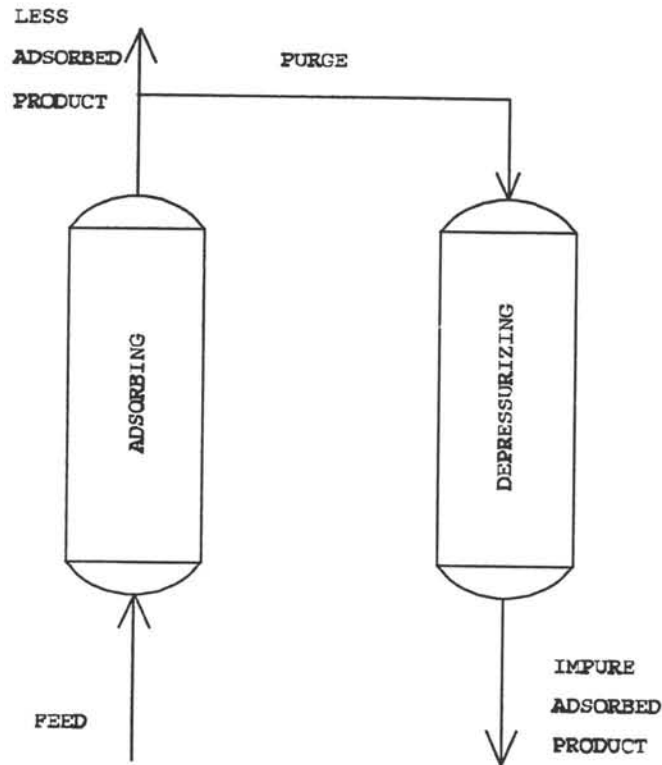


Fig. 8.6. Schematic diagram of a two-bed PSA process.

is introduced. Eventually the bed is completely saturated, with the mass transfer zone having reached the end of the bed, at which point the feed is said to breakthrough into the product stream.

3. Depressurization

To regenerate the saturated bed, the next step involves reducing the operating pressure (blowdown or depressurization). The adsorbate is largely desorbed into the gas phase, released as a waste stream, or collected as a product in case it is valuable.

4. Low pressure purge

To complete the regeneration of the bed, the adsorbent is purged with product quality gas at low pressure, usually countercurrent to the feed flow from the product end. In this case also, distinct regions form and the purge gas will eventually breakthrough the waste.

PSA is well suited to rapid cycling, and it generally operates at a relatively low adsorbent

loading because selectivity is greatest in the Henry's law region. A high pressure operation is desirable because it minimizes the purge loss, but this gain is offset by the greater blowdown losses for a high pressure system. A low temperature operation is needed to maximize capacity and selectivity, but cooling below the ambient temperature is generally not economical. Some of the points that make separation by PSA economically attractive are the following:

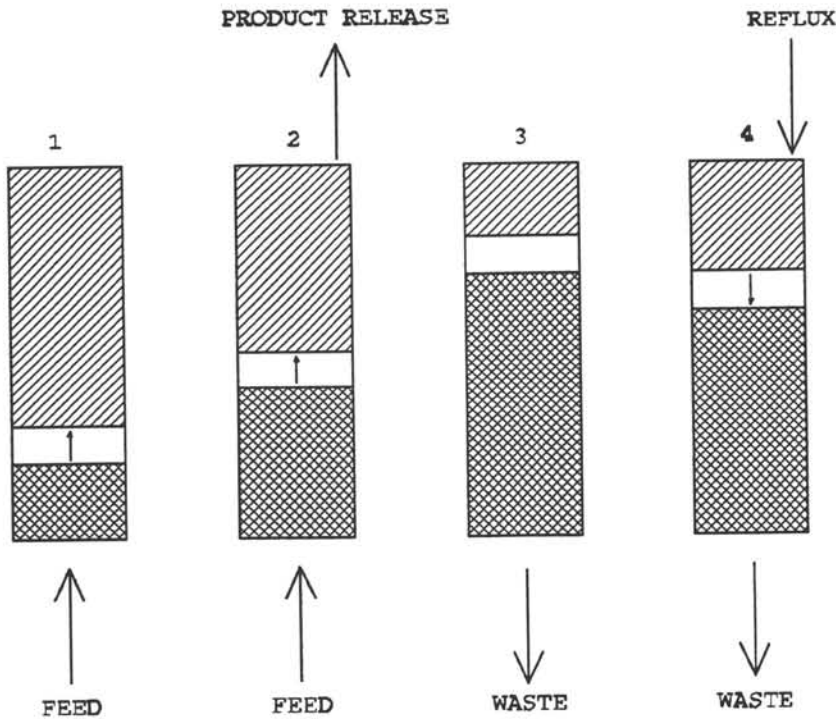


Fig. 8.7. The basic steps in a PSA process.

- A high concentration of the desired and less strongly adsorbed component in the gas mixture.
- Moderate purity requirement for the desorbed product.
- A low heat adsorption (less than 30 kJ/mol) of adsorbable component to facilitate rapid desorption.
- A linear adsorption isotherm for the strongly adsorbed component.
- Small adsorption/desorption pressure ratio.

Major developments in the PSA process

One of the major disadvantages of the earliest PSA system was low recovery of the product gas. For example, in the separation of air for oxygen production, the recovery was rarely

above 25% in a simple two-bed system. To overcome this disadvantage, a variety of process modifications have been made. A critical review of the literature during the past two decades shows that the major part of research in this area has been directed towards improving the process economy by enhancing the product recovery. Some of the most important developments are briefly described in the following.

Pressure equalization step

The major operating cost in PSA is the energy required for pressurization of the feed. In the early PSA systems a large part of this energy was lost during blowdown. Berlin[lit. 2] suggested a pressure equalization step to conserve this energy. After the first bed has been purged and the second bed has completed the adsorption step, instead of blowing down the second bed directly, both the beds are interconnected to equalize the pressure. The first bed is partially pressurized with the product-rich gas from the outlet of the second bed. Following pressure equalization, the beds are disconnected and the first bed is pressurized with the feed gas while the second bed is vented to complete blowdown. Inclusion of this step in the PSA process sequence has not only minimized energy losses but also resulted in a substantial increase in recovery.

Multibed system

More complex processes, including three- and four-bed systems, have been developed to take advantage of the pressure equalization step. A four-bed PSA system is shown in Figure 8.7 on the next page. In this system, one bed is in the adsorption step and the other three beds are in various stages of repressurization, depressurization or purging. The principle is an extension of the Berlin [lit. 2] system.

The process operates at two intermediate pressures between the feed pressure and the exhaust pressure (usually atmospheric). At the end of the adsorption step, bed 2, which is at high pressure, is connected at the discharge end to column 4, which has just completed the purge step and is essentially at atmospheric pressure, and the pressures are equalized. In this way the product-rich gas is conserved and used for partial pressurization of bed 4. Final repressurization of bed 4 is accomplished by using product gas from bed 1, and the feed is connected to the inlet of column 4. A fraction of the remaining gas from bed 2 is used for reverse-flow purging of bed 3. When the pressure in bed 2 has fallen to required level, beds 2 and 3 are disconnected and the residual gas from bed 2 is vented from the bed inlet. Bed 2 is then purged in reverse flow with gas from bed 1 and repressurized to the first intermediate pressure with gas from bed 4, which has just completed the adsorption step. The final repressurization is accomplished by using product gas, and the feed is then connected to the inlet of bed 2.

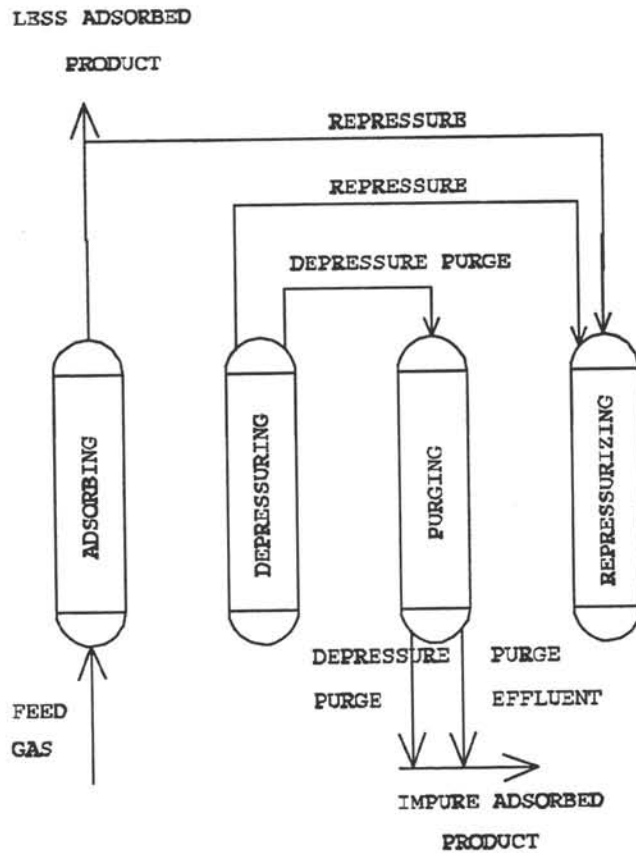


Fig. 8.7. Schematic diagram of a four-bed PSA process.

The advantages over the two-bed process at the cost of a more complex flow sheet are the following:

- Considerable reduction in power consumption.
- High product recovery.
- High purity product.
- Constant product flow rate.

The use of more than one type of adsorbent in a single PSA unit, a guard bed to remove impurities, and rapid cycling to improve the adsorbent productivity are some of the other recent improvements. Other developments in the process include vacuum swing adsorption (VSA), polybed PSA (up to 10 beds), and multiproduct recovery.

Adsorbents used in PSA

The choice of an adsorbent for PSA application is governed by its adsorption capacity and selectivity for the desired component. The selectivity or the separation factor of component A over B is given by

$$\alpha_{A/B} = \frac{x_A y_B}{y_A x_B} \quad (8.2)$$

where x and y are the adsorbed and the gas-phase concentrations. Simulation studies for countercurrent adsorption separation processes reveal that for the process to be economical, the minimum acceptable intrinsic separation factor for the desired component is about 3. With a separation factor of less than 2, it is difficult to design a satisfactory process.

A study by T.C. Frankiewicz and R.G. Donnelly [lit. 7] shows that nitrogen is weakly adsorbed on the surfaces of numerous zeolites. Methane is a weakly polarizable molecule which, as a result of this polarizability, is also weakly adsorbed on zeolites. In general, the heat of adsorption for CH_4 is equal or slightly greater than the heat of adsorption for N_2 and as a result an adsorption separation of these gasses either does not occur or happens by a weakly selective CH_4 adsorption. In the 1950s, there was discovered that N_2 could be selectively adsorbed from a CH_4/N_2 mixture by cooling a 4A molecular sieve to between 0°C and -78°C . However, this system was limited by slow kinetics, a weak selectivity and the need to thermally regenerate the sorbent.

In the begin 1980s, the naturally occurring zeolite clinoptilolite (clino) was found which can selectively adsorb N_2 in the presence of CH_4 near ambient temperature. The zeolite designated ZBM-15 has the largest ability to separate a N_2/CH_4 mixture.

Given the ability of the clino ZBS-15 to effect CH_4/N_2 separation, the named study reviews the results from several series of pressure swing CH_4/N_2 separation experiments and shows how to design a process to use this effect. An economic evaluation suggested the process would be economically viable for small plants (2×10^6 ft³/day).

The pressure swing separation of CH_4 and N_2 .

In order to investigate the process aspects of the pressure swing separation of CH_4 and N_2 the apparatus depicted in figure 8.8 on the next page, was constructed. Valves V1 through V5 are fitted with electric switching mechanisms that are interfaced to a computer. The two pressure transducers, three mass flow meters, and column temperature are also monitored by the same computer. All gas effluent from the apparatus through vent or vacuum is sampled through a molecular leak valve for analysis by the quadrupole mass spectrometer. The mass spectrometer in turn reports these compositions to the computer. Once a run is initiated, the apparatus is under complete control of and monitored by the computer.

The back pressure regulator (BPR) is set to give the desired pressure in the adsorption column (C1 or C2) being used. Although it is possible with this apparatus to involve both columns

V = Electrically Operated Valve
 PT = Pressure Transducer
 FM = Mass Flow Meter
 BPR = Back Pressure Regulator
 C = Adsorption Column
 F = In-line Gas Filter

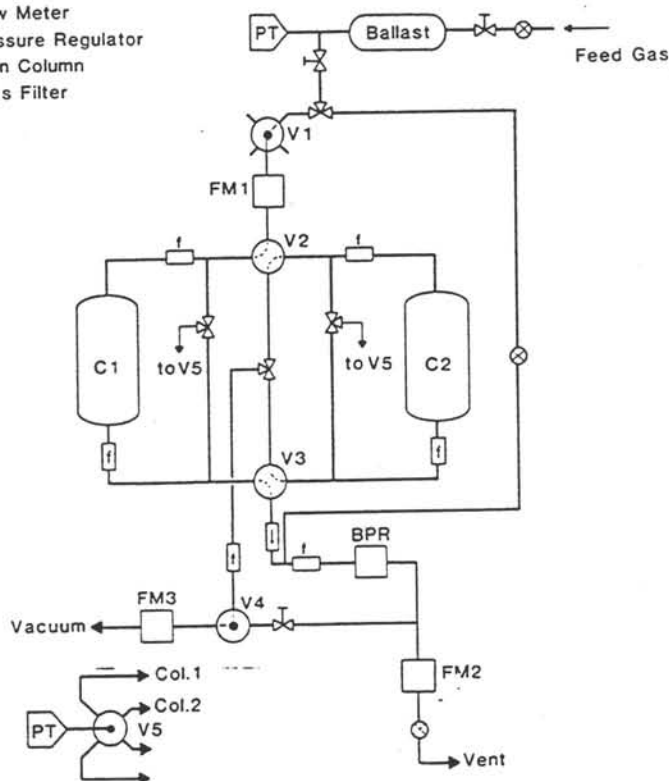


Fig. 8.8. A schematic drawing of the computer controlled pressure swing apparatus.

alternately in a separation experiment, this was not done during the experiments described so that all vent and vacuum regeneration gasses could be analyzed for every cycle in the run. During a pressure swing run, gas enters the adsorption column, C1 or C2, through valve V2. Valve V3 is set to allow gas to exit the selected adsorber through the back pressure regular (BPR) as soon as the desired operating pressure is achieved. For depressurization, V3 is switched to allow the column exit gas to bypass the BPR and be vented through valve V4. V4 is a vacuum/vent selector and V5 allows a single pressure transducer to monitor system pressures at up to four selected points.

A sample data plot is shown in figure 8.9 on the next page. Feed gas is 40% N_2 and 60% CH_4 . A mass balance is calculated for CH_4 and N_2 for each step in the process as well as for the complete adsorption cycle. The mass balance includes a Van der Waals calculation of the amount gas stored in column voids and the volume of gas adsorbed on the zeolite.

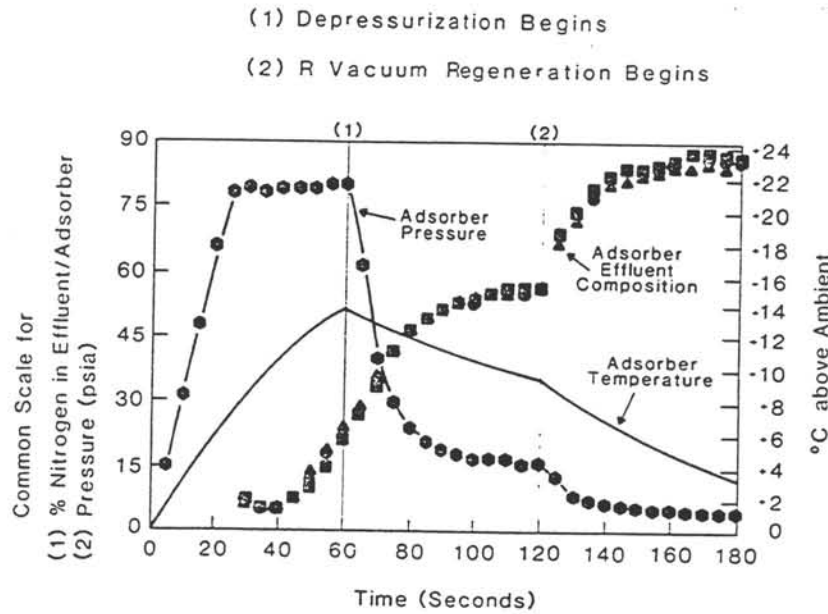


Fig. 8.9. Adsorber parameters are plotted versus time for a pressure swing run. Note the common scale for %N₂ and adsorber pressure.

Table 8.1 on the next page lists the independent variables and their respective ranges which were included in the described study. Dependent variables can be defined in numerous ways depending upon the objectives of the particular experimental series being conducted. Typically, a pressure swing adsorption cycle program is input to the apparatus's computer control program, e.g., per figure 8.9, and the cycle repeated under computer control until the experimental objective is achieved. For this work, 6 to 24 cycles were normally used with total run times up to 8 hours.

Figures 8.10 on the next page and figure 8.11 on page 8-20 illustrate changes in two dependent variables: dynamic N₂ adsorption capacities and CH₄/N₂ separation factors. Independent variables are column temperature, operating pressure, and time allowed for vacuum regeneration. This experimental series used a constant feed rate of 6.0 l/min over a time of 1.00 min into a 1 inch diameter x 24 inch long adsorber filled with 180 g of zeolite. Column depressurization took place for 1.00 min. and this was followed by a variable length vacuum regeneration.

Table 8.1. Operational variables and variable ranges studied for the pressure swing separation of CH₄ and N₂.

Gas Feed Rate	16.7 - 50.0 scc/g. of zeolite-min.
Feed gas composition	10 - 40% N ₂
Adsorber Temperature	0° - 55°C
Operating Pressure	75 - 300 psia
Adsorber Granule Size	.18 - 3.5 mm
Column Regeneration	0 - 20 min. vacuum or CH ₄ purge

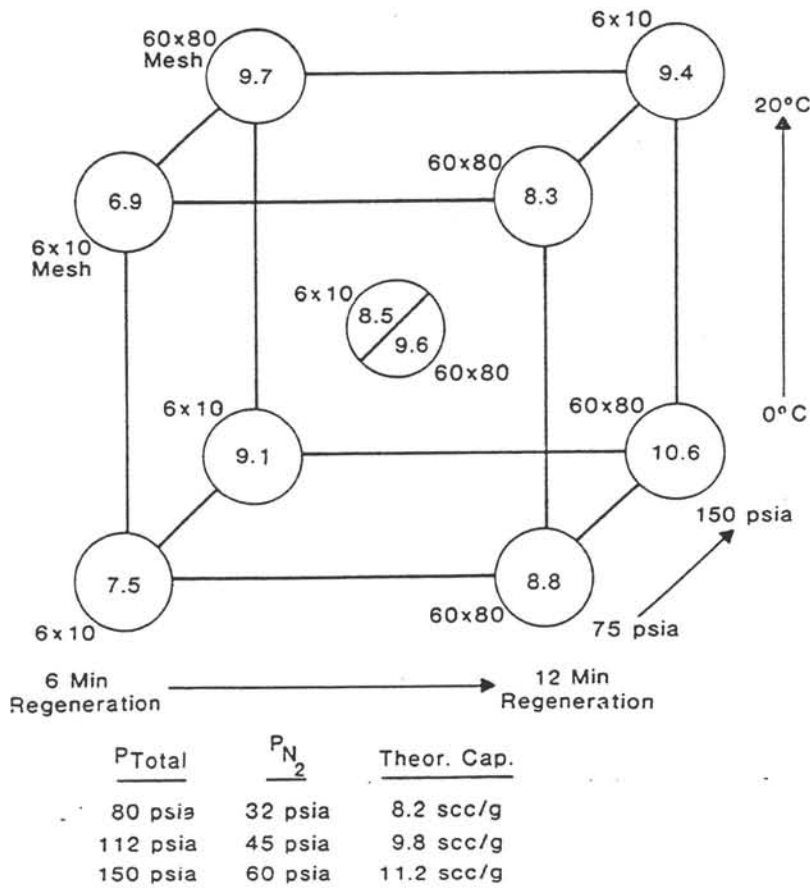


Fig. 8.10. Average SCC of N₂ adsorbed per gram of the clino ZBS-15 for a 3 variable factorial design experimental series. Feed was 33.3 SCC/g of zeolite per cycle of 40% N₂/60% CH₄ gas.

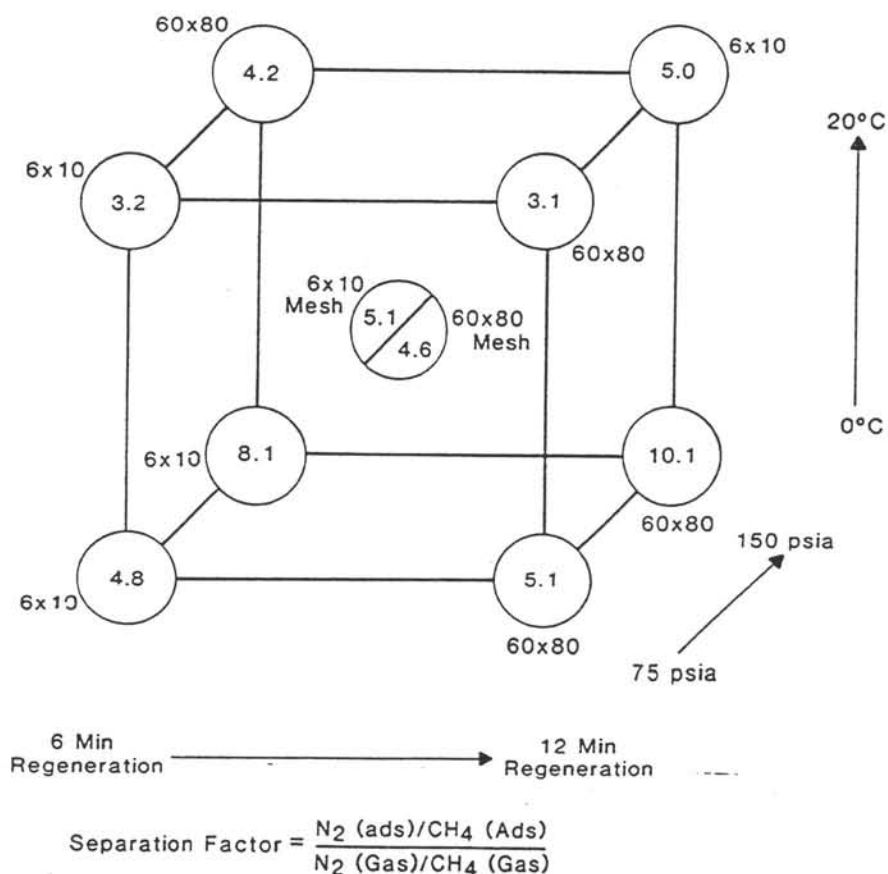


Fig. 8.11. CH₄/N₂ separation factors for a 3 variable factorial design experimental series. Feed was 33.3 SCC/g of zeolite per cycle of 40% N₂/60% CH₄ gas.

Adsorbent particle sizes, although not a design variable, did affect some results for hydrodynamic reasons. As can be seen from the centerpoints, a slight deterioration in column performance resulted with the switch from 60x80 to a 6x10 mesh adsorbent. Although this effect was real and reproducible, the effects on the three design variables were generally more significant.

At the bottom of figure 8.10, capacities derived from the clino N₂ adsorption isotherms are also listed. The factorial design results suggest that over 90% of the theoretical N₂ adsorption capacity can be dynamically used in a pressure swing process provided adsorbent regeneration is adequate. The column temperature does not strongly effect dynamic N₂ adsorption capacity,

but does influence N_2/CH_4 separation factors in the process. This results from a slowing of CH_4 kinetics as temperature is decreased.

The longest step in the pressure swing cycle for CH_4/N_2 over clinoptilolite is regeneration. Because this step involves vacuum and represents unproductive time in the process, regeneration was closely studied. Regeneration times were varied from 10 sec. to 20 min. and regeneration mode was changed from vacuum (approx. 20 torr system pressure) to quiescent to CH_4 purge. The results for 5 of these runs are summarized in figure 8.12 for a 20% N_2 , 80% CH_4 feed gas. It appears that a vacuum regeneration time of at least 6 minutes is required to maintain product quality over multiple cycles. Other modes of regeneration were not successful and it appears that a process design cannot reasonably incorporate anything but vacuum regeneration.

Process design and economics

Using the detailed mass balance data for the experiments described above, a commercial process design was developed and subjected to an economic analysis. The process involves the use of 10 adsorbed beds sized to handle 2.0 or 20.0×10^6 SCF/day of feed gas containing 20-30% N_2 and 70-80% CH_4 . The product varies from 90-95% methane depending upon the feed composition. Pressure swing separation costs were found to be \$ 0.35/ 10^6 Btu for the larger plant and \$ 1.20/ 10^6 Btu for the smaller plant. Corresponding costs for cryogenic separation of the same feedstocks were estimated to be \$ 0.29/ 10^6 Btu for the 20.0×10^6 SCF/day plant and \$ 1.66/ 10^6 for the smaller plant.

Thus pressure swing separation of CH_4/N_2 appears to be viable for small plants.

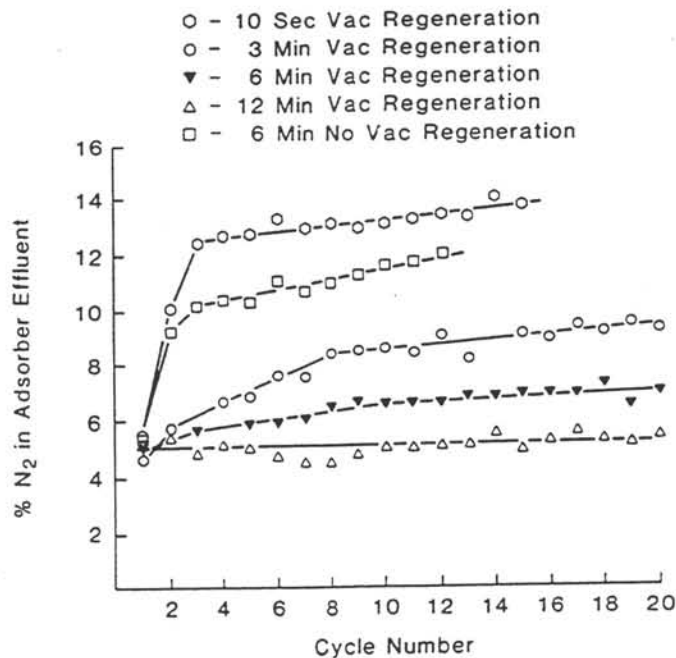


Fig. 8.12. Product quality from an adsorber varies as the regeneration time is varied. Feed gas is 20% $N_2/80\%$ CH_4 .

Potential adsorbents

Although there is some room for improvement in the PSA process, the potential gains in process economics in the future are likely to come from the development of new and improved adsorbents. Major advances have been made in the synthesis of new zeolite structures since the development of the first synthetic zeolite A. The development of silicate, ZSM type, metallosilicates, aluminophosphates ALPOs, silicoaluminophosphates SAPOs, metalloaluminophosphates MeAPOs, particularly the more recent synthesis of VPI-5, aluminoborates, and beryllphosphates shows that the field of zeolite synthesis is yet to reach its maturity even after the considerable synthetic efforts of the last 40 years. At present, mainly A, X, mordenite, and some natural zeolites are being commercially used as adsorbents in the separation of gasses by PSA. The commercial potentials of such zeolites as silicates, ZSM type, and the ALPOs are yet to be explored. These present interesting and varied surfaces, and they call for extensive studies from the adsorption point of view. For example, silicate has a highly hydrophobic surface, while A and X are strongly hydrophilic in nature. ALPOs have an intermediate hydrophilicity. This class of zeolites may become specialized adsorbents for separation processes. Recent studies have shown that silicalite is a potential adsorbent for the separation of N_2/CH_4 and CO_2/CH_4 , with selectivities for N_2 and CO_2 being more than 3, as shown in Table 8.2.

Table 8.2. Separation factors for sorption of CO_2 , N_2 , and CH_4 on silicate.

Temperature (K)	CO_2	CO_2/CH_4	CH_4/N_2
273	20	3.5	3.5
283		3.0	2.8
298		2.7	2.7
323		2.5	2.2
343	4	2.0	-

8.3. Conclusion

In order to remove nitrogen from the natural gas feed, we studied four different methods.

The first, absorption, is a suitable method for separating one specific component. For our situation it is not, because there is no suitable absorbent available for N_2/CH_4 separation.

The second method, cryogenic separation, is a method that is very often used in the separation of nitrogen from methane. Two possibilities were studied: a method which makes use of one column, and a method that uses three small columns. Both methods were patented more than twenty years ago.

The first method is able to remove about 81 percent of the nitrogen. The removed nitrogen including a little methane and some other components are mixed with the total gas net of the entire plant. The difference in concentrations will be very small, so they don't have consequences for other processes. This method costs about 8 million Dfl. No costs for the fact that the method is patented have to be calculated, because the patent is older than twenty years, so the patent is not protected anymore.

The second cryogenic method removes 79.6 percent of the nitrogen; 3.3% of the N_2 remains in the feed. The methane-purity raised from 81.3% to 91.8%, and only 0.03 kmol/hr CH_4 (0.068% of the total feed) remains in the removed nitrogen stream. The latter stream is mixed with the total gas net. Like the first method, no costs for the fact that the method is patented have to be calculated, because the patent is older than twenty years.

The costs of both methods could not be estimated, since the heat exchanging surfaces of the multi-flow heat exchangers, which are necessary for cost calculations, could not be estimated. Because in both methods various multi-flow heat exchangers are used, the calculating of the costs of the plants became impossible.

The third nitrogen-removing method that was studied is separation by membranes. Compact polymeric membranes are able to separate nitrogen from methane, but since their selectivity is not good enough, the separation is slow and not very suitable. This is also the reason of the fact that membranes for N_2/CH_4 separation are not commercial yet.

The last investigated method is pressure swing separation. According to the experiments described, pressure swing adsorption is viable for small plants (2×10^6 ft³/day, which is about 0.6 m³/sec.). Since the present CO-plant has a lower capacity, is pressure swing adsorption a possible method to remove nitrogen from natural gas.

Literature

- [8.1] Messer Griesheim GmbH, Frankfurt, West-Germany Patent 2,022,954 (2 Dec. 1971).
- [8.2] Messer Griesheim GmbH, Frankfurt, West-Germany Patent 2,055,229 (18 May 1972).
- [8.3] Linde A.G., Wiesbaden, Dutch Patent 7205577 (9 November 1972).
- [8.4] E.W. Funk and N.L. Norman, Purification of natural gas by membranes, *Gas Separation Technology*,
- [8.5] E.J. Hoffman, K. Venkataraman and J.L. Cox, Membrane separation for subquality natural gas, *Energy Progress*, March 1988, Vol.8, No. 1.
- [8.6] R.V. Jasra, N.V. Choudary and S.G.T. Bhat, Separation of gases by pressure swing adsorption, *Separation science and technology*, 26(7), pp. 885-930, 1991.
- [8.7] T.C. Frankiewicz and R.G. Donnelly, Methane/Nitrogen separation over the zeolite clinoptilolite by the selective adsorption of nitrogen, American Chemical Society, 1983.

9. PLANT COST CALCULATIONS

9.1. Investment cost

There are several methods to determine the investment cost. An more accurate method needs more detailed information than a method that is not so accurate. The following methods in order of accuracy can be used:

- sales method (first indication)
- scale up method (to determine the influence of capacity)
- 'step' method (to compare different pathways), Zevnik-Buckanan, Wilson, Taylor, etc.
- 'factor' method (a global absolute determination), Lang, Holland, Miller, Guthrie, etc.
- office type (a more detailed absolute method)
- detailed method

The factor method is used here to compare new possibilities for the CO plant (improvements) with the already existing plant and to determine the investment cost. The method requires only the available information about the process equipment, furthermore it is relatively easy and quick to use.

9.1.1. The factor method

The investment cost of a plant, determined with the factor method, is based on the purchase costs of the most important pieces of equipment of the process, these are:

- furnaces
- pressure vessels
- columns
- heat exchangers
- compressors
- pumps

To determine the purchase costs of these pieces of equipment the equations of Dr. Ž. Olujić (Lit.9.1) are used; for comparison the prices from Webci (Lit.9.2) are used.

The purchase cost equipment (PCE) is multiplied with two factors, the f_a factor, to determine the Physical Plant Cost (PPC), and the f_b factor, to estimate the Fixed Capital (FC). Thus:

$$\begin{aligned} \text{PPC} &= f_a * \text{PCE} \\ \text{FC} &= f_b * \text{PPC} \end{aligned}$$

The f_a and the f_b factor can be found in Table 9.1 (Lit.9.3). For the CO plant no storage facilities are needed, and, as the process streams are all fluids, the f_a factor equals 3.25 and f_b equals 1.45. To determine the total investment cost the working capital (5 to 30% of the

fixed capital) must be added to the fixed capital.

Table 9.1. Typical factors for the estimation of project fixed capital cost

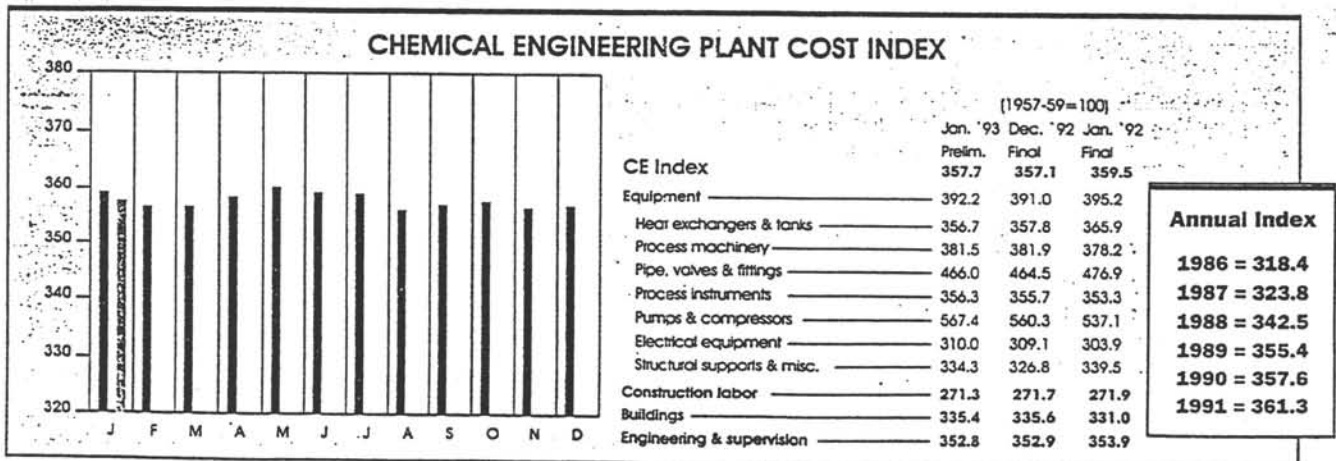
Item	Process type		
	Fluids	Fluids-solids	Solids
1. MAJOR EQUIPMENT, TOTAL PURCHASE COST			
PCE	PCE	PCE	PCE
f_1 Equipment erection	0.4	0.45	0.50
f_2 Piping	0.70	0.45	0.20
f_3 Instrumentation	0.20	0.15	0.10
f_4 Electrical	0.10	0.10	0.10
f_5 Buildings, process	0.15	0.10	0.05
f_6 Utilities	0.50	0.45	0.25
f_7 Storages	0.15	0.20	0.25
f_8 Site development	0.05	0.05	0.05
f_9 Ancillary buildings	0.15	0.20	0.30
2. TOTAL PHYSICAL PLANT COST (PPC)			
PPC = PCE (1 + f_1 ... + f_9) = PCE ×			
	3.40	3.15	2.80
f_{10} Design and Engineering	0.30	0.25	0.20
f_{11} Contractor's fee	0.05	0.05	0.05
f_{12} Contingency	0.10	0.10	0.10
FIXED CAPITAL = PPC (1 + f_{10} + f_{11} + f_{12}) = PPC ×			
	1.45	1.40	1.35

* Omitted for minor extensions or additions to existing sites.

9.1.2. The purchase cost of the equipment

The purchase cost estimation by Dr. Ž. Olujić is given in US dollars (\$). The correlations use an I factor to correct for the annual inflation. This index I can be found in the Chemical Engineering Plant Cost Index of the Journal 'Chemical Engineering' (see Table 9.2).

Table 9.2. The chemical engineering plant cost index.



For the different pieces of equipment the following I values were found (December 1992):

Table 9.3. The I values of different pieces of equipment

Type of equipment	CE index
Furnace	357.8
Pressure vessel	357.8
Column	357.8
Internals	357.8
Packing	391.0
Heat exchanger	357.8
Compressor and pump	560.3
Driver	381.9

9.1.2.1. The furnace

The correlation developed for furnaces includes both the furnaces for heating the process fluids and the cracking and the reforming furnaces:

$$C_{fu} = C_{fu0} * F_m * F_p * F_t * (I/336.2) \quad (9.1)$$

Here C_{fu0} (US \$) is the base cost (mid 1985), F_m (-) the material of construction factor, F_p (-) the design pressure cost factor, F_t (-) the Chemical Engineering Plant Cost Index and 336.2 the CEPC index for this type of equipment in 1985. The corresponding cost factors can be found in the following table, and the base cost is correlated against the heat transfer duty ($3000 < Q(\text{kW}) < 150000$):

$$C_{fu0} = \exp(5.639 + 0.85 * \ln Q) \quad (9.2)$$

Table 9.4. Cost factors of furnaces

Material of construction	F_m	Design pressure (bar)	F_p	Type of furnace	F_t
Carbon steel	1	10	1	Heater	1
Alloy steel	1.36	50	1.1	Pyrolysis	1.1
Stainless steel	1.6	100	1.25	Reformer	1.5

9.1.2.2. The pressure vessel

Pressure vessels are columns, drums, tanks and reactor shells. The cost of a pressure vessel is calculated from:

$$C_{pv} = C_{pvo} * F_m * F_p * (l/336.2) \quad (9.3)$$

The correlation for the base cost is:

$$C_{pvo} = (a + b*l)*d^{1.1} \quad (9.4)$$

where l (m) is the length or height of the vessel and d (m) the corresponding internal diameter. The values of the coefficients a and b are given in Table 9.5.

Table 9.5. Vessel coefficients.

Vertical vessels	a	b	Horizontal vessels	a	b
1.5 < l (m) < 5	1500	1100	1.5 < l (m) < 6	890	705
5 < l (m) < 90	1294	1141	6 < l (m) < 25	512	768

Table 9.6. Cost factors for material of construction and design pressure

Material of construction	F_m	Pressure (bar)	F_p
Carbon steel (CS)	1	1 - 5	1
CS, SS-lined	2.3	5 - 10	1.1
Stainless steel (SS)	3.7	10 - 20	1.2
CS, Monel-lined	3.9	20 - 30	1.4
Monel	6.4	30 - 40	1.6

9.1.2.3. The column

The shells of columns may be considered as vertical pressure vessels, therefore correlation (9.3) can be used for prediction of the corresponding price. In addition the cost of column internals has to be estimated. Hence, the correlation for the cost of a distillation, absorption or desorption column is:

$$C_{co} = C_{pv,vert} + C_{int} \quad (9.5)$$

where index 'vert' denotes vertical pressure vessel and index 'int' the cost of column internals which differs for tray and packed columns.

9.1.2.3.1. Correlations for cost of trays

The cost of trays may be estimated by:

$$C_{\text{int}} = N_{\text{tr}} * C_{\text{tr}} * F_{\text{m}} * F_{\text{nt}} * F_{\text{tt}} * (I/336.2) \quad (9.6)$$

where N_{tr} (-) is the number of trays in the column, F_{nt} (-) cost factor for the number of trays and F_{tt} (-) cost factor for the tray type. C_{tr} (US \$) is the base cost for a valve tray of carbon steel, which may be estimated for columns with internal diameters d (m) from 0.3 to 4 m by:

$$C_{\text{tr}} = 58.7 + 88.4*d + 52.9*d^2 \quad (9.7)$$

Values of characteristic cost factors are given in Table 9.7.

Table 9.7. Characteristic cost factors of trays.

Tray type	F_{tt}	Material of construction	F_{m}	Number of trays	F_{nt}
Sieve	1	CS	1	20	1
Valve	1.45	SS	1.7	10	1.5
Bubble cap	2.90			7	2
Grid	0.90			4	2.5
				1	3

9.1.2.3.2. Cost of packings

The prices for random packings obtained in 1990 are ± 5000 f/m³ or ± 2940 \$/m³ (1 \$ = f1.7) for pall rings (stainless steel) with a diameter of 50 mm. For distributors the prices are ± 6000 f/m³ or ± 3530 \$/m³. For cost updating the given prices must be multiplied with (I/393), where I is the CE index of packings (I=391).

9.1.2.4. The heat exchanger (shell and tube)

The cost of a shell and tube heat exchanger is calculated from:

$$C_{\text{he}} = C_{\text{heo}} * F_{\text{p}} * F_{\text{t}} * (I/336.2) \quad (9.8)$$

C_{heo} (US \$) is the base cost, which is given as a function of the transfer area A (m²).

$$C_{\text{heo}} = \exp(a + b*\ln A) \quad (9.9)$$

The values of coefficients a and b as well as the values of corresponding cost factors are given in Tables 9.8 and 9.9.

Table 9.8. The coefficients of shell and tube heat exchangers.

Material of construction Shell/Tube		a	b
CS/CS	10 < A(m ²) < 900	7.085	0.656
CS/Brass	10 < A(m ²) < 900	6.980	0.7133
CS/SS	10 < A(m ²) < 900	6.587	0.9955
SS/SS	10 < A(m ²) < 60	7.439	0.8861
SS/SS	60 < A(m ²) < 900	8.6063	0.601

Table 9.9. The cost factors of shell and tube heat exchangers.

Design pressure (bar)	F _p	Type of heat exchanger	F _t
1 - 10	1	Kettle reboiler	1.4
10 - 20	1.1	Floating head	1
20 - 30	1.25	U-tube	0.85
30 - 50	1.3	Fixed tube	0.8
50 - 70	1.5		

9.1.2.5. The compressor

The cost of gas transporting equipment does not include a driver, whose cost must be estimated separately. The correlation includes the cost of the base device and a material of construction cost factor:

$$C_{\text{com}} = C_{\text{como}} * F_m * (I/418.3) \quad (9.10)$$

where the base cost C_{como} is correlated against the power consumption (kW):

$$C_{\text{como}} = [\exp (a + b * \ln P)] * 0.6 \quad (9.11)$$

The values of the coefficients corresponding to the common compressor types and the cost factors for the material of construction are given in Table 9.10.

Table 9.10. Coefficients and cost factors of the compressor.

Type of compressor	a	b	Type of compressor	F _m		
				CS	SS	Ni-alloys
Axial, centrifugal, reciprocating (30 < P < 10000)	6.628	0.968	Axial	1	5.8	14.1
			Centrifugal	1	4.5	10.5
			Reciprocating	1	5	12
Rotary (30 < P < 1000)	7.817	0.618	Rotary	1	4	9

9.1.2.5.1. The driver (electric motor)

For standard electric motors the following correlation may be used:

$$C_{em} = C_{emo} * F_t * F_{red} * (I/331.2) \quad (9.12)$$

where F_{red} is a factor which takes into account the cost of the reduction in the number of revolutions. Common value is 1 but in the cases when the reduction is needed then the value 1.35 should be used. The values of the F_t factor are 1.4 for explosion proof motors and 0.75 for open drip proof motors. The reference price is for a totally enclosed electric motor and is correlated against the required brake power ($0.5 < P(\text{kW}) < 5000$):

$$C_{emo} = \exp [5.33 + \ln P * (0.3 + \ln P * (0.162 - 0.014 * \ln P))] \quad (9.13)$$

9.2. Operating costs

An estimation of the operating costs is needed to judge the viability of a project and to choose between possible alternative processing schemes. Generally the operating costs are divided into two groups, the fixed costs and the variable costs.

- Fixed costs:
1. Maintenance (labour and materials)
 2. Operating labour
 3. Laboratory costs
 4. Plant overheads
 5. Capital charges
 6. Rates (and any other local taxes)
 7. Insurance
 8. Licence fees and royalty payments

- Variable costs:
1. Raw materials
 2. Miscellaneous operating materials
 3. Utilities
 4. Shipping and packaging

The costs listed above are direct costs of producing the product at the plant site. In addition to these costs, the site will have to carry its share of the Company's general operating expenses. These will include:

1. General overheads
2. Research and development costs
3. Sales expense
4. Reserves

All these various components of the operating costs are summarized in Table 9.11. This table will be used to estimate the total production costs.

Table 9.11. Summary of the operating costs.

<i>Variable costs</i>	<i>Typical values</i>
1. Raw materials	from flow-sheets
2. Miscellaneous materials	10 per cent of item (5)
3. Utilities	from flow-sheet
4. Shipping and packaging	usually negligible
Sub-total A
<i>Fixed costs</i>	
5. Maintenance	5-10 per cent of fixed capital
6. Operating labour	from manning estimates
7. Supervision	20 per cent of item (6)
8. Plant overheads	50 per cent of item (6)
9. Capital charges	15 per cent of the fixed capital
10. Insurance	1 per cent of the fixed capital
11. Rates	2 per cent of the fixed capital
12. Royalties	1 per cent of the fixed capital
Sub-total B
Direct production costs A + B
13. Sales expense	20-30 per cent of the direct
14. General overheads	production cost
15. Research and development	
Sub-total C
Annual production cost = A + B + C =
Production cost £/kg =	$\frac{\text{Annual production cost}}{\text{Annual production rate}}$

9.3. Calculations of the investment cost

9.3.1. Design process

The purchase costs of the equipment are given below. The prices calculated with the method of Dr. Ž. Olujić are compared with the Webci prices (guilders are converted to US dollars, 1 dollar equals 1.7 guilders).

Table 9.12. The purchase cost of the furnace (reformer).

Equipment code	Number	PCE according to Olujić (10 ³ x \$)	PCE according to Webci (10 ³ x \$)
H-2101	1	857.5	857.5
TOTAL PCE		857.5	857.5

Webci has no prices for furnaces, so the same prices as Olujić are taken.

The price for the catalyst in the reformer is 30 guilders per litre or 18 \$ per litre. The total price is 1560 x 18 (\$) = 28 x 10³ US \$.

Table 9.13. The purchase costs of the catalyst.

TOTAL PCE (10 ³ x \$)	28
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Table 9.14. The purchase costs of the pressure vessels.

Equipment code	Number	PCE according to Olujić (10 ³ x \$)	PCE according to Webci (10 ³ x \$)
V-2101	1	5.7	11
V-2102	2	8.1	12
V-2103	1	5.2	13
V-2104	1	19.3	14
V-2107	1	12.7	15
V-2108	1	11.0	15
V-2109	1	8.1	14
V-2110	1	15.5	10
V-2111	1	2.0	10
V-2112	1	8.1	12
V-2113	2	6.9	14
V-2114	2	5.2	14
V-2115	1	5.2	11
V-2116	1	3.0	10
V-2117	3	3.3	11
V-2119	1	1.8	10
TOTAL PCE		148.3	253

Table 9.15. The purchase costs of the columns without trays and packings.

Equipment code	Number	PCE according to Olujić (10 ³ x \$)	PCE according to Webci (10 ³ x \$)
V-2105	1	24.1	83
V-2106	1	57.6	131
V-2118	3	12.1	38
T-21	1	89.0	68
T-22	1	23.0	54
T-23	1	14.1	37
T-24	1	13.1	37
TOTAL PCE		257.2	524

Table 9.16. The purchase costs of the trays of the columns.

Equipment code	Number	PCE according to Olujić (10 ³ x \$)	PCE according to Webci (10 ³ x \$)
V-2105	1	0.5	0.4
V-2106	1	3.3	2.1
V-2118	3	0.7	-
T-21	1	5.9	2.2
T-22	1	13.4	3.0
T-23	1	8.9	1.4
T-24	1	8.9	1.4
TOTAL PCE		43.0	10.5

Table 9.17. The purchase costs of the packings (Pall rings and distributors) of the columns.

Equipment code	Number	PCE according to Olujić (10 ³ x \$)	PCE according to Webci (10 ³ x \$)
V-2105	1	19.1	27.0
V-2106	1	19.9	25.7
V-2118	3	-	-
TOTAL PCE		39.0	52.7

The purchase cost of MEA in the columns V-2105 and V-2106 is calculated by subtracting the packing volumes (11.7 m^3) from the column volumes (22.31 m^3) and adding 10% for the MEA in the pipes. The price of MEA is $f1.94/\text{kg}$ or $1.14 \text{ \$/kg}$, so the total purchase cost of MEA is $13 \times 10^3 \text{ \$}$.

Table 9.18. The purchase cost of MEA.

TOTAL PCE ($10^3 \times \text{\$}$)	13
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Table 9.19. The purchase costs of the heat exchangers.

Equipment code	Number	PCE according to Olujić ($10^3 \times \text{\$}$)	PCE according to Webci ($10^3 \times \text{\$}$)
H-2102	1	3.4	10
H-2103	1	15.2	27
H-2104	1	7.0	20
H-2105	2	5.4	12
H-2106	1	8.9	16
H-2107	1	13.3	18
*H-2108	1	14.1	32
*H-2109A	1	12.3	25
*H-2109B	1	25.0	30
*H-2110	1	59.0	78
*H-2111	1	36.2	45
*H-2112	1	18.4	30
H-2113	2	6.6	15
*H-2114	2	5.1	14
*H-2115	3	8.5	18
*H-2116	1	4.6	13
*H-2117	1	4.5	13
H-2118	2	7.1	16
H-2119	2	4.1	12
*H-2120	1	3.7	14
E-21	1	14.5	21
E-22	1	42.7	25
E-23L	1	126.7	53
E-23U	1	184.9	88
E-24	1	10.2	16
E-25	1	10.2	16
E-27	1	10.2	16
TOTAL PCE		706.8	797

The heat exchangers denoted with * are U-tubes, and their prices are 10-15% more expensive than the prices for heat exchangers given by Webci. Therefore 15% is taken into account in the prices of the U-tubes.

Table 9.20. The purchase costs of the compressors and ventilators.

Equipment code	Number	PCE according to Olujić (10 ³ x \$)	PCE according to Webci (10 ³ x \$)
K-2101	2	12	12
K-2102	2	7	7
K-2103	4	102.6	94
K-2104	3	151.9	126
K-2105	4	72.6	62
TOTAL PCE		1194.6	1041

K-2101 and K-2102 are ventilators. Olujić has no prices for these equipment, therefore the same prices as Webci are taken.

Table 9.21. The purchase costs of the drivers.

Equipment code	Number	PCE according to Olujić (10 ³ x \$)	PCE according to Webci (10 ³ x \$)
K-2103	4	19.2	31
K-2104	3	26.7	35
K-2105	4	14.2	15
TOTAL PCE		213.7	290

The purchase costs, the fixed capital and the investment cost of the whole plant (design) are given in Table 9.22.

Table 9.22. The investment costs of the design plant.

10 ³ x US \$	according to Olujić	according to Webci
Total PCE	3502	3867
Total PPC $PPC=f_a \cdot PCE$	11380	12568
Fixed Capital $FC=f_b \cdot PPC$	16501	18223
Working Capital (10%)	1650	1822
Investment cost	18151	20046

9.3.2. The new process (only extra CO₂)

For the new process a CO₂ storage tank is needed with a volume of 15 m³ and several heat exchangers have to be expanded by adding some new heatexchangers. The purchase costs can be found in the Tables 9.23 and 9.24.

Table 9.23. The purchase cost of a storage tank.

Equipment code	PCE according to Olujić (10 ³ x \$)	PCE according to Webci (10 ³ x \$)
CO ₂ storage tank	11.7	29

Table 9.24. The heat exchangers.

Equipment code	Number	Extra transfer area (m ²)	PCE according to Olujić (10 ³ x \$)	PCE according to Webci (10 ³ x \$)
H2108 extra	1	7.63	4.1	13.5
H2109A extra	1	6.21	3.9	12.9
H2109B extra	1	6.21	4.4	18.3
H2110 extra	1	14.2	10.1	25.0
H2111 extra	1	8.0	5.7	19.6
H2112 extra	1	5.0	3.3	12.2
H2113 extra	2	3.0	2.3	9.7
H2114 extra	1	2.0	1.9	10.8
H2115 extra	1	2.0	2.2	10.8
H2115 extra	2	3.5	3.0	11.5
H2116 extra	1	1.5	1.8	10.8
H2117 extra	1	1.5	1.5	10.8
H2118 extra	2	7.0	4.0	11.5
H2119 extra	1	4.0	2.8	10.0
H2120 extra	1	1.5	1.4	10.8

The other equipment remains the same. The investment cost is shown in Table 9.25.

Table 9.25. The investment cost.

$10^3 \times \text{US } \$$	according to Olujić	according to Webci
Extra PCE	73	260
Extra PPC ($f_a = 2.55$)	187	663
Extra FC ($f_b = 1.45$)	271	961
Total fixed capital	16772	19184
Investment cost	18449	21102

9.3.3. The new process (using extra CO_2 and a pre-reformer)

For the new process a CO_2 storage tank is needed with a volume of 15 m^3 and a pressure vessel with catalyst. The total amount of catalyst needed per year remains the same, because due to the pre-reformer the life time of the catalyst in the reformer increases. The purchase costs can be found in the Tables 9.26 and 9.27.

Table 9.26. The purchase cost of a storage tank.

Equipment code	PCE according to Olujić ($10^3 \times \$$)	PCE according to Webci ($10^3 \times \$$)
CO_2 storage tank	11.7	29

Table 9.27. The purchase cost of a pressure vessel

Equipment code	PCE according to Olujić ($10^3 \times \$$)	PCE according to Webci ($10^3 \times \$$)
pre-reformer vessel	12.0	20

The other equipment remains the same. The investment cost is shown in Table 9.28.

Table 9.28 The investment cost.

10 ³ x US \$	according to Olujić	according to Webci
Extra PCE	24	49
Extra PPC ($f_a = 2.55$)	61	125
Extra FC ($f_b = 1.45$)	88	181
Total fixed capital	16589	18404
Investment cost	18640	20244

9.4. Calculation of the operating costs

9.4.1 Design process (KTI)

The annual operating costs are calculated by using Table 9.11 and the investment cost determined by Olujić. The operating time is 95% of one year, 345 days/year or 8280 hours/year.

A. The variable costs:

1.Raw materials

- Natural gas 0.19 Dfl/Nm³ or 0.00277 \$/mol
53.1 kmol/h x 8280 h/y x 0.00277 \$/mol 1218 x 10³ \$
- Boiler feed water 1.15 Dfl/m³ or 0.68 \$/m³
6.466 m³/h x 8280 h/y x 0.68 \$/m³ 36 x 10³ \$

2.Miscellaneous materials

165 x 10³ \$

3.Utilities

- Electricity for compressors
2260 kW x 8280 h/y x 0.064 Dfl/kWh (or 0.038 \$/kWh) 704 x 10³ \$
- Degradation costs of the catalyst
1/5 y⁻¹ x 28,000 \$ 6 x 10³ \$
- Degradation costs of MEA
0.4 y⁻¹ x 11670 kg x 1.14 \$/kg 5 x 10³ \$

4.Shipping and packaging - not applicable

Subtotal A

2134 x 10³ \$

B. The fixed costs:	
5.Maintenance	1650 x 10 ³ \$
6.Operating labour	
Four manpower are needed to operate the plant continuously. One continuous manpower costs f350,000 per year or 206,000 \$ per year.	
	824 x 10 ³ \$
7.Supervision	165 x 10 ³ \$
8.Plant overheads	412 x 10 ³ \$
9.Capital charges	2475 x 10 ³ \$
10.Insurance	165 x 10 ³ \$
11.Rates	330 x 10 ³ \$
12.Royalties	165 x 10 ³ \$

Subtotal B	6186 x 10 ³ \$
Direct production costs A + B	8320 x 10 ³ \$
C. Share of the Company's general operating expenses (25% of the direct production costs)	
13.Sales expense	
14.General overheads	
15.Research and development	

Subtotal C	2080 x 10 ³ \$
	=====
Annual operating costs (A + B + C)	10400 x 10 ³ \$
Production cost (1221.4 kg CO/h)	1.03 \$/kg.

9.4.2. The new process (using only extra CO₂)

For the variable costs just the prices for raw materials and miscellaneous change:

1. Raw materials
 - Natural gas: $50.1 \text{ kmol/h} \times 8280 \text{ h/y} \times 0.00277 \text{ $/mol} = 1149 \times 10^3 \text{ $}$
 - Carbondioxyde: $14.2 \text{ kmol/h} \times 44.0 \text{ kg/kmol} \times 8280 \text{ h/y} \times 0.185/1.7 \text{ $/kg} = 584 \times 10^3 \text{ $}$
 2. Miscellaneous materials = $168 \times 10^3 \text{ $}$
- Subtotal A = $2652 \times 10^3 \text{ $}$

For the fixed costs only the costs for operating labour, supervision and plant overheads don't change. The other costs will be:

5. Maintenance = $1677 \times 10^3 \text{ $}$
 9. Capital charges = $2516 \times 10^3 \text{ $}$
 10. Insurance = $168 \times 10^3 \text{ $}$
 11. Rates = $335 \times 10^3 \text{ $}$
 12. Royalties = $168 \times 10^3 \text{ $}$
- Subtotal B = $6265 \times 10^3 \text{ $}$

The direct production costs (A + B) are 8917×10^3 \$, therefore the subtotal of C (25% of the direct production costs) is 2229×10^3 \$.

The annual operating costs for the new process are 11146×10^3 \$ and the production cost for CO will be $(55.402 \text{ kmol CO/h} \times 28 \text{ kg/kmol} \times 8280 \text{ h/y})^{-1} \times 11146 \times 10^3 \text{ \$/y} = 0.87 \text{ \$/kg}$.

9.4.3. The new process (using extra CO₂ and a pre-reformer)

For the variable costs just the prices for raw materials and miscellaneous change:

1. Raw materials

- Natural gas: $50.2 \text{ kmol/h} \times 8280 \text{ h/y} \times 0.00277 \text{ \$/mol} = 1151 \times 10^3$ \$

- Carbondioxyde: $13.9 \text{ kmol/h} \times 44.0 \text{ kg/kmol} \times 8280 \text{ h/y} \times 0.185/1.7 \text{ \$/kg} = 550 \times 10^3$ \$

2. Miscellaneous materials = 168×10^3 \$

Subtotal A = 2620×10^3 \$

For the fixed costs only the costs for operating labour, supervision and plant overheads don't change. The other costs will be:

5. Maintenance = 1660×10^3 \$

9. Capital charges = 2488×10^3 \$

10. Insurance = 166×10^3 \$

11. Rates = 332×10^3 \$

12. Royalties = 166×10^3 \$

Subtotal B = 6213×10^3 \$

The direct production costs (A + B) are 8833×10^3 \$, therefore the subtotal of C (25% of the direct production costs) is 2208×10^3 \$.

The annual operating costs for the new process are 11041×10^3 \$ and the production cost for CO will be $(53.809 \text{ kmol CO/h} \times 28 \text{ kg/kmol} \times 8280 \text{ h/y})^{-1} \times 11041 \times 10^3 \text{ \$/y} = 0.89 \text{ \$/kg}$.

LITERATURE

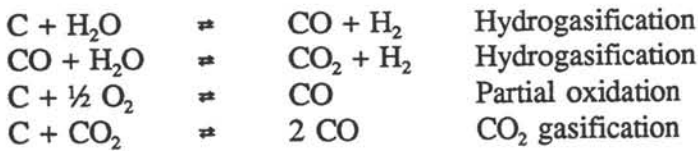
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10 ALTERNATIVE METHODS FOR CO-PRODUCTION

10.1 Possibilities for CO-production

10.1.1 Coal Gasification

Another process to make CO is coal gasification. In this process solid coal is reformed into gaseous CO and sometimes, depending on the feed, H₂ which has to be separated later in the process. Coal is gasified in either by partial combustion or by CO₂-hydro-gasification. Very often a process is a combination of both modes, depending on the product desired. The main reactions that take place in a coal gasifier are :



The product needed is CO, thus when you look at the reactions it is obvious that reactions with O₂ and CO₂ are preferred. In practice gasification of coke can be carried out in a fixed bed slagging gasifier which is continuously operated at slightly above atmospheric pressure. A possible scheme is illustrated in fig.10.1.

The feed needed for this process contains O₂, CO₂, sized coal and flux (typically limestone). Flux is added to the coal in order to obtain favorable ash melting temperatures and slag viscosities. The solid feed is fed to the reactor via a fully automatic controlled system consisting of conveyors, bunkers, chutes and locks. The gaseous feed (CO₂ and O₂) is injected in the reactor's bottom section via a mixing tube. The generated gas ascends countercurrently to the coal bed and the crude gas leaves the reactor at about 870 °C at the top with a CO concentration of 90-92 % depending on feedstock qualities. Some coke fines are also carried over from the reactor along with the crude gas stream. These fines are separated in a cyclone and recycled to the reactor. After the cyclone the crude gas is passed through a scrubber where it is water saturated and cooled. Here the gas is not only cooled but also residual traces of dust are being removed. Finally, the crude gas is cooled to 35 °C. At the bottom liquefied coke ash (slag) is removed batchwise with a automatic deslagging system. Of course the gasproduction continuous during deslagging.

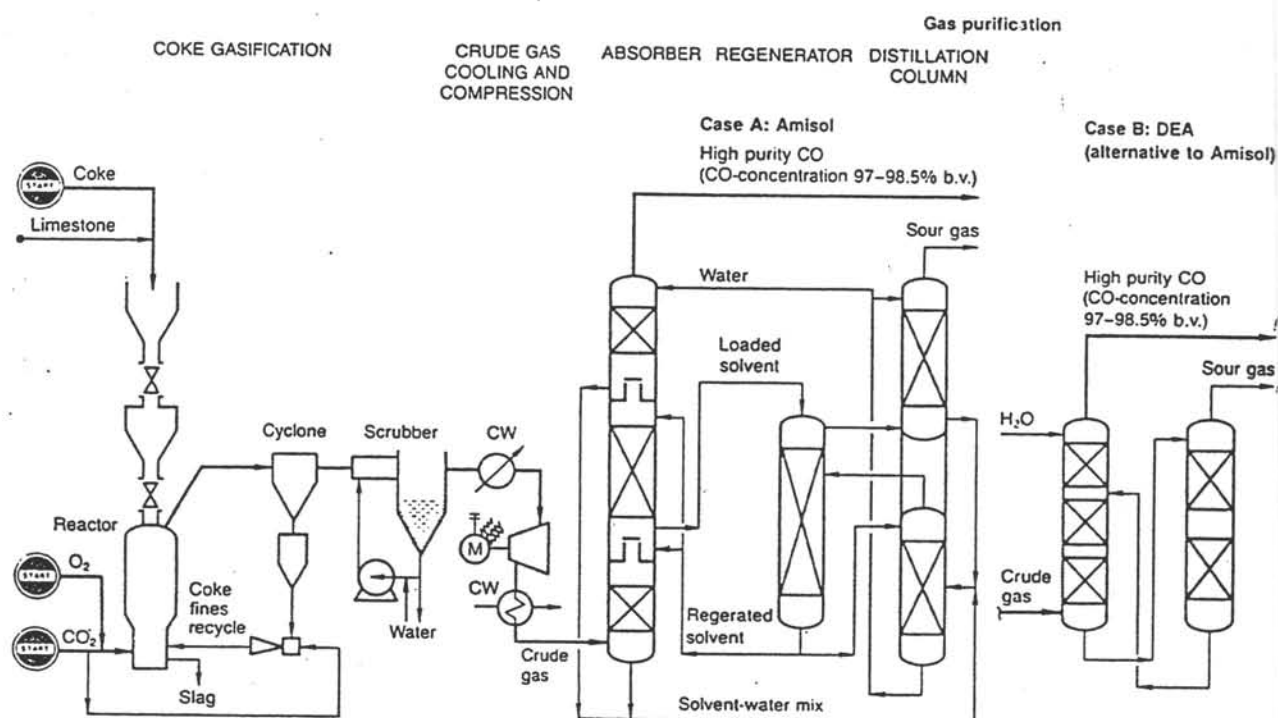


fig.10.1 : scheme for coke gasification and gas purification units.

After the reactor the crude gas contains about 90-92 % CO. To remove the CO₂ there are some different methods. In [1] there are two options mentioned, the Amisol process and the DEA process. The Amisol process is more expensive but applied if there are stringent requirements with respect to sulfur compounds. The Amisol is capable of removing all sulfur compounds down to a level of less than 0.1 ppm total sulfur.

The Amisol process works with a mixture of two solvents, consisting of diethylamine (chemical solvent) and methanol (physical solvent). The DEA process on the other hand has only a chemical solvent diethanolamine in water. This is a cheaper process which can be selected when a sulfur concentration of approx. 500 ppm is acceptable.

After the CO₂ removal a CO concentration of 97 to 98.5% by volume is available. This CO makes a good feedstock for most syntheses. This process is commercially proven by Lurgi GmbH, Frankfurt, Germany. Both investment and operating costs are rather competitive in comparison to processes based on natural gas. Investment costs come out rather low because:

- * After the reactor the CO concentration is much higher in comparison to the CO concentration for processes based on natural gas. 90 to 92 % for the coke process versus 28 % for the natural gas process. This makes the gas volume to be handled much lower; hence smaller equipment size will do latter in the process.
- * Normally no gas separation unit is required as a CO concentration of 98+ % can be achieved just by removal of CO₂ and sulfur compounds.

Also the operating costs are rather low because of:

- * Low specific consumption figures for coke, O₂ and CO₂ due to the high selectivity for CO of the gasification process.
- * Low specific utility consumption rates like electricity and cooling water. This is also based on the high selectivity for CO, plus the fact that a gas separation unit is not normally required.

The costs for producing one ton of CO are given below in table 10.1 and are for a CO plant with a capacity of minimal 6.5 ton/hour. When in a natural gas plant the hydrogen only can be used as fuel, the gasification process is 70 to 100 % below the cost of CO production in a natural gas plant. lit.[10.1]. The reason of not seriously being a candidate for an alternative production process is the minimum capacity for being competitive. It is only economical attractive with a production of at least 6.5 ton/hour. The current capacity for the two GEP plants is about 2.2 ton/hour.

Of course there are other reactor types which are not of the moving bed type but are entrained flow reactors. They operate at higher temperatures with greater mass flow; hence the entrained flow reactors are only economical at higher capacity.

Table 10.1: Cost in DM for the production of one ton CO (1986).

coke	0.36 t	DM 300 / t	108
oxygen	200 Nm ³	DM 0.16 / t	32
CO ₂	250 Nm ³	DM 0.14 / t	35
limestone	0.02 t	DM 100 / t	2
electrical power	75 kWh	DM 0.12 / t	9
cooling water	18 m ³	DM 0.1 / t	2
labor	6 pers.	DM 60.000/year	7
capital investment			65
total production cost for one ton CO			260

10.1.2 The CO₂ reformer

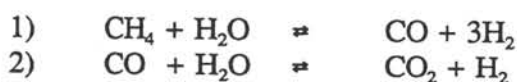
For the alternative CO production plant a reformer is used which has the same principle as the reformer of the existing plant. The reformer consists of the same compositions as the existing reformer: vertical tubes with catalyst, heaters in the wall but the reformer has a different feed composition due to different feed and different recycle streams.

The total feed to the reformer (feed + reflux) is given in table 10.2.

Table 10.2 : feed composition alternative reformer

reformer feed	kmol/h
CH ₄	40.75
CO ₂	58.33
N ₂	13.6
H ₂ O	113.0

In the reformer the following reactions describe the CO-conversion:



with:

$$K_1 = \exp \left[-\frac{27464}{T} + 30.707 \right] \quad (10.1)$$

$$K_2 = \exp \left[\frac{4084}{t} - 3.765 \right] \quad (\text{only for } T > 773 \text{ K}) \quad (10.2)$$

The calculations for this equilibrium reactions are calculated with mercury at 860 K. The result of these calculations is the output of the reformer (table 10.3).

Table 10.3: alternative reformer output

reformer output	kmol/h
CH ₄	1.942
CO ₂	42.15
N ₂	13.6
H ₂ O	90.38
CO	55.00
H ₂	100.2

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To prevent coke forming in the reformer, according to group 1, a equation which shows a coke forming value is mentioned:

$$\frac{1/2CO_2 + H_2O}{CH_4} > 3.0 \quad (10.3)$$

For this situation the coke forming value is 3.5, which is good enough.

10.2 Membranes for H₂ separation

Below are some possible H₂ membrane applications:

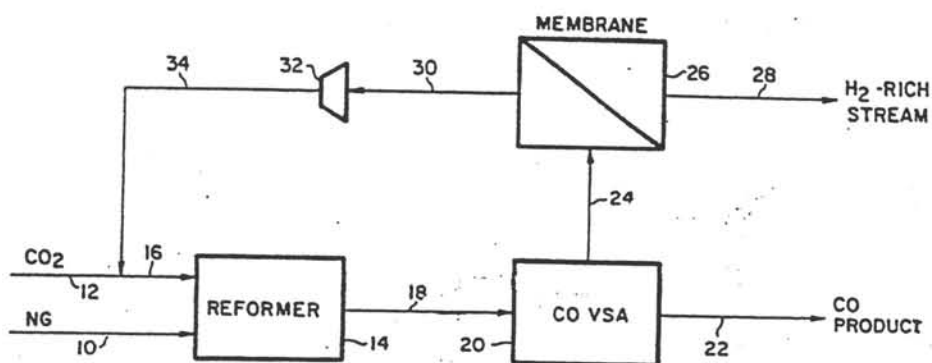


fig. 10.4: H₂ membrane system: first CO recovery

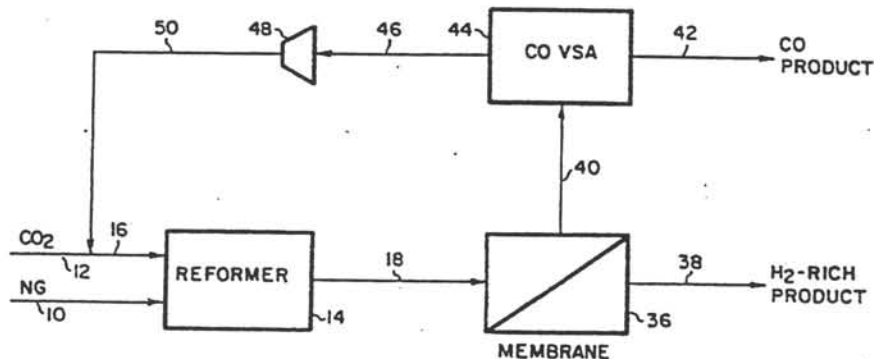


fig. 10.5: H₂ membrane system: first H₂ recovery

The difference between the alternative processes above in fig. 10.4 and fig. 10.5 is that the hydrogen and carbon monoxide separation stages are reversed. When the H₂ membrane separation comes before the CO removal, the membrane stage sees a heightened partial pressure of carbon monoxide and therefore more carbon monoxide is lost through the membrane as permeate in the hydrogen-rich stream. Thus recovery of carbon monoxide in fig. 10.5 is much lower than the recovery of CO in fig. 10.4. (92 % vs 99 %). That is the reason why in the next alternatives always first CO recovery takes place and later the H₂ recovery.

In this membrane process the carbon monoxide lean gas mixture enters a H₂-rejection membrane stage in which H₂ permeates through a membrane to the low pressure side of the membrane as permeate, while the remaining gas mixture passes over the surface of the membrane as the reject stream. A preferred membrane that is selective for hydrogen over the remaining constituents of the gas mixture comprising CO, CO₂, CH₄, N₂, Ar is polymeric cellulose acetate or polysulfone. It should be noted that, as an alternative, the feed stream to the membrane can be boosted in a compressor to aid the H₂ separation in the membrane.

The advantage of using a H₂ membrane separation unit is the high product purity (95-99%) and the high recovery (85%). The right feed condition necessary for a high recovery is a high feed concentration. The H₂ concentration after the CO separation is about 69 %, which is not extremely high.

The disadvantages of using a H₂ membrane unit is the large pressure drop on the product side. Also a H₂ membrane has a poor CO₂ and H₂O separation selectivity. Because the H₂ rich stream consists (except H₂) mainly of CO₂, a membrane system is not recommendable.

A recent (1990) study lit.[10.4] of separating H₂ from coal-gas has indicated that for 99.5 % purity membrane, the separation efficiency is only 40 %. This is the result of bad feed conditions.

10.3 PSA for CO - H₂ separation

The design of the H₂ PSA system

Separation by an adsorptive process is based on the selective adsorption of one or more components of a gas mixture on the surface of a microporous solid. When a gaseous mixture is exposed to an adsorbent for a sufficient time, the gas phase becomes richer in the less selective adsorbent. Desorption can be achieved either by increasing the temperature or by reducing the adsorbate pressure. Thus, the adsorptive separation process consists of a cyclic sequence of adsorption and desorption steps. When desorption is achieved by decreasing the partial pressure of the selective component, the process is called pressure swing adsorption (PSA). A modern pressure swing adsorption system is made up of four or five pressure vessels containing an appropriate adsorbent (adsorbent beds) with interconnecting piping and valving with an automatic control device as shown in fig.10.6

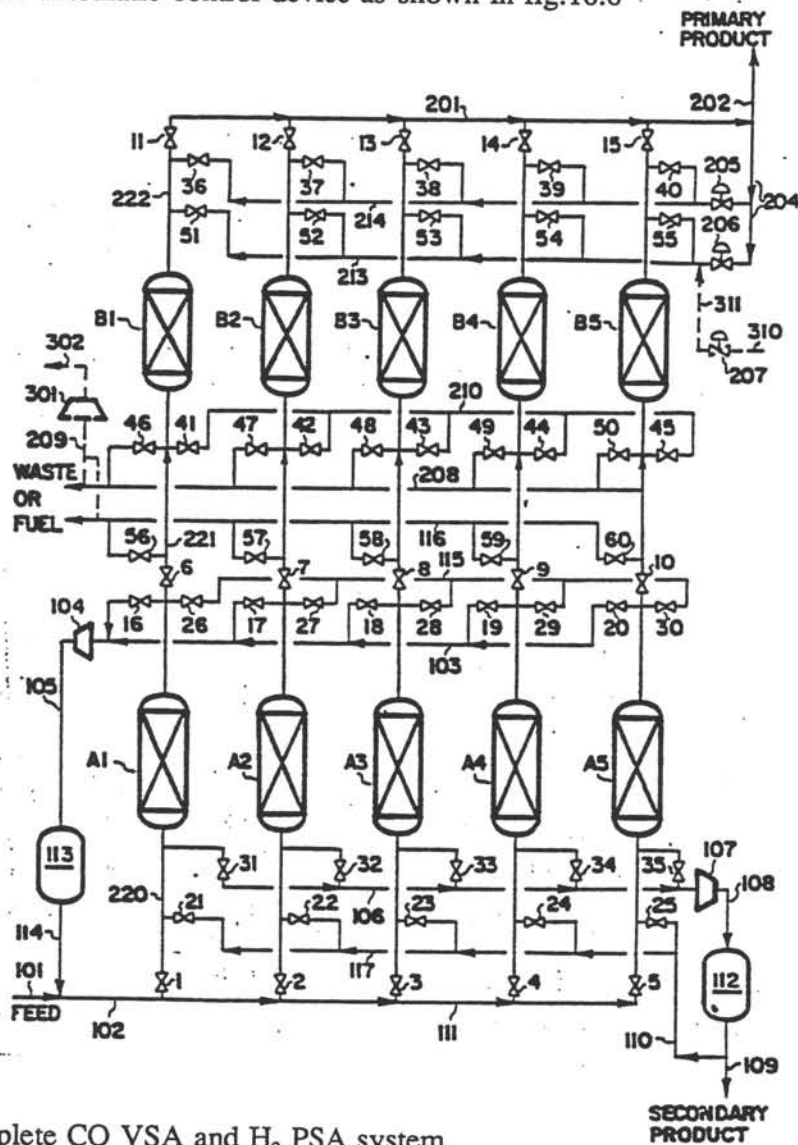


fig 10.6 : complete CO VSA and H₂ PSA system.

In the A-beds the H₂ adsorbent is placed and in the B-beds the CO adsorbent is placed. Hence the primary product is Carbon Monoxide and the secondary product is hydrogen.

The adsorbent - in this case of the type 5A Zeolite sieves - is selected on the basis of the application requirements. This zeolite has a greater affinity for carbon-dioxide, methane and carbon monoxide than for hydrogen.

With the equation according to White lit.[10.5], the fluidization limit for packed beds is determined, and with that the minimal column diameter. The maximum interstitial velocity according to White:

$$v_s(\text{max}) = -\frac{\mu E_1}{2\rho_0 E_2} + \sqrt{\left(\frac{\mu E_1}{2\rho_0 E_2}\right)^2 + \frac{\rho_a g}{\rho_0 E_2}} \quad (10.4)$$

with:

μ = viscosity in Pa s

ρ_0 = density of the gas in kg/m³

ρ_a = density of the adsorbent (zeolite 5A) in kg/m³

and:

$$E_1 = 1471 \frac{(1-\epsilon)^2}{\epsilon^3 D_p^2 g} \quad E_2 = 17.16 \frac{(1-\epsilon)}{\epsilon^2 D_p g} \quad (10.5)$$

$$(10.6)$$

with:

ϵ = total void factor

D_p = particle diameter

From the maximum velocity, the minimal column diameter is determined with the minimal area :

$$A = \frac{\omega_0}{\rho_0 v_s} \quad (10.7)$$

with:

ω_0 = mass flow in kg/s

An estimation of the column height can be determined from the amount adsorbents necessary for the adsorption of the residual components.

$$L = \frac{1}{A \rho_a n_k} * \sum \frac{\Phi_{mol,i} t_c}{(Q_{a,i})_{P_{ads}} - (Q_{a,i})_{P_{des}}} \quad (10.8)$$

with:

- n_k = number of beds
- t_c = cycle time in sec.
- $\Phi_{mol,i}$ = flow of component i in mol/s
- $Q_{a,i}$ = amount adsorbed component i in mol/kg
- P_{ads} = adsorption pressure
- P_{des} = desorption pressure

The pressure drop over the bed is calculated with the modified Sabri-Ergun equation lit.[10.6]:

$$\frac{\Delta P}{L} = E1 \mu v_s + E2 \rho_0 v_s^2 \quad (10.9)$$

The results from these equations can be found in the appendix 10.(page A10-1)

The equipment costs for a H₂ VSA system, the so called Purchased Cost Equipment (PCE), are:

=====> fl. 1.12 * 10⁶

The total investment costs are determent by the factoring method. The so called Fixed Cost (FC) are:

=====> fl. 5.26 * 10⁶

The calculations of these costs can be found in appendix 10 (page A10-3)

The design of the CO VSA adsorption system

This Vacuum Swing Adsorption system is located before the H₂ PSA system and after the dryers to remove the water. The adsorbent in this case is an adsorbent consisting of a silica and/or alumina carrier with an amount of cupric salt in the range of 1 to 5 mmols/g and a reducing agent to convert cupric ion into cuprous ion in the range of 0.01 to 0.3 g/g carrier. The adsorbent has a adsorption capacity of 21.7 cc/cc adsorbent at atmospheric pressure and releases 9.9 cc/cc adsorbent at 0.04 bar lit.[10.8], 9.9 cc / cc adsorbent is the same as 3.6 mol CO/gr. Because CO is the only component that has to be adsorbed, the beds are smaller than in the H₂-PSA system.

With the equation of White (10.4) the fluidization limit for packed beds is determined and with those results the minimum column diameter.

From the maximum velocity, the minimal column diameter is calculated, using equation 10.4. An estimation of the column height is calculated with equation 10.8 with only CO is adsorbed.

The result of the design of the CO VSA system is in appendix 10.(page A10-2)

The equipment costs for a CO VSA system, the so called Purchased Cost Equipment (PCE), are:

$$\text{=====}> \text{ fl. } 0.877 * 10^6$$

The total investment costs are determined by the factoring method. The so called Fixed Cost (FC) are:

$$\text{=====}> \text{ fl. } 4.13 * 10^6$$

Of course the costs of a combined CO-VSA and H₂-PSA system are lower because piping and control system are combined. The combined equipment costs (PCE) are:

$$\text{=====}> \text{ fl. } 1.83 * 10^3$$

The total investment costs for a combined CO-VSA and H₂-PSA system are:

$$\text{=====}> \text{ fl. } 8.61 * 10^3$$

The calculation of these cost can be found in appendix 10 (page A10-3/4)

10.4 COSORB® for CO-raw gas separation

10.4.1 The COSORB® Process

Absorption of carbon monoxide is important in the processes for purification of CO as well as in the removal of CO from various gas streams. The COSORB® Process for the recovery and purification of carbon monoxide from synthesis gases and industrial waste gas streams was developed during the late 1960's and early 1970's. The process utilizes a proprietary COSORB® Solvent composed of a pi bond complexing compound, copper aluminum tetrachloride, dissolved in an aromatic solvent base. The first commercial COSORB® plant was successfully started up at the Dow Chemical Company Plant in Freeport, Texas, U.S.A. in early 1976. Since this early success, the COSORB® Process has enjoyed wide acceptance in the chemical process industry. The COSORB® Process has been applied to numerous diverse feed gases containing carbon monoxide and other inert gases to produce a high purity carbon monoxide product.

The COSORB® Process is based on the selective complexation of carbon monoxide in a liquid solvent (toluene). The solvent has an aromatic base with an active component referred to as CuAlCl_4 . The high selectivity of this solvent for carbon monoxide at ambient temperature and the ease of decomplexation at a higher temperature are the basis for the simplicity of the process. The solvent can effectively remove carbon monoxide from gaseous streams at any practical operating pressure. The COSORB® process is essentially a simple absorption/desorption process for carbon monoxide, similar to a carbon dioxide removal process. Its development followed a rather conventional pattern. Extensive vapor liquid equilibrium data were collected in the laboratory. This was followed by bench scale laboratory evaluations of the process. These data were then utilized as the basis for the design and simulation program (appendix 10) of the COSORB® unit.

The basic design of a COSORB® unit is illustrated in Fig. 10.7 and consists of four sections: 1) Feed gas preparation, 2) the COSORB® solvent-CO complexer and COSORB® solvent-CO decomplexer circuit, 3) aromatic recovery from the product gas streams, and 4) the compression and power recovery of the product streams.

The feed gas preparation section consists of the process steps required to remove the irreversibly reactive compounds from the feed gas prior to its contact with the COSORB® solvent. Water is usually removed by a combination of refrigeration and molecular sieves. After conventional treatment for the removal of sulfur compounds, the sulfur compounds are reduced to acceptable levels by treatment with activated carbon. In those cases where hydrogenation of the feed gas is required, zinc oxide beds may be used to reduce sulfur compounds to an essentially "zero" level upstream of the hydrogenation catalyst.

The complexer and decomplexer section of the COSORB® process consists two major columns, which are either packed or trayed columns, with internals identical to standard absorption and stripping columns. The feed gas is countercurrently contacted with the COSORB® solvent in the complexer, and the CO is released from the COSORB® solvent in the decomplexer.

The other common components in a carbon monoxide-rich stream, carbon dioxide, methane, hydrogen, nitrogen and oxygen are physically absorbed.

Since there is a low physical absorption of these inert gases, only a fraction of the inert gases in the feed gas will be absorbed in the COSORB[®] solvent and then exit with the solvent from the complexor. These physically dissolved gases will be produced with the decomplexed CO product. However, these impurities may be removed from the COSORB[®] solvent by various combinations of flashing, stripping and complexation in an inert-removal column upstream of CO production in the decomplexor. The design requirements of this third column are determined by the feed compositions and product purity requirements.

The carbon monoxide-rich solvent flows from the bottom of the absorber to a flash unit and then to a heat exchanger and is heated against recycling lean solvent, and where it gives up a large fraction of the dissolved carbon monoxide to the gas phase. The inerts are preflashed, and the final carbon monoxide product is taken overhead of the stripper. The aromatic base of the solvent is generally the stripping medium. The hot lean solvent flows through the heat exchanger, where it is cooled and then recycled to the absorber. The complexor, decomplexor and optional inerts-removal column overhead streams require further treatment to reduce losses of the volatile aromatic base of the COSORB[®] solvent. Aromatic recovery in the complexor and decomplexor overhead streams is accomplished by a combination of compression, refrigeration and treatment with activated carbon. The inerts column overhead stream is small and is generally sufficient for economical recovery of the aromatics. A turbine compressor compresses the carbon monoxide by making use of pressure from the feed gas to the absorber. With this approach, the stripper could be operated, for example at five psig. Carbon monoxide product of 99% is easily attained in this system. The loss of the aromatic base of the solvent is reduced by use of recovery facilities on the carbon monoxide product stream and the hydrogen-rich stream.

The COSORB[®] plant can be built with carbon steel. The gas feed must be substantially free of water, hydrogen sulfide, and sulfur dioxide, as these compounds all irreversibly react to a significant extent with the solvent and produce traces of hydrogen chloride in the product streams. In those instances where downstream processes are sensitive to parts-per-million levels of hydrogen chloride, the complexor and decomplexor overheads may require treatment to remove hydrogen chloride to below detectable limits.

Packed towers are more than adequate to allow the separation to produce high-purity carbon monoxide at multimillion-pound-per-day rates. The only utilities necessary are cooling water, low pressure steam, and power sufficient to pump the solvent from the stripper to the absorber.

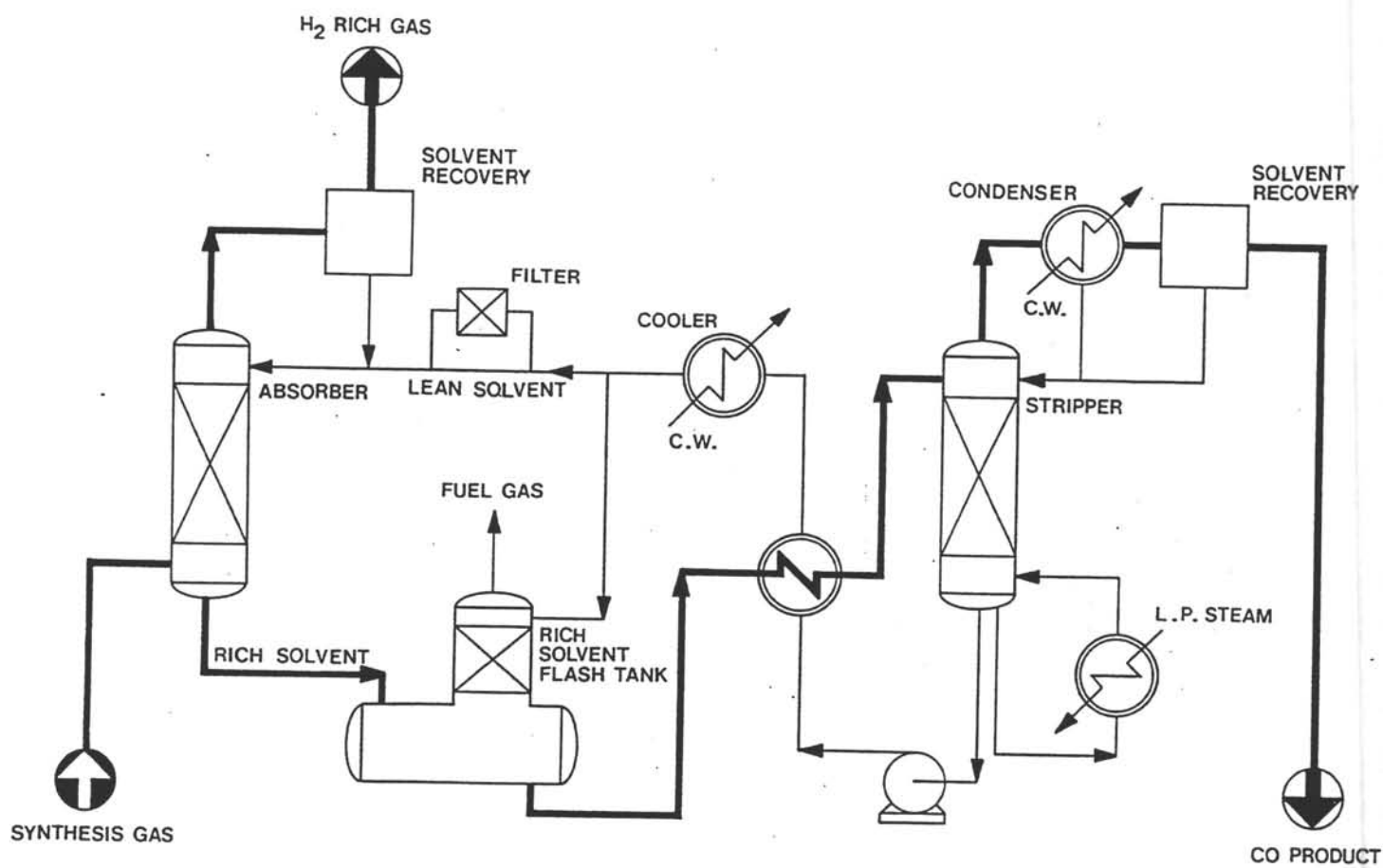


Figure 10.7 Cosorb process for recovering carbon monoxide from industrial gases

10.4.2 Simulation of the COSORB[®] unit (Non-isothermal gas absorption)

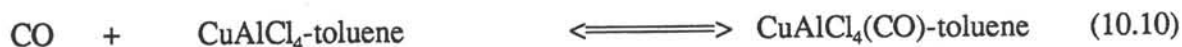
10.4.2.1 Introduction

Gas Absorption with chemical reaction induces the absorption rate which is expressed by means of "enhancement factor" corresponding to the liquid film resistance. The enhancement factor differs by different systems and different operation conditions. The molecular interaction energy of chemical absorption-complexing is often much greater than that of *Van der Waals* force in physical absorption, but still less than covalent energy. And since complexing absorption is reversible, it is easy for desorption (regeneration), therefore it is a highly selective, widely useful and practical separation process which involves large heat effects such as heat of reaction, heat of solution, sensible heat transfer between liquid and gas phase. In this case, the computation and design of these absorbers have to consider of chemical equilibrium, phase equilibrium, mass transfer and heat transfer.

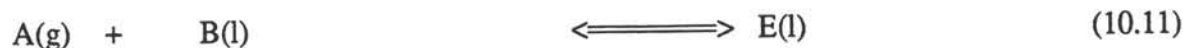
10.4.2.2 Fundamental Theory

COSORB[®] Solvent utilizes the pi complex bonding between the aromatic solvent, e.g., toluene, to the metal salt adduct of cuprous chloride and aluminum chloride (CuAlCl_4) to achieve very high concentrations of the solid CuAlCl_4 in the aromatic liquid. The species in COSORB[®] Solution are $\text{C}_7\text{H}_8:\text{CuAlCl}_4$ + approximately 1.5 C_7H_8 ; Thus an overall total molar ratio toluene/copper = 2.5. The pi complexed copper is capable of further complexing with electron donating ligands, such as carbon monoxide, which can donate pi electrons to form additional pi complex bonds to the copper(I) metal. Of key importance to the COSORB[®] process is the fact that the additional pi bond can be made at ambient temperature and low CO partial pressure and be broken simply by elevating the temperature. The mixture of $\text{C}_7\text{H}_8:\text{CuAlCl}_4$ plus 1.5 free C_7H_8 is simply convenient vehicle to transport the copper(I) metal and its inherent complexing nature from the CO Absorber to the CO Stripper and back again. COSORB[®] Solvent, as defined, has physical properties much like the toluene portion of the solvent and thus exhibits reasonable heat transfer and good mass transfer characteristics.

The absorption of carbon monoxide in $\text{CuCl}-\text{AlCl}_3$ -toluene complex solution can be described as a stoichiometric reaction as follows :



This reversible reaction can be represented by the following scheme :



The theory of gas absorption with fast reversible reaction has been described by Onda (1970) lit. [10.10] and the relevant equations applicable to the present case are described in detail by Gholap and Chaudhari (1988) lit. [10.11]. The ratio of chemical absorption rate and that of physical absorption namely "enhancement factor" (E) expresses the effect of chemical reaction to the absorption, and it differs by different systems, operating and boundary conditions. The equations for enhancement regime and the following condition must be satisfied :

$$3 \leq E \leq E_{\infty} \quad (10.12)$$

where E_{∞} is the enhancement factor for instantaneous reversible reaction and Danckwerts and his coworkers (1970) lit. [10.12] have declared the enhancement factors by the film theory, where

$$E = 1 + \frac{K_{eq} * [CuAlCl_4]_{Bulk}}{(1 + K_{eq} * [CO]^*)} \quad (10.13)$$

Where, K_{eq} is the true equilibrium constant and $[CO]^*$ is the concentration of A at the gas-liquid interface and $[CuAlCl_4]_{Bulk}$ is the concentration of Copper in the bulk

The K_{eq} as function of the temperature is obtained from the literature Golaph and Chaudhari (1992) lit. [10.13] using the plot data of that publication.

In chemical absorption, the overall gas phase mass transfer coefficient K_{og} can be expressed in the form :

$$K_{og} * a = \frac{1}{\frac{1}{k_{g,CO} * a} + \frac{1}{E * H * k_{l,CO} * a}} \quad (10.14)$$

Where, E is the enhancement factor and H is the Henry constant obtained by physical absorption. In this case physical solubility in toluene has been aided from Seidell (1940) lit. [10.14].

The gas and liquid mass transfer coefficient have been obtained from the Onda model from Coulson (1983) lit. [10.15]

The mass-transfer coefficient in the liquid phase, $k_{l,CO}$ is given by the dimensionally consistent equation :

$$k_{l,CO} * \left(\frac{\rho_l}{g * \mu_l} \right)^{\frac{1}{3}} = 0.0051 * \left(\frac{L}{a_w * \mu_l} \right)^{\frac{2}{3}} * \left(\frac{\mu_l}{\rho_l * D_{l,CO}} \right)^{-\frac{1}{2}} * (a_p * d_p)^{-2} \quad (10.15)$$

The mass-transfer coefficient in the gas phase, $k_{g,CO}$ is given by the dimensionally consistent equation :

$$k_{g,CO} * \left(\frac{R * T}{a_p * D_{g,CO}} \right) = 5.23 * \left(\frac{G}{a_p * \mu_g} \right)^{0.7} * \left(\frac{\mu_g}{\rho_g * D_{g,CO}} \right)^{\frac{1}{3}} * (a_p * d_p)^{-2} \quad (10.16)$$

The wetted area is provided by :

$$a_w = a_p * (1 - \exp(-1.45 * \left(\frac{L}{a_p * \mu_l} \right)^{0.1} * \left(\frac{a_p * L^2}{g * \rho_l^2} \right)^{-0.05} * \left(\frac{L^2}{a_p * \sigma * \rho_l} \right)^{0.2} * \left(\frac{\sigma}{\sigma_c} \right)^{-0.75}) \quad (10.17)$$

For the viscosity of the liquid we used an complex equation :

$$\mu_l = \text{Exp} \left[-4.12227 - 2.3403 * \text{Exp}(C_2 * Ra^3) + D_2 * Ra^2 + E_2 * Ra + B_2 \right] \quad (10.18)$$

Where,

$$B_2 = \frac{(1051.795 + 2026.852 * \text{Exp}(F_2 * Ra^2 + G_2 * Ra))}{T_l} \quad (10.19)$$

$$C_2 = -5.08093E-2$$

$$D_2 = 12.2693E-2$$

$$E_2 = -6.78159E-2$$

$$F_2 = -9.6315E-2$$

$$G_2 = 8.026396E-2$$

Ra is the molar ratio toluene/copper and
Lo is the molar loading carbon monoxide/copper

For the density of the liquid we used the equation :

$$\rho_l = 1000 * \left[\left(\frac{1.5562}{Ra + 1} \right) + 0.1881 \right] - \left[\left(\frac{0.0002614}{Ra + 1} \right) + 0.0009207 \right] * (T_l - 273.15) - \left(\frac{Lo * 0.06}{Ra} \right) \quad (10.20)$$

The diffusion coefficient of carbon monoxide in the liquid and gas phase is respectively obtained from the Wilke and Chang equation and the Gilliland equation :

$$D_{g,CO} = \frac{1.173 * 10^{-13} * \sqrt{M_{avr,l}} * T_l}{\mu_l * (0.0307)^{0.6}} \quad (10.21)$$

$$D_{g,CO} = \frac{1.013 * 10^{-7} * T_g^{1.75} * \sqrt{\frac{1}{M_{CO}} + \frac{1}{M_{avr,g}}}}{P * \left(18.9^{\frac{1}{3}} + 10.5^{\frac{1}{3}} \right)^2} \quad (10.22)$$

For the mean viscosity of the gas phase we used the next equation :

$$\mu_l = 0.0172 + (T_{g2} - T_{g1}) * 10^{-4} \quad (10.23)$$

For the density of the gas phase we uses the Redlich & Kwong approach :

$$\rho_g = \frac{M_{avr,g}}{V_1} \quad (10.24)$$

$$V_1 = \frac{R * t_g}{P} + b_{RK} - \frac{a_{RK} * (V_1 - b_{RK})}{\sqrt{T_g} * P * V_1 * (V_1 + b_{RK})} \quad (10.25)$$

$$a_{RK} = 0.42748 * R^2 * \frac{T_{c,mix}^{2.5}}{P_{c,mix}} \quad b_{RK} = \frac{0.08664 * R * T_{c,mix}}{P_{c,mix}} \quad (10.26)$$

In this simultaneous mass and heat transfer process, analogy between mass and heat may be used to obtain the heat transfer coefficient. Since the liquid film resistance in the heat transfer process may be neglected, the interface temperature is very close to the liquid film temperature, so the heat transfer coefficient is obtained as follows :

$$\left(\frac{h_g * a}{k_{g,CO} * a} \right) = C_{pg} * (Le)^{\frac{2}{3}} = C_{pg} * \left(\frac{\lambda_g}{\rho_g * C_{pg} * D_{g,CO}} \right)^{\frac{2}{3}} \quad (10.27)$$

For the heat conductivity of the gas phase the next equation has been obtained :

$$\lambda_g = \mu_g * \left(C_{pg} + \frac{10.4}{M_{avr,g}} \right) \quad (10.28)$$

Meanwhile, for sake of the similarity between mass and heat transfer, mass flow rate not only affects on the mass transfer coefficient, but also affects the heat transfer coefficient. If in the chemical reaction absorption which involves appreciable concentration change, the result of this influence should be considered, and the heat transfer coefficient h_g should be modified to h_g' :

$$h'_g = -G_B^* \left(\frac{(C_{pCO}^* \frac{dY_{CO}}{dZ}) + C_{pTol}^* \frac{dY_{Tol}}{dZ}}{1 - \exp \left[G_B^* \frac{C_{pCO}^* \frac{dY_{CO}}{dZ} + C_{pTol}^* \frac{dY_{Tol}}{dZ}}{h_{g,CO}} \right]} \right) \quad (10.29)$$

The evaporation and condensation of the volatile solvent (toluene) may evaporate or condense along with temperature change in the packed column. This changeable phenomenon also influences the heat transfer process and has already been included in Eq. (10.29). In fact, the liquid layer on both the inner and outer surface of packings are available to the evaporation process, so the available wet surface is much larger than that of physical absorption, and the resistance of the liquid phase would be very small. However, in physical absorption the thickness of the liquid layer and the flow rate at each point of the wet surface of the packings are different, so the thin liquid layer or low mobility part of the liquid film would be saturated and might be hardly absorb the solute (carbon monoxide). That is the reason why there would be larger liquid resistance. Therefore, the liquid resistance of evaporation and condensation of toluene in comparison with that of absorption, can be neglected, so k_{gTol} is able to be obtained from Eq. (10.30) :

$$\frac{k_{g,Tol}}{k_{g,CO}} = \frac{D_{g,Tol}}{D_{g,CO}} \quad (10.30)$$

According to the performance of the process described, assumptions for the mathematical model have been reasonably illustrated :

- (1) For sake of the complexing absorption is a fast reaction, the bulk concentration may be considered as equilibrium concentration, and the concentration at the interface may be considered as the equilibrium concentration in physical absorption.
- (2) Since the liquid resistance can be ignored in the heat transfer process, the temperature of the interface is the same as that of the bulk of the liquid.
- (3) Since the liquid resistance can be ignored during toluene is evaporating, the partial pressure of toluene may approach to its vapor pressure.
In addition, backward mixing along axis would not be taken into account; the surface of heat transfer is the same as that of mass transfer.

10.4.2.3 Mathematical Modelling

(1) Equation of mass transfer rate :

1 Mass transfer of carbon monoxide :

$$N_{CO} * a * dz = K_{og} * a * P * (Y_{CO} - Y_{COe}) * dz = -G_B * dY_{CO} \quad (10.31)$$

From This equation, the rate of change of carbon monoxide concentration in differential packings is as follows :

$$\frac{dY_{CO}}{dz} = \frac{-K_{og} * a * P * (Y_{CO} - Y_{COe})}{G_B} \quad (10.32)$$

2 Mass transfer of toluene :

From Eq. (10.32), the rate of change of toluene concentration in differential packings is as follows :

$$N_{Tol} * a * dz = k_{gTol} * a * P * (Y_{Tol} - Y_{Tole}) * dz = -G_B * dY_{Tol} \quad (10.33)$$

Where k_{gTol} can be obtained from Eq. (10.30).

Hence, the change rate of toluene concentration in differential packings is :

$$\frac{dY_{Tol}}{dz} = \frac{-k_{gTol} * a * P * (Y_{Tol} - Y_{Tole})}{G_B} \quad (10.34)$$

(2) Enthalpy balance, in differential packings : (Fig.10.8)

1 For gas phase (I) :

OUT = IN - CONVERSION

IN : Input rate of enthalpy = $G_B * H_G$

OUT : Output rate of enthalpy =

$$G_B * (H_G + dH_G) - (G_B * dY_{CO}) * [C_{pCO} * (t_G - t_0) + H_{OS}] + \\ -(G_B * dY_{Tol}) * [C_{pTol} * (t_G - t_0) * H_V] \quad (10.35)$$

where the last two terms of heat transfer are caused by mass transfer and :

$$H_G = C_{pB} * (t_G - T_0) + Y_{CO} * [C_{pCO} * (t_G - t_0) + H_{OS}] + Y_T * [C_p * (t_G - t_0) + H_V] \quad (10.36)$$

$$dH_G = C_{pB} * dt_G + Y_{CO} * C_{pCO} * dt_G + [C_{pCO} * (t_G - t_0) + H_{OS}] * dY_{CO} + \\ Y_{Tol} * C_{pTol} * dt_G + [C_{pTol} * (t_G - t_0) + H_V] * dY_{Tol} \quad (10.37)$$

CONVERSION : the rate of heat transfer from gas to liquid phase =

$$q_G * a * dz = h_G * a * (t_G - t_L) * dz \quad (10.38)$$

therefore,

$$-G_B * (C_{pB} + C_{pCO} * Y_{CO} + C_{pTol} * Y_{Tol}) * dt_G = h_G * a * (t_G - t_L) * dz \quad (10.39)$$

Then the temperature profile of gas phase along the packings height is as follows :

$$\frac{dt_G}{dz} = \frac{-h_G * a * (t_G - t_L) * (C_{pB} + C_{pCO} * Y_{CO} + C_{pTol} * Y_{Tol})}{G_B} \quad (10.40)$$

If mass transfer flux is considerably large, h_G has to be modified as h_G' .

2 For liquid phase (II) :

OUT = IN - CONVERSION

IN : Input rate of enthalpy =

$$(L + dL)(h + dh_L) + (-G_B Y_{CO})[C_{pCO}(t_G - t_0) + H_{OS}] + (-G_B dY_{Tot})[C_{pTot}(t_G - t_0) + H_V] \quad (10.41)$$

OUT : Output rate of enthalpy = $L h_L$

CONVERSION : rate of transfer from liquid phase to gas phase =

$$-q_G a dz = -h_G a (t_G - t_L) dz \quad (10.42)$$

Then

$$L C_q \left(\frac{dt_L}{dz} \right) = G_B (C_{pB} + C_{pCO} Y_{CO} + C_{pTot} Y_{Tot}) \left(\frac{dt_G}{dz} \right) + G_B [(C_{pCO}(t_G - t_0) + H_{OS}) \left(\frac{dY_{CO}}{dz} \right) + G_B [C_{pTot}(t_G - t_0) + H_V] \left(\frac{dY_{Tot}}{dz} \right) \quad (10.43)$$

Where t_0 is the reference temperature. As a result, the temperature profile along the packings in the liquid phase dt_L/dz can be obtained from Eq. (10.32), (10.34), (10.40).

(3) The overall material balance and heat balance :

Overall material balance :

$$L_1 * X_{CO1} - L_2 * X_{CO2} = G_B * (Y_{CO1} - Y_{CO2}) \quad (10.44)$$

Where subscript 1 denotes the bottom and 2 denotes the top of the column,

$$Y_{CO} = \frac{y_{CO}}{(1 - y_{CO} - y_{Tot})}, \quad G_B = G_1 * (1 - y_{CO1} - y_{Tot1}) \quad (10.45)$$

From the overall heat balance, the temperature of the column bottom t_{L1} can be obtained :

$$Cp_{ql} * (t_{L2} * L_2 - t_{L1} * L_1) =$$

$$(G_2 * C_{pg2} * T_{g2} - G_1 * C_{pg1} * T_{g1}) + H_{OS} * G_B * (Y_{CO2} - Y_{CO1}) + H_V * G_B * (Y_{Tot2} - Y_{Tot1}) \quad (10.46)$$

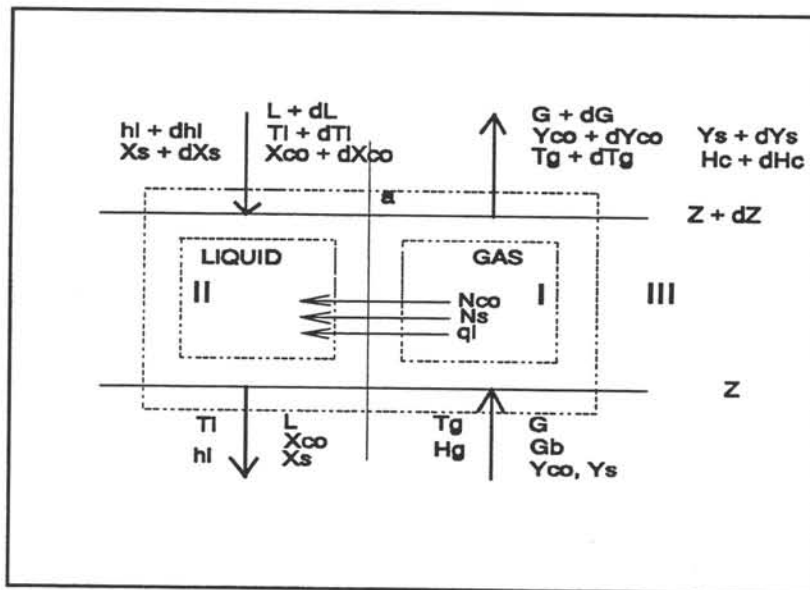
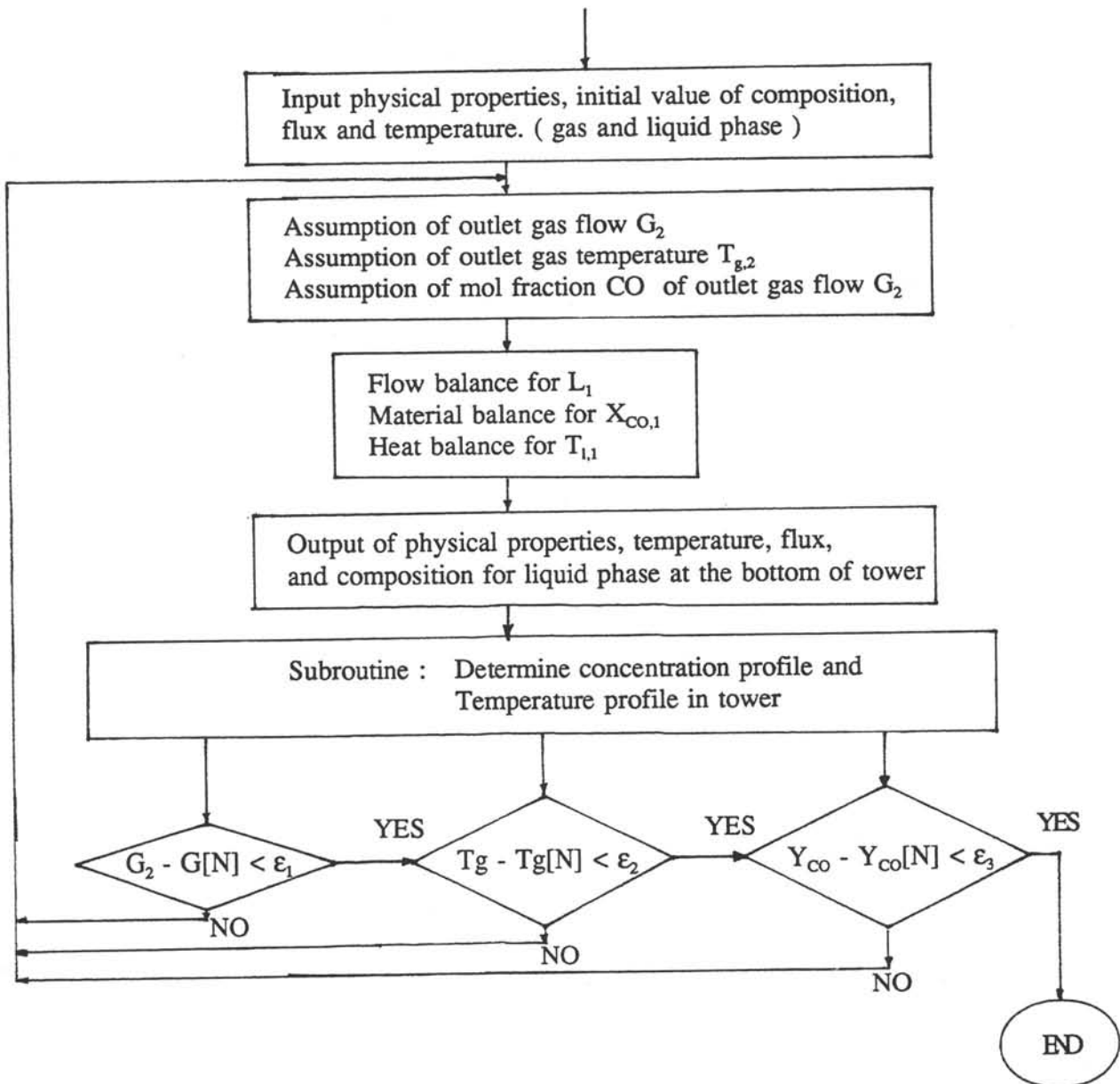


Figure 10.8 Mass and heat transfer in differential packings of the absorber.

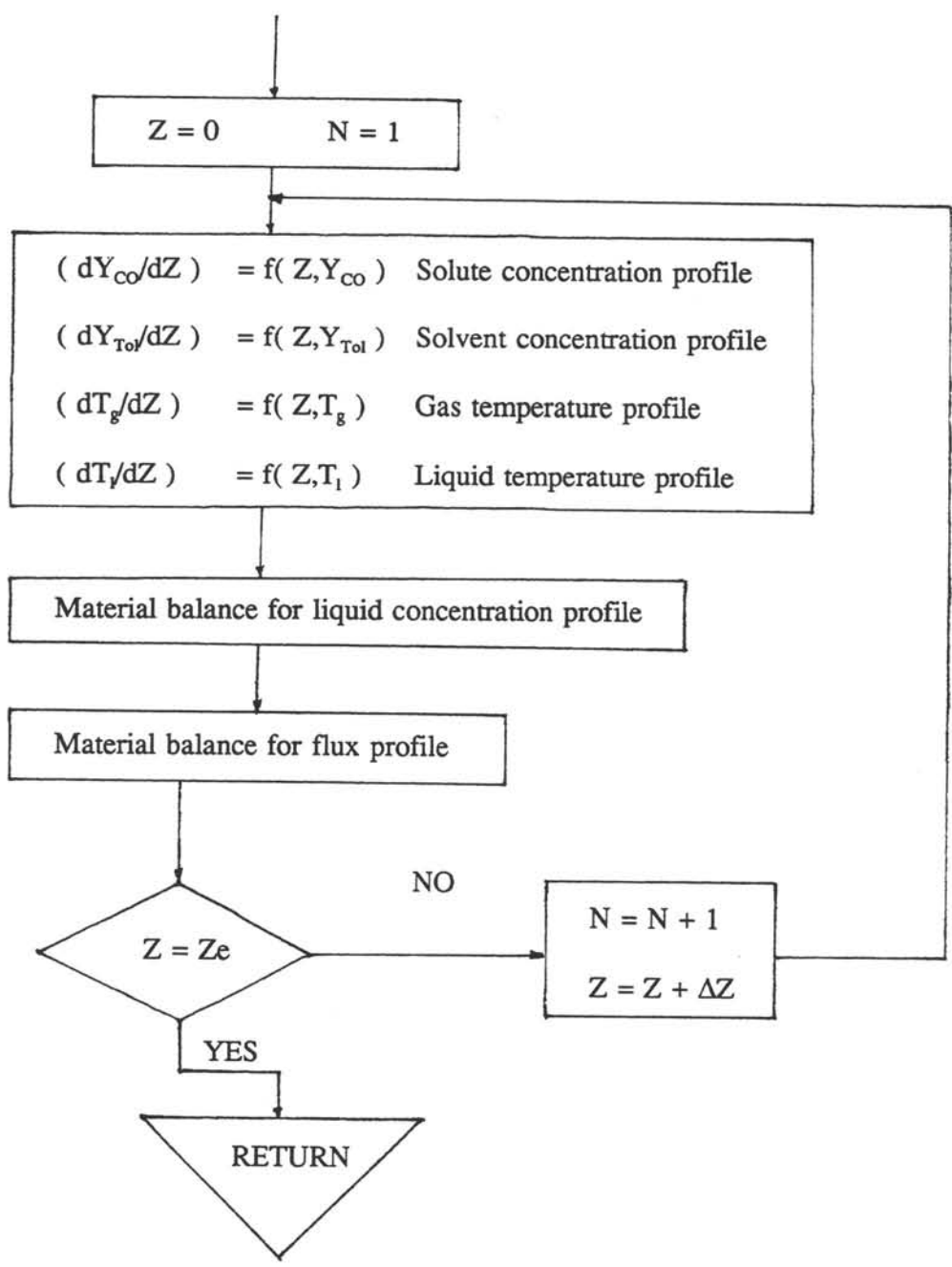
10.4.2.3 Computing and simulation

From the initial conditions, the flux of both gas and liquid phase, their concentration and temperature and from the known parameters-enthalpy of complexing, enthalpy of solvent evaporation, the specific heat, density, molar weight of gas and liquid mixture, diffusivity, enhancement factor, etc., the liquid flux, temperature and composition for the bottom of the tower can be determined from the total heat and total mass balance. Furthermore, with the aid of fourth-order Runge-Kutta approach, Eqs. (10.32), (10.34), (10.43) have been solved, from which the temperature profile and concentration profile along the packed tower has also been obtained.

The flow diagram for computing the carbon monoxide absorber tower have been shown in the following algorithms (The turbo Pascal program is printed in appendix 10) :



SUBROUTINE



10.4.2.4 Result and discussion

For the calculation of carbon monoxide complexing absorbed in the aluminum cuprous tetrachloride-toluene solution, the computer program COSORBER have been used, while the initial data of the COSORB[®] tower (table 10.4) have been put in, the results of computing conformed with the data obtained from the experimental of a pilot plant.

COMPOS. FEEDGAS	CO Flow	55.02 [kmol/h]	Temp. Gas	308.15 [K]
	H ₂ Flow	129.3 [kmol/h]	Tol. Flow	0.000 [kmol/h]
	CO ₂ Flow	29.35 [kmol/h]	Inrt. Flow	175.4 [kmol/h]
	N ₂ Flow	12.33 [kmol/h]	Tot. Flow	230.5 [kmol/h]
	CH ₄ Flow	4.412 [kmol/h]	Y _{CO} Flow	0.3136 [-]
	Ar Flow	0.000 [kmol/h]	Y _{Tol} Feed	0.0000 [-]
COSORB [®] SOLVENT	Tot. Flow	225.0 [kmol/h]	Temp. Liq.	313.15 [K]
	X _{CO} Feed	0.0028 [-]	Cu Flow	62.55 [Kmol/h]
	X _{Cu}	0.2774 [-]	CO Flow	0.631 [Kmol/h]
TOWER SPEC	Pressure	27.0 [bar]	Crit. Surf.	75 [Dyne/cm]
	Height	8.00 [m]	Packing	0.038 [m]
	Cross a.	0.70 [m]	Step h	0.04 [m]
	Spec. a.	130.0 [m]	Efficiency	0.90 [-]
ESTIMATIONS	Flow Top	175.0 [kmol/h]	# iteration	100 [-]
	Tg Top	314.15 [K]	Init. Flag	0 [-]
	Y _{CO} Top	0.00005 [-]	error	5E-6 [-]

Table 10.4 Input for COSORB[®] simulation with COSORBER (appendix 10)

For the gas feed we used the gas composition and gas flux of the calculated optimum from the reformer. For the COSORB[®] solvent we used the composition of the average analytical COSORB[®] data. For the packing of the tower we used metal pall rings (0.038 m). To get the required recovery the tower has to be 8 m high and a cross area of 0.7 m with a pressure of 28 bar.

Also the results of computing of the pressurized tower conformed with the theoretical regulations. The results of the input from table 10.4, have been shown in table 10.5 and the typical temperature profiles and concentration profiles along the packed height have been shown in Fig 10.9 and 10.10 .

It is visualized for the fact that the temperature of the liquid phase at the bottom (T_{L1}) decreases with the increasing of the liquid flux, and the required height of the packed tower also decreases. Meanwhile, the maximum temperature of the liquid has eliminated gradually, and similarly for the maximum temperature of the gas phase. At the same time, according to the different feed temperature of the gas phase at the bottom of the tower, the required height of the packings and the maximum gas temperature are different.

These results illustrated the reasonable accuracy of the mathematical model and simulation of the simultaneous mass and heat transfer process in the chemical absorber.

The average value of the overall mass transfer coefficient $K_{o,g}$ approaches to the experimental data in pilot plant. Because of the situation of the solute in liquid phase of the column bottom, the liquid resistance induced and the enhancement factor reduced, so the mass transfer coefficient is rather small. And because of the solute is rather concentrated in the gas phase of the tower bottom, the solute gas flux is rather large, so the gas resistance is considerably small and the liquid resistance is obviously the dominant.

Height [m]	G Flow [kmol/h]	L Flow [kmol/h]	Y_{CO} [-]	Y_{Tol} [-]	T_{liquid}	T_{gas}
0.0	230.452	279.876	2.387514E-1	0.0000000	359.32	308.15
0.4	192.477	241.901	7.992858E-2	8.626694E-3	328.86	311.05
0.8	181.145	230.057	2.259267E-2	5.609578E-3	319.19	312.15
1.2	177.663	227.088	8.407853E-3	4.151519E-3	316.13	312.62
1.6	176.573	225.997	2.825435E-3	3.635643E-3	315.18	312.89
2.0	176.229	225.653	1.057301E-3	3.467124E-3	314.89	313.07
2.4	176.121	225.546	4.983931E-4	3.414167E-3	314.81	313.23
2.8	176.087	225.512	3.218511E-4	3.399694E-3	314.80	313.36
3.2	176.076	225.501	2.661109E-4	3.396578E-3	314.80	313.49
3.6	176.074	225.498	2.485232E-4	3.397110E-3	314.81	313.60
4.0	176.073	225.498	2.429829E-4	3.398684E-3	314.82	313.71
4.4	176.073	225.498	2.412457E-4	3.400480E-3	314.83	313.81
4.8	176.073	225.498	2.407085E-4	3.402246E-3	314.84	313.90
5.2	176.074	225.498	2.405493E-4	3.403912E-3	314.85	313.98
5.6	176.074	225.498	2.405086E-4	3.405461E-3	314.86	314.06
6.0	176.074	225.499	2.405046E-4	3.406895E-3	314.87	314.13
6.4	176.074	225.499	2.405115E-4	3.408220E-3	314.88	314.19
6.8	176.075	225.499	2.405213E-4	3.409443E-3	314.89	314.25
7.2	176.075	225.499	2.405313E-4	3.410573E-3	314.90	314.31
7.6	176.075	225.499	2.405409E-4	3.411616E-3	314.91	314.36
8.0	176.075	225.500	2.405498E-4	3.412579E-3	314.92	314.41

Table 10.5 Results of the COSORB[®] simulation output

Up in the packings, the gas resistance increased and the mass transfer coefficient decreased because of the higher efficiency of the absorption and the rapidly reducing solute concentration. However the gas solute concentration decreases while the height of the packings increases and the active concentration in absorbent increases, so the enhancement factor increases, and the liquid mass transfer coefficient at the top nearly doubles, but still smaller than the gas phase. This phenomenon becomes more appreciable when the liquid flux is smaller. This agrees with the fact and also explores the mechanism of the process. In the pressurized tower, if the same absorbent flux, temperature and feed composition are used, the required recovery for the same packings height would be higher than that of the conventional tower. This fact is caused by the increasing of the overall mass transfer coefficient. Further to analyze the mass transfer coefficient of the gas and liquid phase respectively, it would show that because the gas diffusivity reduces fifteen times (if 15 atm of pressure in the tower), the corresponding mass transfer coefficient decreases appreciably.

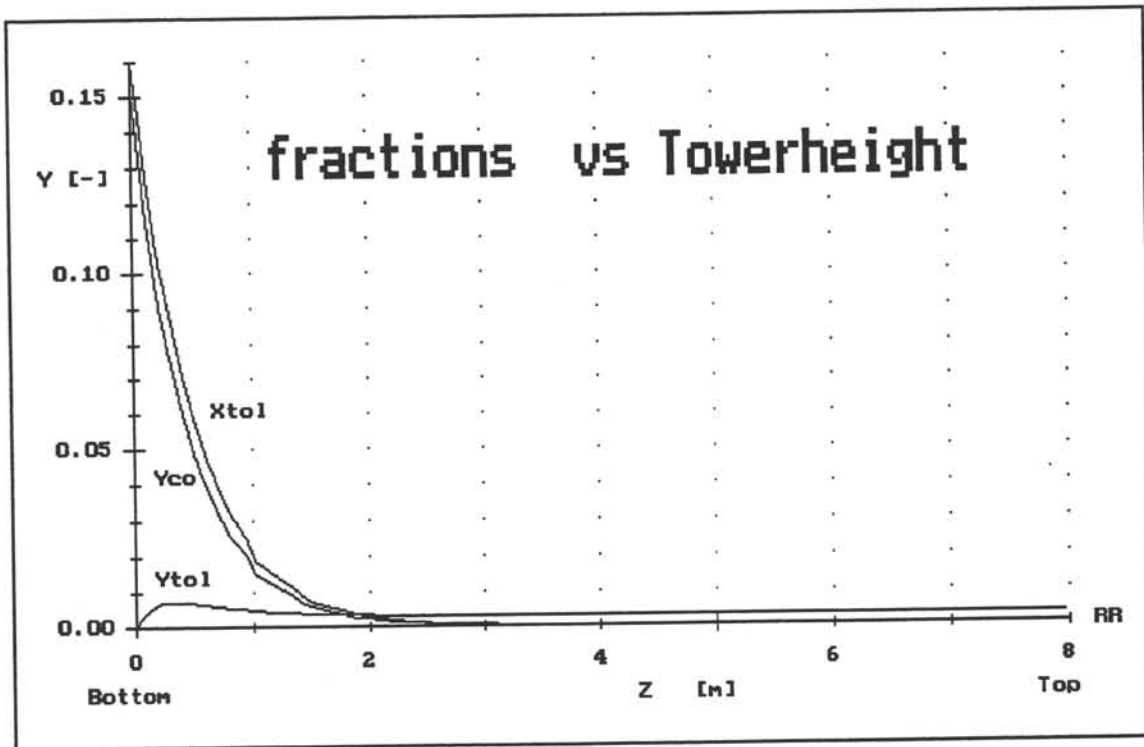


fig 10.9 The concentration profile of solvent and solute in the column

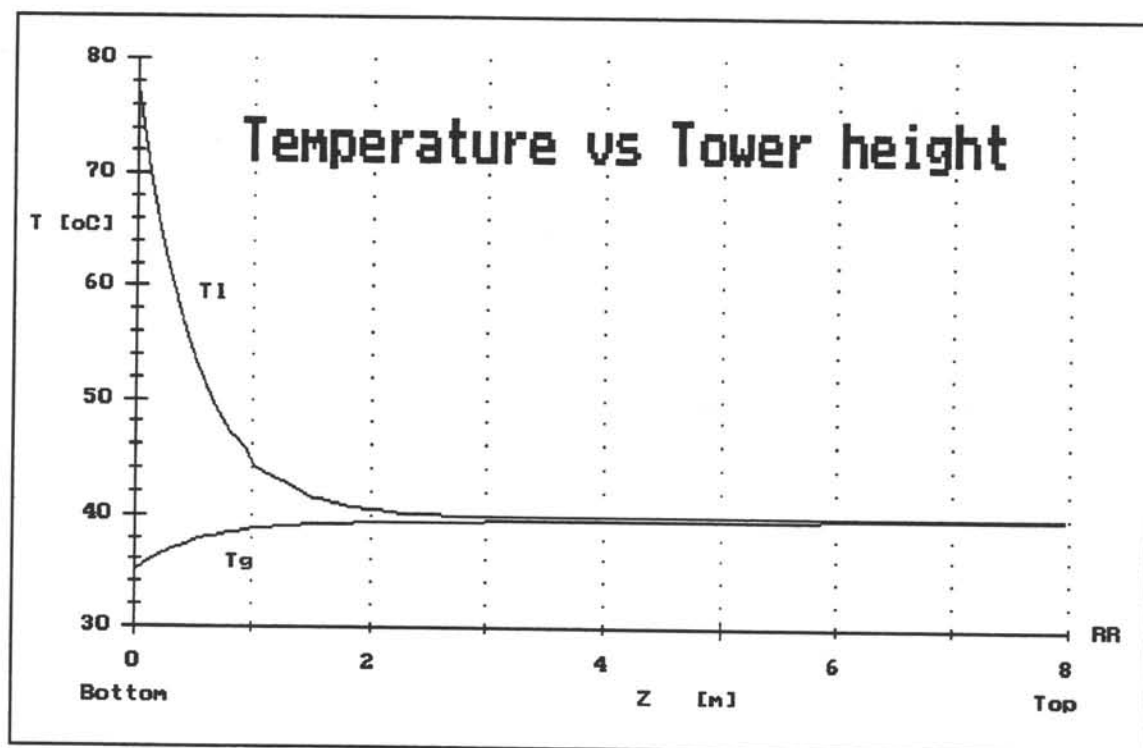


fig 10.10 Temperature profile of gas and liquid in absorber

However, the liquid phase in the pressurized tower reacts with the highly concentrated solute, the rate of complexing absorption increases while the surface tension on the wet packings increases, the convection effect and the surface renewal of the liquid phase increases, so that the enhancement factor and the rate of absorption increased. As a result, the liquid mass transfer coefficient at conventional absorber, the liquid mass transfer coefficient at the bottom of the pressurized absorber also increases several times compared with that in the conventional absorber. Obviously, the gas film resistance would be the dominant factor along the whole tower. From the result, it can be shown that the main factors for the change of gas and liquid film resistance in the chemical absorption process. Due to the different operating conditions, the result of simulating and computing, the mechanism and principles of the process can be explored clearly.

Since in this process the input gas is wet, the evaporation and condensation of the volatile solvent is limited therefore, the maximum temperature of the liquid is not appreciable, that means there would be no hot point along the absorber. (fig 10.10). The minimum driving force region is close to the bottom of the absorber, while the ratio of the liquid and gas is quite small, the minimum driving force region is still at the top of the tower. Because of the required high recovery, the concentration of the active component of the gas phase is so dilute that it is difficult to be absorbed.

10.4.2.5 Costs of COSORB®

The investment costs, feed gas and utility requirements for a reformer feed gas is presented in appendix 10 table A 10.10 . The investment requirements for the battery limits of COSORB® units include the initial charge of all expendable items, but do not include peripheral facilities such as buildings, flare systems and other equipment or process systems associated with supplying utilities, or administration of the unit. The utilities shown are based on operating the plant at design capacity.

The feed gas composition to the COSORB® unit is saturated with water in addition to the components shown in table 10.4 and contains particle matters and sulfur compounds. The gas stream require compression and cooling before particulates sulfur and water can be removed economically. The investment and operating costs presented herein include feed gas compression, water removal, sulfur removal and particulate removal facilities necessary to permit feeding the synthesis gas steam to COSORB® unit complexor.

The recovery costs consists the operating and investment cost for the COSORB® unit. The required sales price is the recovery cost plus net cost of the synthesis gas feed stream. The variable and fixed costs of recovering CO were calculated at design capacities of 1500 kg/h of CO. Fixed cost are based on 8,000 operating hours per year. Manpower includes one operator plus a fraction of direct supervision. Maintenance and taxes are calculated using industrial averages as shown, and depreciation is included based on ten year straight line rates. The annual cost of investment is taken as 30 percent of the installed cost of the battery limits COSORB® unit. This is equivalent to assuming a rate of return on investment of 30 percent before taxes. The recovery costs including investment costs are presented in appendix 10 table A 10.11 .

10.4.2.6 Conclusion

The mathematical model of adiabatic absorber with chemical reaction has been computed by the computer program COSORBER for the COSORB® system, and the results has conformed with the experimental data of a pilot plant. The enhancement factor, mass transfer coefficient and the profile of both temperature and concentration for both gas and liquid phase have all been described and visualized. By means of the forth order Rung-Kutta method, the differential equations of both heat and mass transfer have been solved, the results are reasonably accurate and less time consuming. Further more, the more precise of this model for scale up designing would be suggested as follows :

The effect of back mixing along the axis may be provided to modify the original one, also, the influence on the boundary and bulk conditions by simultaneous mass and heat transfer need to be further examined.

10.5 produced H₂ in a fuel cell system

Looking at the CO plant it is obvious that not only CO is produced but also a lot of H₂. In plant II for example the H₂ production is now some 150 kmol/h. Most of this quantity is used as fuel for the heating of the reformer. This is a rather expensive heating fuel. Having in mind that for selling the H₂ purifying is needed, another possible use of the hydrogen in the future could be a fuel cell system.

Fuel cell systems are electrochemical conversion systems, where the chemical energy escaping from the oxidation reaction, directly is transformed into electrical energy. For that purpose H₂ is the feed of the anode and O₂ out of the air is the feed to the cathode. The electrochemical reaction takes place at the electrodes, where an electric voltage is generated. As a result of ion-transport in the cell - using an electrolyte between the electrodes - a electron transport using an extern circuit exists, which can deliver electrical power.

In the past years a lot of different fuel cell systems are developed. Generally those systems are called after there electrolyte. The most important fuel cell are the alkaline fuel cell (AFC), the phosphoric acid fuel cell (PAFC), the molten carbonate fuel cell (MCFC), the solid oxide fuel cell (SOFC) and the polymer electrolyte fuel cell (PEFC). The temperature in those cells varies from 80 °C (PEFC) to 1000 °C (SOFC). For electricity production the most important are the PAFC-, the MCFC- and the SOFC-systems. These systems produce a DC-voltage combined with heat.

The classical natural gas systems for electricity production has a maximum efficiency of 40 - 45 %. The modern STEG-units (steam- and gas-turbines) have a maximum efficiency of 50 - 55 %. At this moment an efficiency of 50 % is the technical limit. The conversion of hydrogen and oxygen in a fuel cell is not a Carnot-cycle with its thermodynamic laws, but a electrochemical proces. That is why the efficiencies can be higher in a fuel cell. The results of a study of an MCFC-system show that the efficiency of such a system is about 52 -58 %. In fig.10.11 a comparison between efficiencies of different electrical power systems is made. The good environment characteristics are also a result of the fact that fuel cells are working with electrochemical conversion. Because of the fact that there is no direct combustion of fuel and oxygen the emission of SO_x, NO_x, CO and hydrocarbons can be very low.

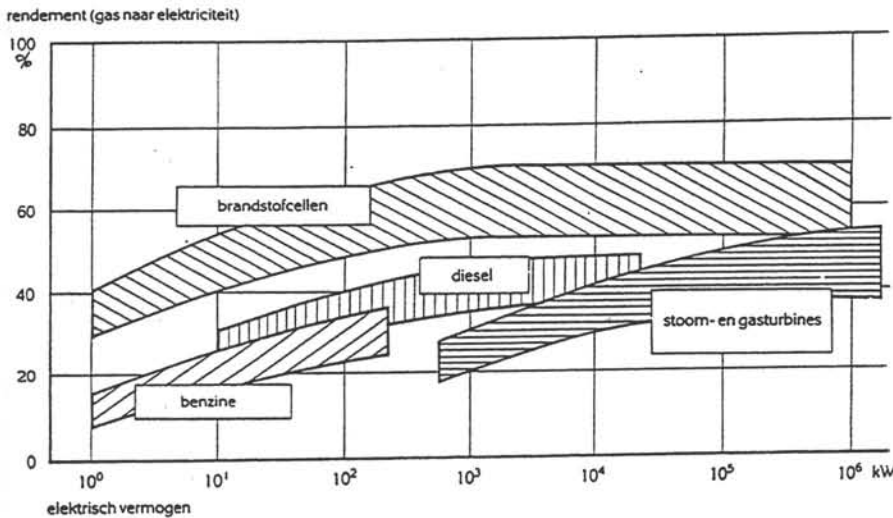


fig.10.11 : efficiencies of different systems.

The characteristics and advantages of fuel cell systems make a variety of applications possible. Fuel cells can be placed for generating a continue basic load, which is interesting for its high efficiency. The fact that the high electrical efficiency also remains by peak loads, makes fuel cells possible for so called loadfollowing uses, but in this case the use of gasturbines is preferred because of the lower installation costs and the fast characteristics of gasturbines. The most promising application in the near future is the cogeneration of electricity and heat. In first instance only PAFC-systems will be commercial available, but these systems have the disadvantage of producing heat at low temperature and have only an efficiency of 35-45 %. More useful, but only commercial in the second half of the nineties, are the MCFC- and the SOFC-systems. These systems produce high quality heat and have an efficiency of almost 60 %. This is the reason for choosing for the MCFC-system.

The hart of a fuel cell system is formed by the fuel cell itself. A fuel cell consists of a stack of basic-elements, such as a anode, cathode and in between the characteristic electrolyte. These cells, connected in series, are separated by a bipolar separationplate. which has not only a separation function but has also the function of distribution the fuel and the oxide-gasses correct. A schematic view of a fuel cell is given in fig. 10.12 .

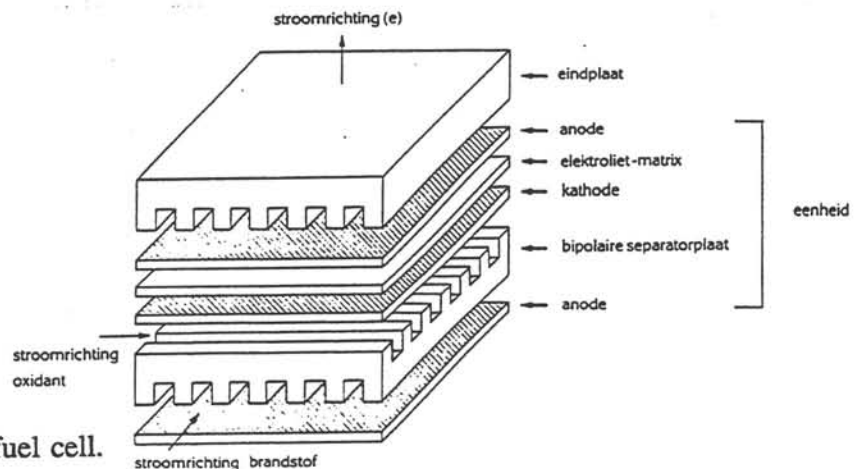


Fig.10.12: lay-out of a fuel cell.

The reactions that take place in the fuel cell (MCFC) are:

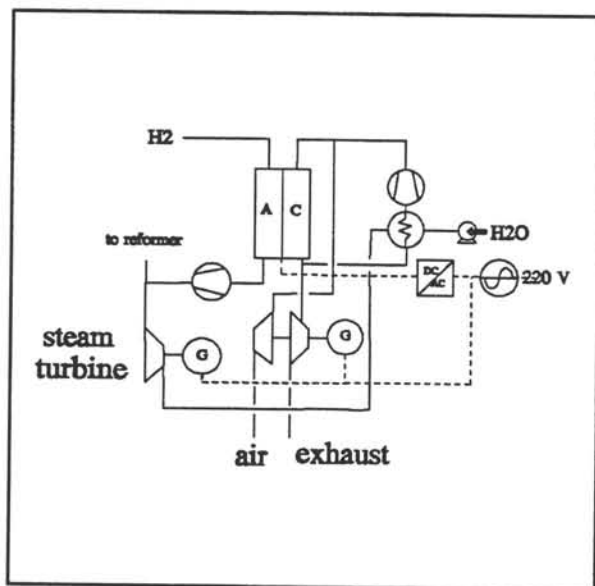
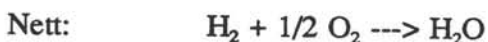
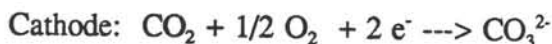


Fig 10.13 : MCFC-system

Fig.10.13 gives the circuit diagram of a MCFC-system with H_2 and air as feed. The feed reacts at the anode and produces steam, carbon dioxide and electricity. Some CO_2 is mixed with air and fed to the cathode after being compressed. The cathode off-gas comes free under pressure and can deliver power for compression air and electricity generation. The pressure is one of the most important operating parameter. At high pressure the fuel cell gives better results than at atmospheric pressure. This difference can be explained by the Nernst equation (10.47). This equation gives the correlation between the open cell voltage (OCV) and the gas composition:

$$OCV = E^0 + \frac{RT}{2F} * \ln \frac{[pH_2]}{[pCO_2] * [pH_2O]} + \frac{RT}{2F} * \ln [pCO_2] * [pO_2]^{1/2} \quad (10.47)$$

- With :
- E_0 = the OCV by reference conditions
 - R = universal gas constant
 - T = absolute temperature
 - F = Faraday constant
 - p = partial pressure

Looking at the Nernst equation, it follows that the OCV is higher when the partial pressures from H_2 and O_2 are higher. The addition of inert components, such as N_2 from the air, will lower the partial pressure which will also lower the OCV. The best is to avoid inert components as much as possible. This is possible with recycle flows. A higher operating pressure gives not only a higher efficiency of the cell but also the volume of the gas decreases which results in smaller recycle compressors. On the other hand the performance of the steam turbine decreases with higher pressure. The optimum operation pressure is 7-8 bar.

The question of this system will be economical, has to be calculated later when such system is on the commercial market. At this moment several institutions around the world are working on the development of fuel cells. In the Netherlands there is "BCN" which means BrandstofCel Nederland (Fuel Cell Holland). This is a cooperation between "Verenigde Machine Fabrieken Stork NV, De Koninklijke Schelde Groep BV, ECN (Energy Centre Netherlands) and "Interduct clean technologie Delft".

The only economical system is the PAFC system. In lit [10.18] a study at the economics of a H_2 fuel cell is made. In Table 10.6 below are the results for the situation of a 20 % more CO producing plant with a H_2 separation unit which recovers 99% of the hydrogen at high purity.

Table 10.6 : cost estimation of H_2 fuel cell

Costs, in Dfl * 1000 (1990)	H_2 -PAFC stack 4 MW
investment costs	8000
annual capital costs	1248
operation & maintenance	400
over-all efficiency (%)	52
fuel (1 Nm ³ H_2 = 0.34 Nm ³ NG)	1690
total annual costs	3338
kWh's produced	32 * 10 ⁶
kWh-cost (ct/kWh _e)	10.4

The investment costs are based on data presented in lit.[10.19]. The investment cost of the fuel cell is the average of two figures mentioned : \$ 1056 / kW (1980) and fl. 3960 for small units (1990). As the current system is not a small unit the price per kW is fl.2000.-. Capital charge is based on 15.6% (10 years, 9 % interest rate). Maintenance and operational costs are assumed at annual 5% of the total investment costs, whereas the stack is supposed to run during 8000 hrs/a. The H₂ fuel price is the price of the amount natural gas with the same heat value. Also given in lit.[10.19] is the fact that 1 Nm³/h of hydrogen corresponds to 1.23 kW electric power.

Looking at the price for one kWh electricity (10.4 ct/kWh) and comparing this with the price of the electricity on the plant (6.4 ct/kWh), the first conclusion is that the H₂ flow can not be used in a fuel cell system. However when you take into account that:

- 1) the calculations made in this paragraph are very rough.
- 2) the price of a fuel cell will decrease rapidly when large amounts are being produced.
- 3) because of the fact that this is a clean technology, maybe a subsidy will be given by the government.
- 4) the price of 10.4 ct/kWh is lower than the price of the common household user.

So there is a possibility of selling the electricity to the local power station, just as electricity from windmill generators is being sold.

10.6 EVALUATING THE NEW PLANT DESIGN

The new plant consists of various units. After the reformer there are several separation equipments. In this paragraph the influence of the different units and their conditions will be considered in order to understand the behavior of the total plant. The equipment that are involved in the plant are a reformer, a dryer, a COSORB[®] or a CO-vacuum swing adsorption unit (CO-VSA), and possibly a H₂-pressure swing adsorption (H₂-PSA) and a N₂-removal unit. The conditions of the several equipment are shown in fig.10.14

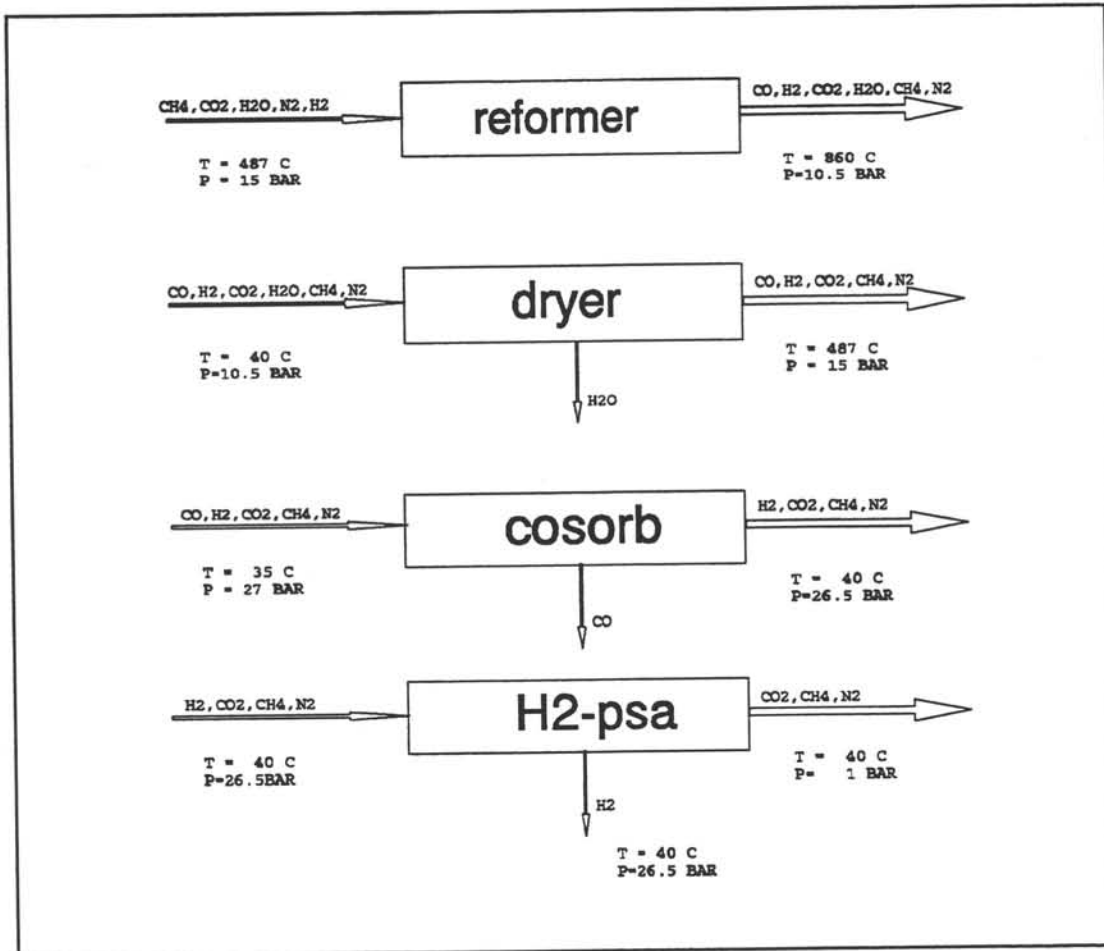


fig.10.14 : working conditions of equipment

After the reformer the gas stream consists mainly of CO, H₂, CO₂, CH₄, H₂O and N₂. In the dryer all the H₂O from the gas stream has to be removed before the gas is led to the COSORB[®] unit. In the COSORB[®] or CO-VSA unit almost all the CO (>99.9%) is removed with high purity (>99.9%). The other stream which leaves the unit consists mainly of CO₂, H₂, CH₄ and N₂. With this stream it is possible to do several possibilities, which now will be discussed.

One option is to remove all H_2 in order to sell it. The H_2 -PSA unit removes almost all the H_2 (99.9%) with high purity (99.9%), which is necessary if the H_2 will be sold. The other gas stream which leaves the PSA unit contains CO_2 , CH_4 and N_2 , so a part of this stream could be recycled to the reformer. This stream has to be compressed from 1 to 14.2 bar and the temperature has to be brought from $40^\circ C$ to $487^\circ C$. A part of the stream has to be purged in order to prevent accumulation of some substances. Because not all CO_2 reacts in the reformer, the little purge stream may consist of a high CO_2 -percentage. A little MEA-stripper could be used to remove this CO_2 , which will be recycled to the reformer, so that less CO_2 will be required in the feed and the feedcost will decrease.

Another possibility is to recycle a part of the H_2 from the H_2 -PSA unit to the reformer in order to increase the CO-production. This is achieved by the fact that H_2 will react with the CO_2 , so that more CO_2 (and less CH_4) is converted to CO in the reformer. A little MEA stripper for removing the CO_2 from the purge gas can also be used in this case. It seems that if 60% of the produced H_2 is recycled to the reformer a maximum CO production can be achieved with a certain feed. Higher H_2 -recycles decrease the CO-production, because too much H_2 decreases the conversion of CH_4 to CO and H_2 .

If it is able to work with low purgestreams, the benefit of recycling H_2 becomes insignificant and the first option is recommended.

As the purge stream seems to have very little combustion value, it has no use burning this stream in the furnace.

The effect of a N_2 -removal unit for the feedgas is that less N_2 is circulating in the recycleloop so that less purging is possible. If less purging is possible, less feed is required and the feedcost is lowered.

The amount of H_2 which is produced by the H_2 -PSA depends on the feedgas composition. If a certain feed is used, this amount does not seem to differ very much in the cases recycling H_2 to the reformer or not. When recycling H_2 , the amount of H_2 which come out of the reformer increases also.

Another option is to work without a H_2 -PSA unit. The gas stream coming out from the COSORB[®] or CO-VSA unit contains H_2 , CO_2 , CH_4 and N_2 , so part of this stream is recycled to the reformer while the other part is purged in order to prevent accumulation. The recycle stream doesn't have to be compressed, but it has to be brought to a temperature of $487^\circ C$. If the recycle is getting too much (so less is being purged), H_2 is piling up in the loop. This can decrease the conversion of CH_4 to CO and H_2 , which results in a lower CO production. To solve this problem more CO_2 and less CH_4 (and so less steam) is fed to the reformer. The use of a N_2 -removal unit is only advantageous if high recycles are being applied. As no higher recycle than 60% is applied, this unit will not be necessary in this case. The purge stream consists mainly of H_2 and also some CO_2 , and the use of a MEA stripper to remove the CO_2 from the stream before burning it in the furnace have to be considered.

In the reformer mentioned in the earlier chapters, a gas stream of about 300 kmol/hr is going into the reformer (in the 20% capacity increased case). If the same reformer is being considered, the amount of gas going into the reformer at one of the new possibilities must be lower than this value. The gastemperature in the tubes will otherwise not be able to rise from 487°C to 860°C due to the decreased heat transfer, and the reaction will not be able to occur as required.

With recycling the different streams not only the amount of the stream that goes into the reformer should be considered, but also the composition of the stream should be analyzed. This is very important, because coke formation in the reformer has to be avoided. This can be done by adding steam. Also more CO₂ has also has a positive influence on avoiding coke formation. More steam is required when H₂ is recycled, because then more CO₂ reacts, which result in less CO₂ in the recycleloop. The problem that occur in this case, is that more steam has a negative effect on the CO-production. This can be solved by adding more CO₂ instead of steam and less CH₄.

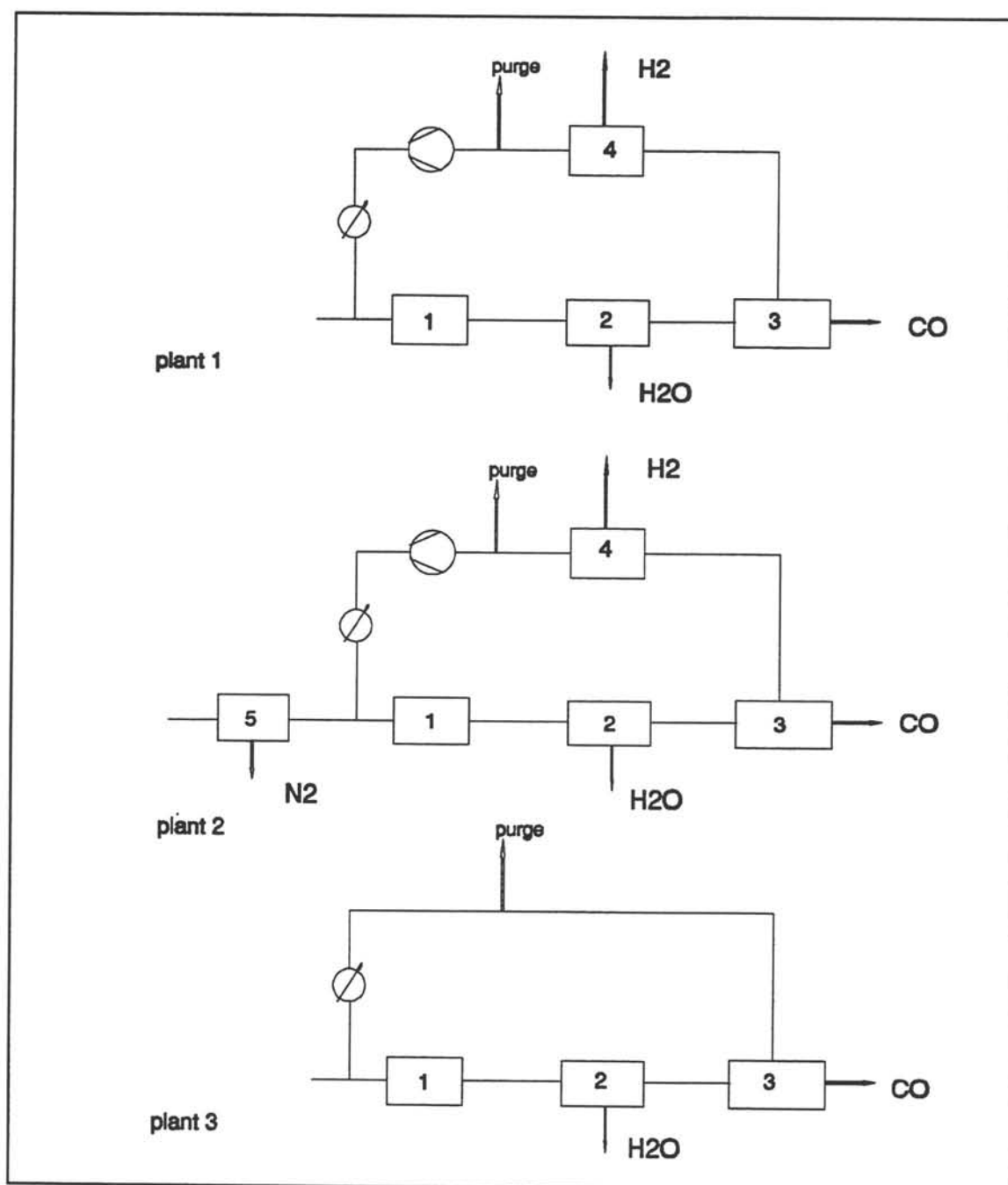


Fig 10.15: three options considered

- index: 1 = CO₂-reformer
 2 = dryer
 3 = CO-VSA
 4 = H₂-PSA
 5 = N₂ removal unit

10.7 CONCLUSIONS FOR THE NEW PLANT

To come to a conclusion which plant is economically the best for producing 55 kmol CO/hr the original reformer with different feed is used. The cases that are considered here are :

- PLANT 1: a plant with a H₂ -PSA and without a N₂-removal unit. No H₂ from the H₂-PSA unit is being applied as feed to the reformer. A recycle of 90% is applied.
- PLANT 2: a plant with a H₂ -PSA and with a N₂-removal unit. No H₂ from the H₂-PSA unit is being applied as feed to the reformer. A recycle of 60% is applied.
- PLANT 3: a plant without a H₂ -PSA and without a N₂-removal unit. As a recycle of 60% is applied, a N₂-removal unit is not necessary.

The three plants are shown in fig.10.15. The different streams that are involved in the three processes are shown in table 10.7 to 10.9. The feedgas compositions are calculated by trial and error with a computerprogram, Mercury, which simulated the plant mathematically. When searching for a feed for the plants, some aspects had to be taken into consideration. These where:

- a. the plant has to produce 55 kmol CO/hr.
- b. the amount of gas into the reformer had to be about 250 kmol/hr.
- c. the coke formation criterium has to be applied:

$$\frac{1/2CO_2 + H_2O}{CH_4} > 3 \quad (10.46)$$

To work safe a value of 3.5 is applied.

- d. the amount of all the gasses except for H₂ going into the H₂ -PSA may not exceed a value of 60 kmol/hr, otherwise this unit would become too expensive.
- e. the amount of CO₂ in the feed has to brought to a minimum, because CO₂ is expensive

Table 10.7 : streams of plant 1

PLANT 1 (kmol/h)	H ₂	CO	CO ₂	CH ₄	H ₂ O	N ₂
feed			34	39	110	6.86
recycle			25.55	1.11		10.29
reformer in			59.55	40.11	110	17.15
reformer out	97.81	55.23	42.58	1.85	88.71	17.15
products	97.81	55.23				
purge			17.03	0.74		6.86

Table 10.8 : streams of plant 2

PLANT 2 (kmol/h)	H ₂	CO	CO ₂	CH ₄	H ₂ O	N ₂
feed			20.4	39	113	1.36
recycle			37.93	1.75		12.24
reformer in			58.33	40.75	113	13.60
reformer out	100.2	54.99	42.15	1.94	90.38	13.60
products	100.2	54.99				
purge			4.22	0.19		1.36

Table 10.9 : streams of plant 3

PLANT 3 (kmol/h)	H ₂	CO	CO ₂	CH ₄	H ₂ O	N ₂
feed			40	28.5	80	4.97
recycle	77.92		17.61	2.65		7.36
reformer in	77.92		57.61	31.15	80	12.33
reformer out	129	55.00	29.35	4.41	81.53	12.33
products		55.00				
purge	51.08		11.74	1.77		4.97

To make an estimation for the prices for producing CO, the method of chapter 9 is applied. However, the cost estimation of this chapter is not as accurate as in chapter 9, because in chapter 9 flowsheets of a consisting plant were available.

Chosen is for a CO-VSA unit, because the price of this unit seems to be lower than that for a COSORB® unit. The investment cost of the N₂-removal unit is estimated to be \$ 3.000.000.

In order to calculate the costs of the utilities, the heat duties of the reformer, compressor work and the heat duties for the different recycles were calculated.

In table 10.10 the investment costs of the three plants are calculated. As expected the investment costs are the cheapest for plant 3 and most expensive for plant 2. Finally in table 10.11 the production cost of CO is calculated. It can be seen that although the price of raw materials and utilities of plant 3 is the highest, the production costs for this plant are the lowest.

Table 10.10 : investment costs for the three options

Costs in 10 ³ \$	plant 1	plant 2	plant 3
Furnace	885.5	885.5	885.5
Dryer system	75	75	75
Recycle compressor	39.4	48.5	
Compressor drivers	9.2	12.3	
CO-VSA			516
CO-VSA + H ₂ -PSA	1074	1074	
N ₂ -removal unit		570	
PCE	2083.1	2653.1	1476.5

Total PPC	PPC=f _a *PCE	6770.1	8622.6	4798.6
Fixed Capital	FC=f _b *PPC	9816.6	12502.8	6958.0
Working Capital (10%)		981.7	1250.3	695.8
Investment Cost		10798.3	13753.1	7653.8

Table 10.11 : CO-product cost calculation

A. Variable Costs in 10 ³ \$	plant 1	plant 2	plant 3
1. Raw materials			
- Feed gas price	2769	2246	2619
- Net H ₂ Credit	-640	-656	
2. Utilities			
- Electricity for recycle compressors	39	55	
- Natural Gas:			
- heat exchangers	8	11	48
- furnace	416	419	
Subtotal A	2592	2075	2667

B. Fixed Costs in 10 ³ \$			
1. Maintenance	981.7	1250.3	695.8
2. Operating Labour	824	824	824
3. Supervision	165	165	165
4. Plant Overheads	412	412	412
5. Capital charges	1472.5	1875.4	1043.7
6. Insurance	98.5	125	69.6
7. Rates	196.4	250	139.2
8. Royalties	98.2	125	69.6
Subtotal B	4248	5026.7	3418.9

Direct Production Costs A+B	6840	7102	6086
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C. Share of the Company's general operating expenses (25% of direct prod. costs)	1710	1776	1522
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Annual Operating Costs (A + B + C) in 10 ³ \$	8550	8878	7608
Production Cost (1540.6 kg CO/hr)	\$ 0.67/kg	\$ 0.70/kg	\$ 0.60/kg

When a new reformer should be designed, it could be designed for bigger gas streams so that it is possible to work with H₂ -recycles or with higher total recycles if no H₂ -PSA unit is used. This can cause a cheaper feed gas, because less feed is then required to produce the same amount of CO. But there are also some disadvantages when working under these conditions. The recycle streams have to be brought to reformer conditions (temperature, pressure) and more heat will be required in the reformer as the streams are bigger. All these factors should be considered when a new reformer will be designed, because a theoretical optimum does not have to be a practical optimum, just because of the economic aspects that are involved.

10.8 CONCLUSIONS AND RECOMMENDATIONS

Searching for an alternative CO-production process, several different options are found. For these options the costs for investment, feed, maintenance, operating and labour are calculated. After evaluation of costs, the most economical CO-production plant is chosen. The conclusions, based on the different options mentioned in this chapter, are:

1. For a CO production process is coal gasification only economical with a production higher than 6.5 ton/hr. Because the production in the existing plant is only 1.22 ton/hr., the choice is made to use the existing furnace fed with a different feed containing much CO₂.
2. After cost evaluation, the most economical way to produce CO is with a plant consisting of a CO₂-reformer and a CO-Vacuum Swing Adsorption system. The CO is produced for \$ 0.60 /kg CO.
3. The COSORB[®] process is reliable process to recover CO from a mixed gas stream. However the investment cost are rather high the COSORB[®] process operating cost are low. The recovery cost of CO amount \$ 0.70 /kg CO.
4. If there is a H₂ rich stream coming out of the plant, the H₂ can be put in a Phosphoric Acid Fuel Cell (PACF) to produce electricity for 10.4 dutch cents/ kWh. This is more than the electricity generated on the site, but less expensive than the normal electricity rate.

RECOMMENDATIONS

1. Because the equipment prices are coming from different sources, it is recommendable to evaluate the actual prices on the current market for the different equipment. This might change the price for producing CO.
2. Some further investigations can be made on the recent developments in the area of H₂ membrane separation systems. This is recommendable because the developments for such systems are going fast.

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11 CONCLUSIONS AND RECOMMANDATIONS

11.1 Conclusions

- The capacity increase of 20% is possible in two ways:
 1. Adding more CO₂ in the feed stream.
 2. Adding more CO₂ in the feed stream in combination with a pre-reformer.Both solutions lead to the same CO price (\$ 0.88). This is less expensive than the actual CO price (\$ 1.03). Option 1 will make adaption of the MEA equipment necessary. Option 2 will not, but a pre-reformer has to be installed. The choice will depend on costs and ease of installation.
- Separate N₂ removal is economically not a better option for the current process, although it helps to increase purity.
- A completely different plant designed by group 4 could produce CO for \$0.60/kg. This is less expensive than the current CO production which has a CO price of \$0.78/kg.
- Coal gasification is not a good option because it is only economically by productions higher than 6.5 ton/hr.
- ASPEN is a very good aid designing new processes and process equipment. However working with ASPEN gave a lot of difficulties but most of them were problems with our computernetwork. The large memory use also gave some problems. The program itself has been very usefull to calculate the new streams and all physical properties of the stream but the enthalpy balances were unreliable.
- As alternative separation methods only cryogenic separation and PSA have to be taken into account. At this point membranes are commercially not available. For absorption is at this point no absorbent available. Cryogenic separation can be done by using a process designed by Linde or a process designed by Griesheim GmbH. Linde's process is commercially available. Both methods remove approximately 80% N₂. PSA is too experimentally at this moment. If applied, it will only work in smaller plants like the one in use.

11.2 Recommendations

- The 20% capacity increase by adding CO₂ to the feed in combination with a pre-reformer is a good option and relatively inexpensive.
- The effect of the feed stream with a smaller amount of nitrogen on the Coldbox should be investigated.



12. EXPERIENCES WITH ASPEN PLUS.

12.1. Introduction to Aspen-Plus.

Aspen Plus (hereafter to be referred to as Aspen) is one of the many commercially available packages for the simulation of processes in the chemical industry. The use of Aspen is limited to static simulations, so no time-dependent processes can be modelled. The version used was release 8.5-4, in combination with ModelManager version 3-1.

12.2. How to create an Aspen input file.

Since the release of ModelManager, creating an Aspen input file has become very simple, especially when using ModelManager with graphics. In that case, the only thing to be done is drawing the flowsheet on the screen, selecting the various pieces of process-equipment from a menu and connecting these blocks by entering streams; most blocks have a number of icons to choose from. While building the flowsheet you must keep in mind what you want to have calculated; if you are not interested in an energy balance, most of the pumps are unnecessary, since Aspen does not take into account the pressure drop over a pipeline. It is also possible to give the desired outlet pressure after the pump as a specification on the previous block. Likewise, you can model a compressor as a heater with an increasing pressure over the apparatus, and a distillation column as a separation block with given composition and temperatures of the outgoing streams. When all blocks are placed, ModelManager asks for all required input, such as the equations of state to be used, the flows and compositions of the feed streams and the specifications for all blocks, e.g. the number of trays in a column, pressure drops over the various pieces of process-equipment and the reactions taking place in a reactor. When sufficient data is entered, ModelManager gives a prompt that the simulation can be executed and creates an input file to be used by the program itself; before the release of ModelManager these input files had to be written directly, thus requiring a reasonable knowledge of the language used by Aspen. A more detailed description on the use of ModelManager and Aspen is given in appendix 12, as well as all models and an example of an Aspen input-file.

12.3. Aspen's special features.

Aspen is one of the more advanced simulation packages on the market and offers quite a lot of special features. For one, the choice of the model for the equation of state to be used is to be made out of over 25 options, varying from the ideal gas law to the Kent-Eisenberg model which is to be selected when using alkanolamines to remove CO₂ from a stream. It is possible to give every piece of process-equipment its own thermodynamic model.

An other impressive option is the amount of data which can be obtained about the operation of a distillation column; Aspen automatically gives data regarding the vapour/liquid equilibrium, but can also, on request, give properties of the mixture on each individual tray such as the surface tension, the viscosity, the density and the interaction parameters.

If enough expertise on the use of the computer language Fortran is present, one can write the routine to be used in a user-defined block; this can be necessary if the apparatus to be modelled is not present in the standard list used by Aspen (see appendix 12), or if the results of the standard routine are not satisfactory.

Aspen is a user-friendly program, providing an on-screen help facility on most of the topics; for more difficult and acute problems there is the Aspen-Plus-Hotline where a team of experts is ready to solve them.

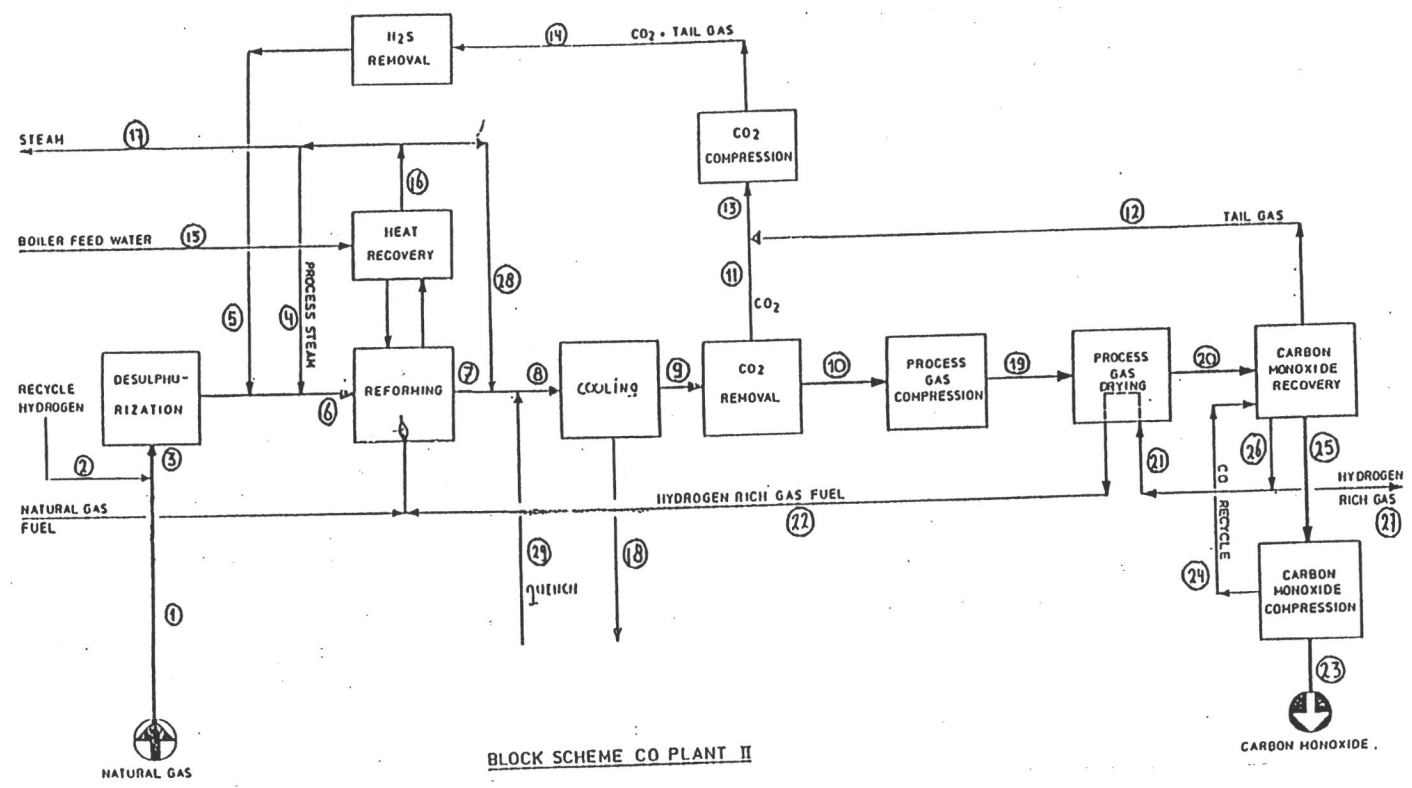
12.4. Problems and possibilities for improvement.

The problems encountered while working with Aspen can be divided into three major groups: problems with the host-computer ("the network"), difficulties understanding and using the program itself and problems occurring while running the simulation.

The problems with the host-computer were the most time-consuming and often occurring; most of them can be described to lack of memory on the host-computer. In a case like that, Aspen gives a prompt that there is not enough memory left to generate the output files and the simulation is terminated. In the worst case the lack of memory does not occur until the simulation is completed; Aspen then tells you to delete a number of other files and tries again to read in the results. If however the available memory is not enough, the entire program is terminated losing all data. When the network is used by someone else running more memory-consuming programs, Aspen will not run at all; there is no solution to this problem, however. Most of the other problems were solved when the amount of memory assigned to us was doubled. In this context I would like to point out the possibility of making a back-up file in which all relevant data is stored and which can be copied from the network to a floppy disk.

The inability to understand Aspen is caused by not having an up-to-date manual on how to use ModelManager. Most of the problems were solved eventually, but some remained. To name a few: it has not become clear how you can make a correction when you made a mistake in entering the name of a particular stream or block. Equally disturbing is the fact that numlock is inactive under ModelManager, even though you have got an enormous amount of data to enter. There is also no way of printing only a part of the entered flowsheet; when the simulated process is rather complicated this certainly is desirable.

Problems occurring when running or attempting to run a simulation are the following: sometimes, when using special routines for the equation of state, Aspen has to link with its library, which can take up to a quarter of an hour. There is, however, the possibility of saving these routines which we did not discover until recently. We already discussed the problems occurring when the network is used by others. Sometimes, Aspen refuses to read in the results, either by ignoring your command to do so or by pretending to be reading them in while doing nothing at all. Aspen would sometimes generate error messages nobody could understand, or present results which are physically impossible without giving as much as a warning. In this context I would like to point out the danger that a heat exchanger cools both streams (without change of phase), even though the enthalpy of one of them rises substantially. This has probably a lot to do with the unaccuracy of the enthalpy-calculating routines; this is the major reason for not being able to simulate processes with very complex heat recovery, such as the Coldbox.



BLOCK SCHEME CO PLANT II

fig. A2.1 : block scheme of CO plant II

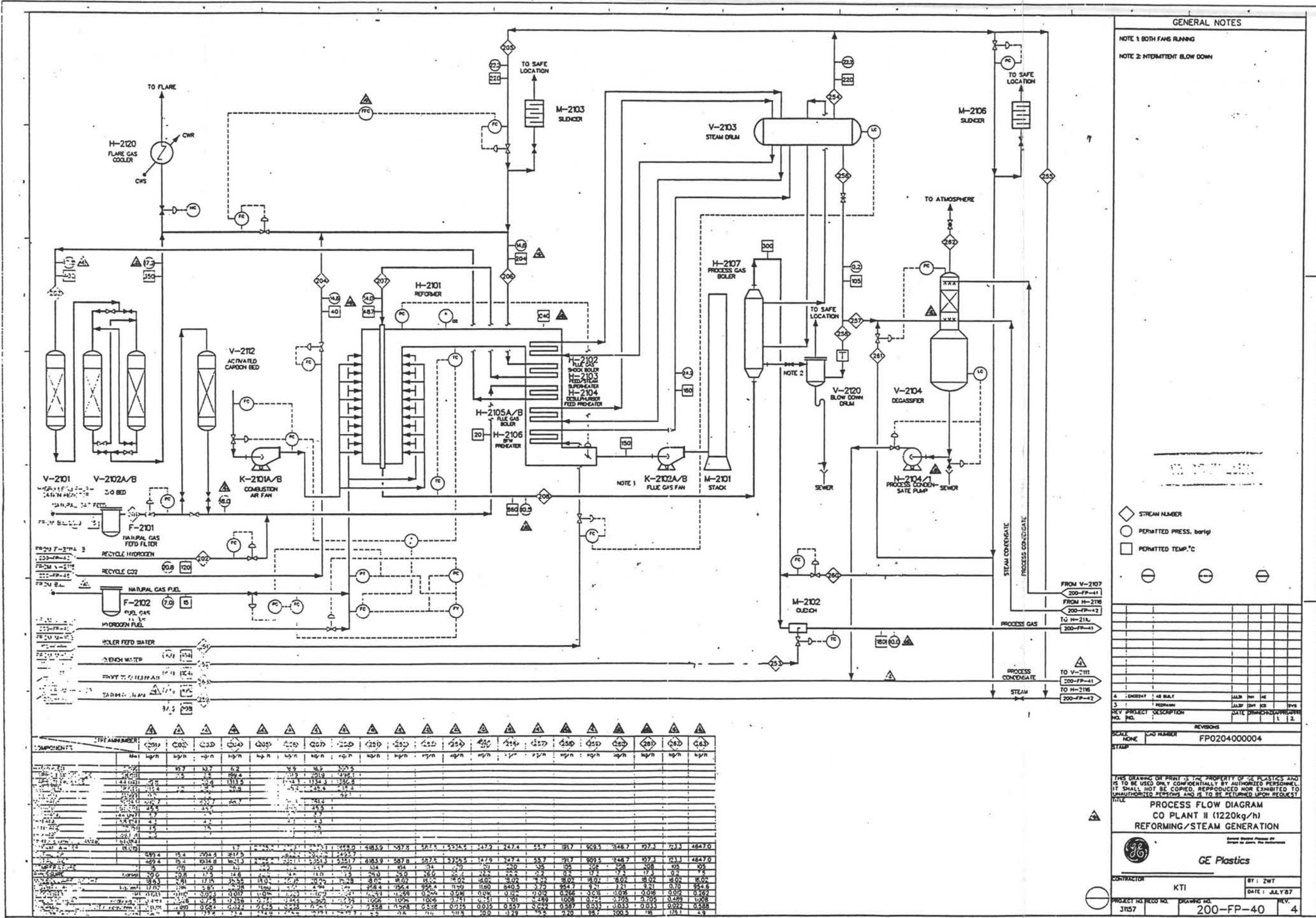
A2-1

Table A2.1 : design streams of CO plant II

Block Scheme		CO plant II		Components (kg/h)													
Stream number	Stream number	pressure [bar]	Temp [°C]	H ₂	CO	CO ₂	N ₂	O ₂	CH ₄	C ₂ H ₆	C ₃ H ₈	C ₄ H ₁₀	C ₅ H ₁₂	C ₆ H ₁₄	MEA	H ₂ O	total
1	201	20	15	-	-	20,8	213,4	0,2	692,7	45,5	8,7	4,3	1,5	2,3	-	-	989,4
2	202	20,8	120	10,7	2,5	-	2,2	-	-	-	-	-	-	-	-	-	15,4
3	203	17,5	400	10,7	2,5	20,8	215,7	0,2	692,7	45,5	8,7	4,3	1,5	2,3	-	-	1004,8
4	205	22,2	220	-	-	-	-	-	-	-	-	-	-	-	-	2725,2	2725,2
5	204	14,6	40	6,2	199,4	1313,5	29,8	-	68,7	-	-	-	-	-	-	3,7	1621,3
6	207	14,0	487	16,9	201,9	1334,3	245,4	-	761,4	45,5	8,7	4,3	1,5	2,3	-	2729,1	5351,3
7	208	10,5	860	300,5	1498,1	1380,6	245,4	-	69,1	-	-	-	-	-	-	1858,0	5351,7
8	209	10,0	180	300,5	1498,1	1380,6	245,4	-	69,1	-	-	-	-	-	-	4191,6	7786,3
9	211	9,4	40	300,5	1498,1	1380,6	245,4	-	69,1	-	-	-	-	-	-	31,4	3525,1
10	212	9,3	40	300,1	1496,0	0,5	245,1	-	69,0	-	-	-	-	-	-	27,9	2138,6
11	215	0,5	40	0,4	2,0	1313,5	0,3	-	0,1	-	-	-	-	-	-	28	1344,3
12	227	1,0	31	5,8	197,3	-	29,4	-	68,6	-	-	-	-	-	-	-	304,1
13	216	0,5	40	6,2	199,4	1313,5	29,8	-	68,7	-	-	-	-	-	-	28	1645,6
14	217	15,2	40	6,2	199,4	1313,5	29,8	-	68,7	-	-	-	-	-	-	3,7	1621,3
15	251	26,0	104	-	-	-	-	-	-	-	-	-	-	-	-	6183,9	6183,9
16	254	22,2	220	-	-	-	-	-	-	-	-	-	-	-	-	5936,5	5936,5
17	255	22,2	220	-	-	-	-	-	-	-	-	-	-	-	-	347,9	347,9
18	266	2,5	98	-	-	-	-	-	-	-	-	-	-	-	-	4261,2	4261,2
19	219	27,7	35	300,1	1496,0	0,5	245,1	-	69,0	-	-	-	-	-	-	7,8	2118,5
20	223	27,2	35	298,3	1487,1	-	243,7	-	68,6	-	-	-	-	-	-	-	2097,8
21	222	2,7	27	241,4	56,7	-	50,1	-	-	-	-	-	-	-	-	-	348,2
22	222 [energy flow]	2,0	40	243,2	65,6	0,5	51,5	-	0,4	-	-	-	-	-	-	7,8	369,0
23	226	8,6	35	-	1221,4	-	153,6	-	-	-	-	-	-	-	-	-	1375,0
24	225	8,6	35	-	238,6	-	390	-	-	-	-	-	-	-	-	-	268,7
25	224	1,0	31	-	1480,1	-	183,6	-	-	-	-	-	-	-	-	-	1643,7
26	228	3,3	25	292,5	68,3	-	60,7	-	-	-	-	-	-	-	-	-	421,5
27	229	2,0	24	511	11,7	-	10,6	-	-	-	-	-	-	-	-	-	73,4
28	260	17,3	208	-	-	-	-	-	-	-	-	-	-	-	-	1846,7	1846,7
29	253	26,0	104	-	-	-	-	-	-	-	-	-	-	-	-	581,8	581,8

A2-2

confidential



GENERAL NOTES
 NOTE 1 BOTH FANS RUNNING
 NOTE 2 INTERMITTENT BLOW DOWN

◇ STREAM NUMBER
 ○ PERMITTED PRESS. barg
 □ PERMITTED TEMP. °C

FROM V-2107	200-FP-41
FROM H-2106	200-FP-42
TO H-2101	200-FP-41
TO H-2106	200-FP-42

4	DESIGN	AS BUILT	JAN 87	ME	
3	DESIGN	AS BUILT	JAN 87	ME	

SCALE: NONE
 DRAWING NO: FP0204000004

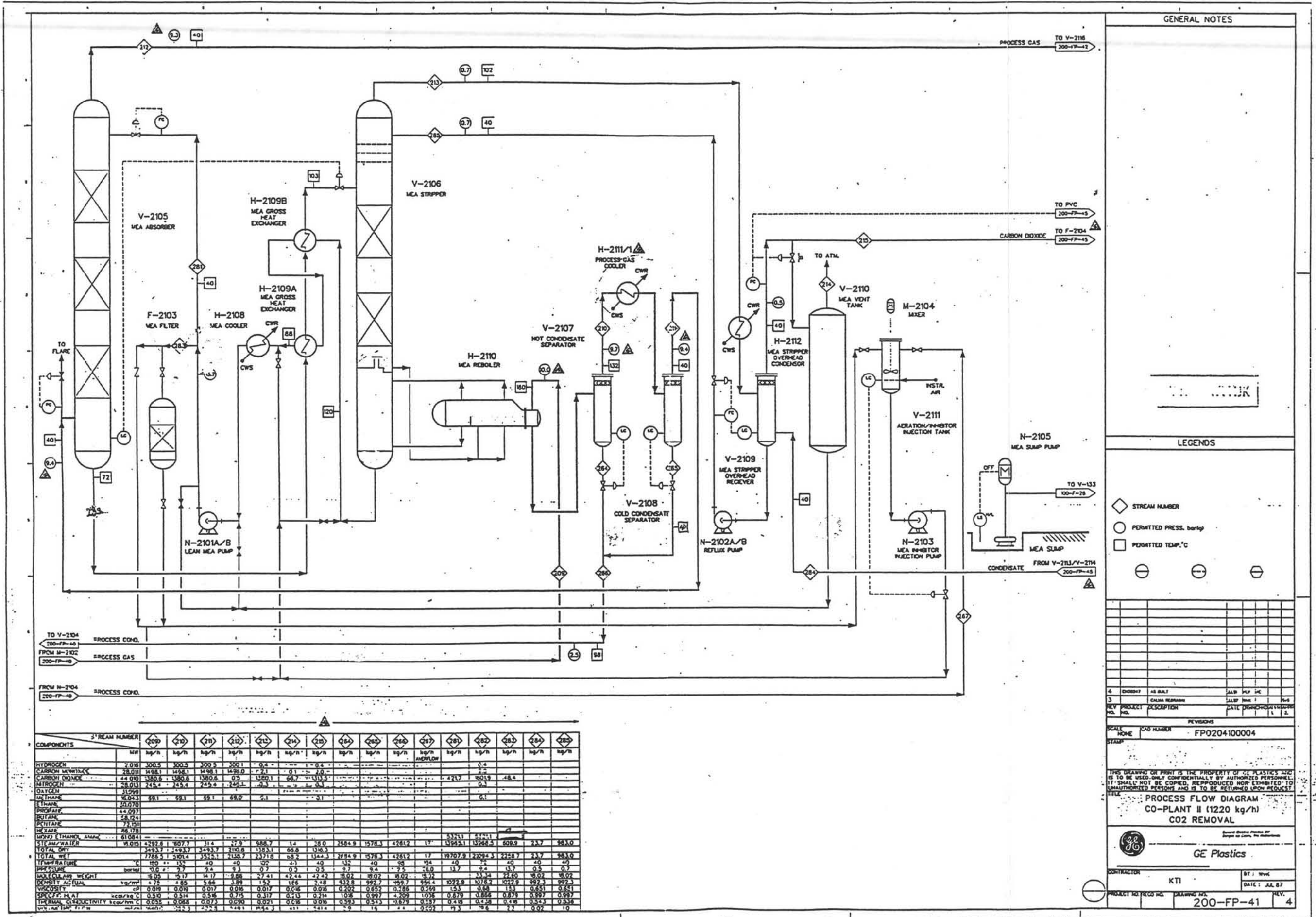
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PROCESS FLOW DIAGRAM
 CO PLANT II (1220kg/h)
 REFORMING/STEAM GENERATION

CONTRACTOR	KTJ	BY:	ZWT
DATE:	JULY 87		
PROJECT NO. / REV. NO.	3157	DRAWING NO.	200-FP-40
		REV.	4

COMPONENTS	STREAM NUMBER	CO ₂	CO	H ₂	CH ₄	N ₂	O ₂	Ar	HCN	CO ₂	CO	H ₂	CH ₄	N ₂	O ₂	Ar	HCN	CO ₂	CO	H ₂	CH ₄	N ₂	O ₂	Ar	HCN
...





GENERAL NOTES

LEGENDS

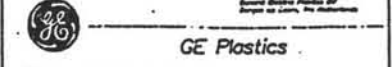
- ◇ STREAM NUMBER
- PERMITTED PRESS. bar(g)
- PERMITTED TEMP. °C

REV	PROJECT NO.	DESCRIPTION	DATE	DESIGNED BY	CHECKED BY
4	200-FP-41	AS BUILT		JLB	MLV
3	200-FP-41	CALM REPAIR		JLB	MLV

SCALE: NONE
 SHEET NUMBER: FP0204100004

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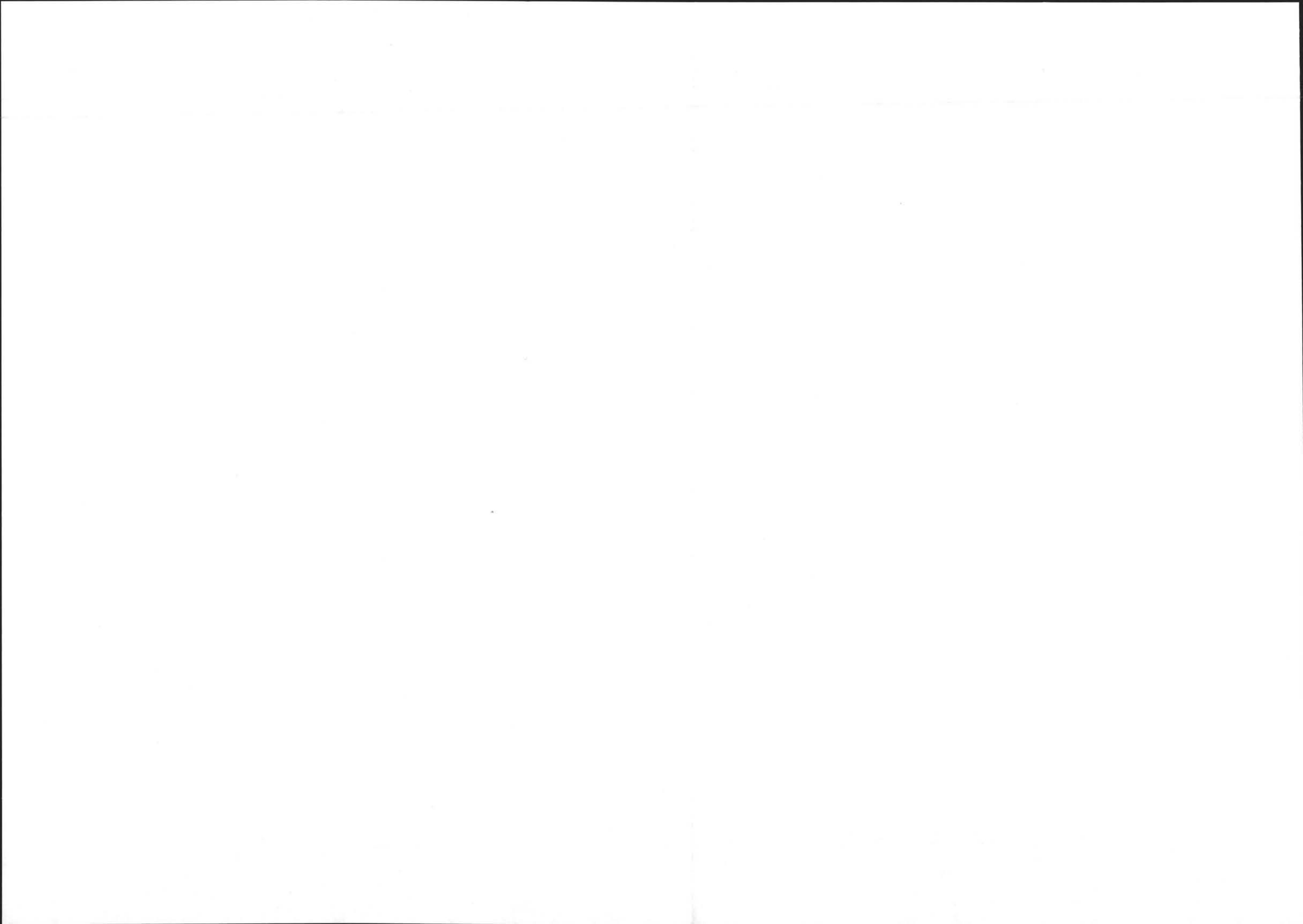
PROCESS FLOW DIAGRAM
 CO-PLANT II (1220 kg/h)
 CO2 REMOVAL

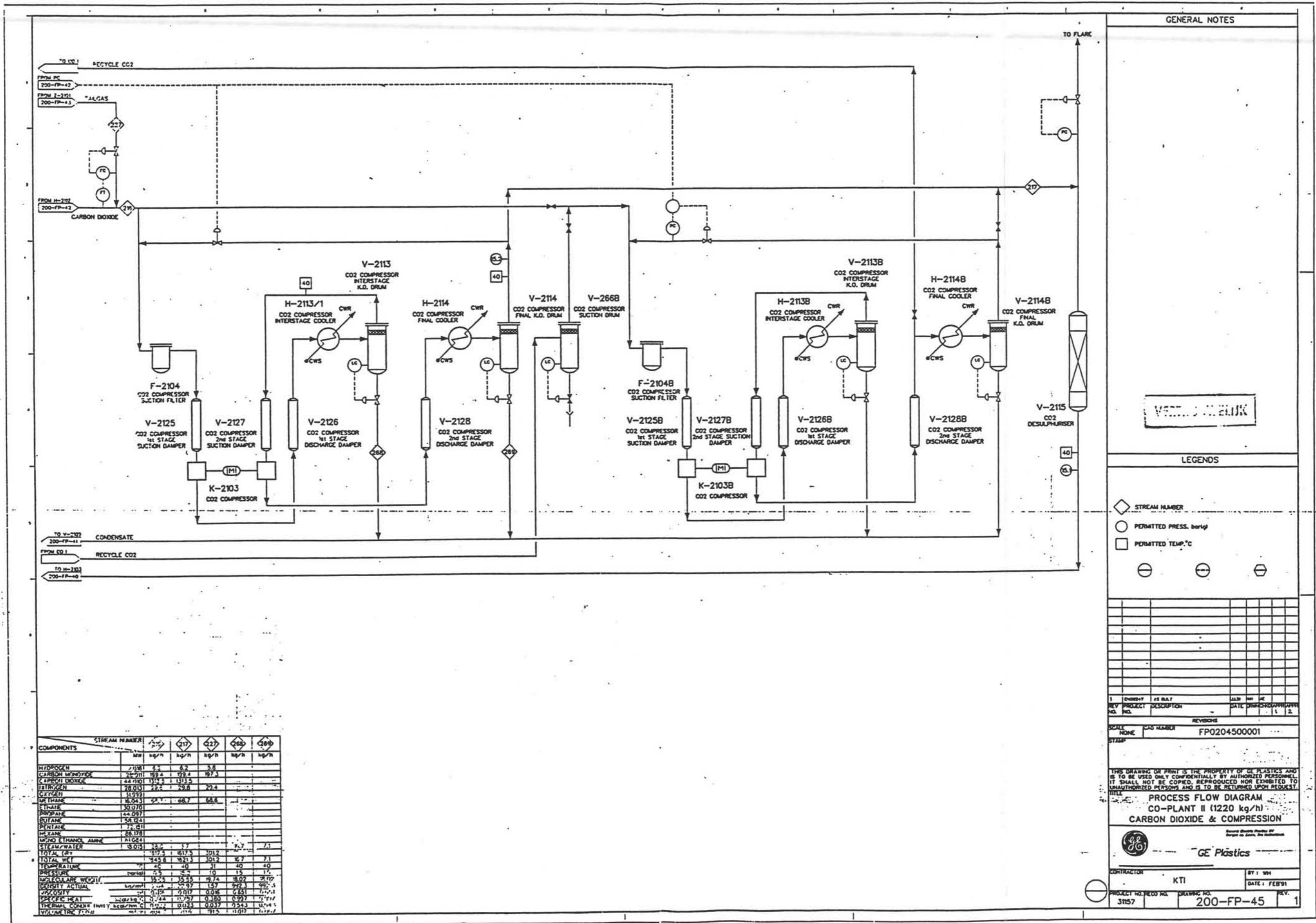


CONTRACTOR: KTI
 DATE: JUL 87

PROJECT NO. 200-FP-41
 DRAWING NO. 4

COMPONENTS	STREAM NUMBER																
	200	210	211	212	213	214	215	216	217	218	219	220	221	222	223		
HYDROGEN	2.018	300.5	300.5	300.5	300.1	0.4											
CARBON DIOXIDE	28.011	1498.1	1498.1	1498.1	1498.0	0.1											
NITROGEN	28.011	245.4	245.4	245.4	245.4	0.3											
OXYGEN	31.099																
METHANE	16.043	59.1	59.1	59.1	59.0	0.1											
ETHANE	20.070																
PROPANE	4.42971																
BUTANE	18.241																
PENTANE	72.151																
HEXANE	16.178																
METHYL THIOCARBONATE	61.084																
STEAM/WATER	16.018	4292.4	1807.7	31.4	27.9	588.7	1.4	18.0	2584.9	1578.3	4281.2	1.7	1325.1	1325.1	608.9	23.7	983.0
TOTAL DRY		3493.7	1493.7	3493.7	3493.7	1183.1	66.8	130.3									
TOTAL WET		7788.5	1501.4	3525.1	2158.7	2371.8	0.2	134.3	2884.9	1578.3	4281.2	1.7	19707.9	12294.3	2228.7	23.7	983.0
TEMPERATURE	°C	150	135	40	40	50	40	135	40	35	95	40	75	40	40	40	40
PH		10.0	10.0	10.0	10.0	10.0	10.0	10.0	10.0	10.0	10.0	10.0	10.0	10.0	10.0	10.0	10.0
WATER WEIGHT	kg/m ³	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000
DENSITY	kg/m ³	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000	1000
VISCOSITY	cp	0.019	0.019	0.017	0.016	0.017	0.016	0.026	0.202	0.252	0.286	0.259	1.53	0.68	1.53	0.651	0.651
SPECIFIC HEAT	kJ/kg°C	0.510	0.514	0.516	0.516	0.517	0.517	0.514	1.036	0.997	0.879	0.856	0.879	0.879	0.997	0.997	0.997
THERMAL CONDUCTIVITY	W/m°C	0.035	0.038	0.037	0.036	0.037	0.036	0.036	0.036	0.034	0.037	0.037	0.036	0.036	0.036	0.036	0.036
WATER TEMP. °C		150	135	40	40	50	40	135	40	35	95	40	75	40	40	40	40





GENERAL NOTES

V-2115

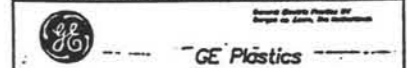
LEGENDS

- ◇ STREAM NUMBER
- PERMITTED PRESS. barg
- PERMITTED TEMP. °C

REV	PROJECT NO.	DESCRIPTION	DATE	BY	CHK
1					

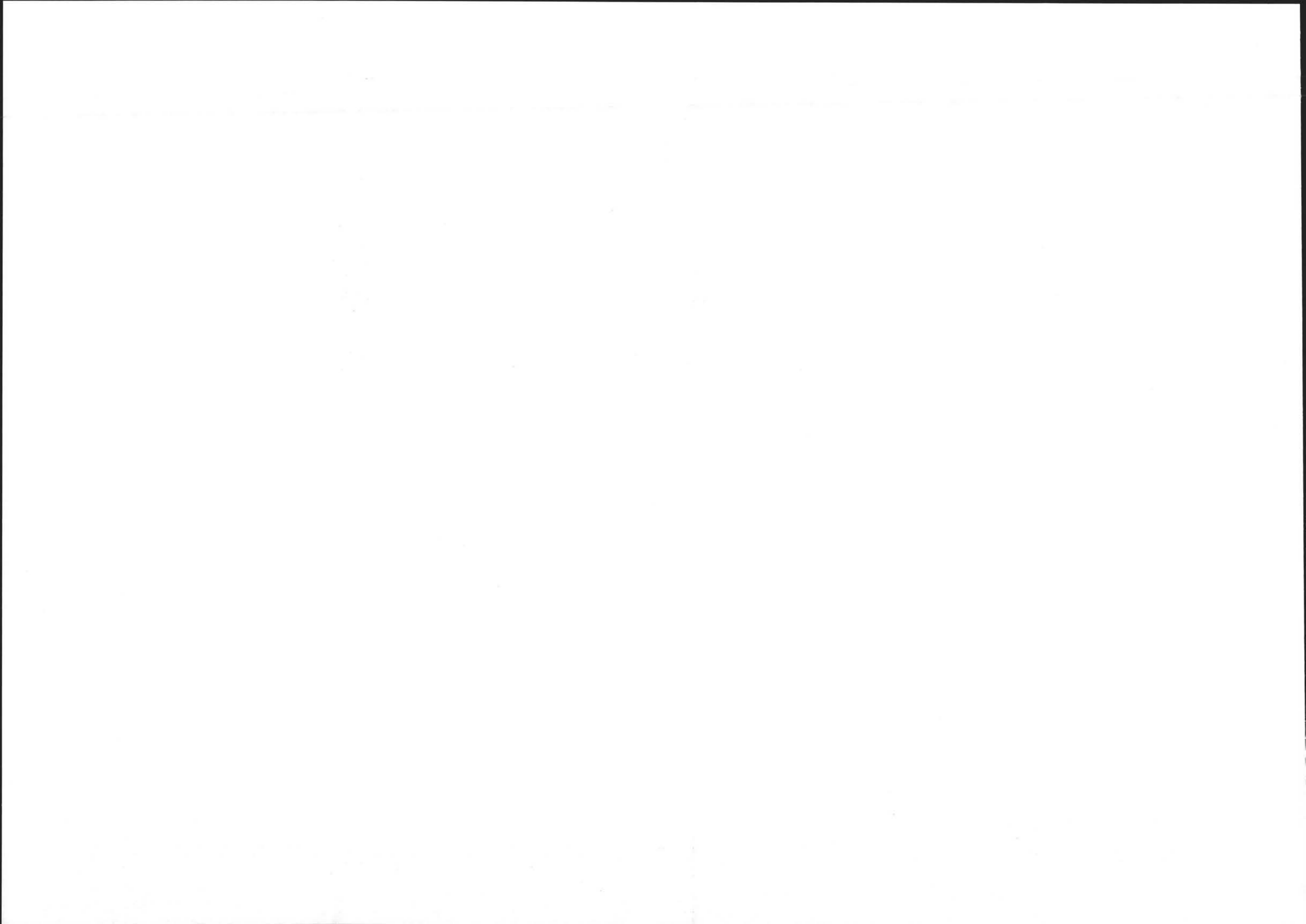
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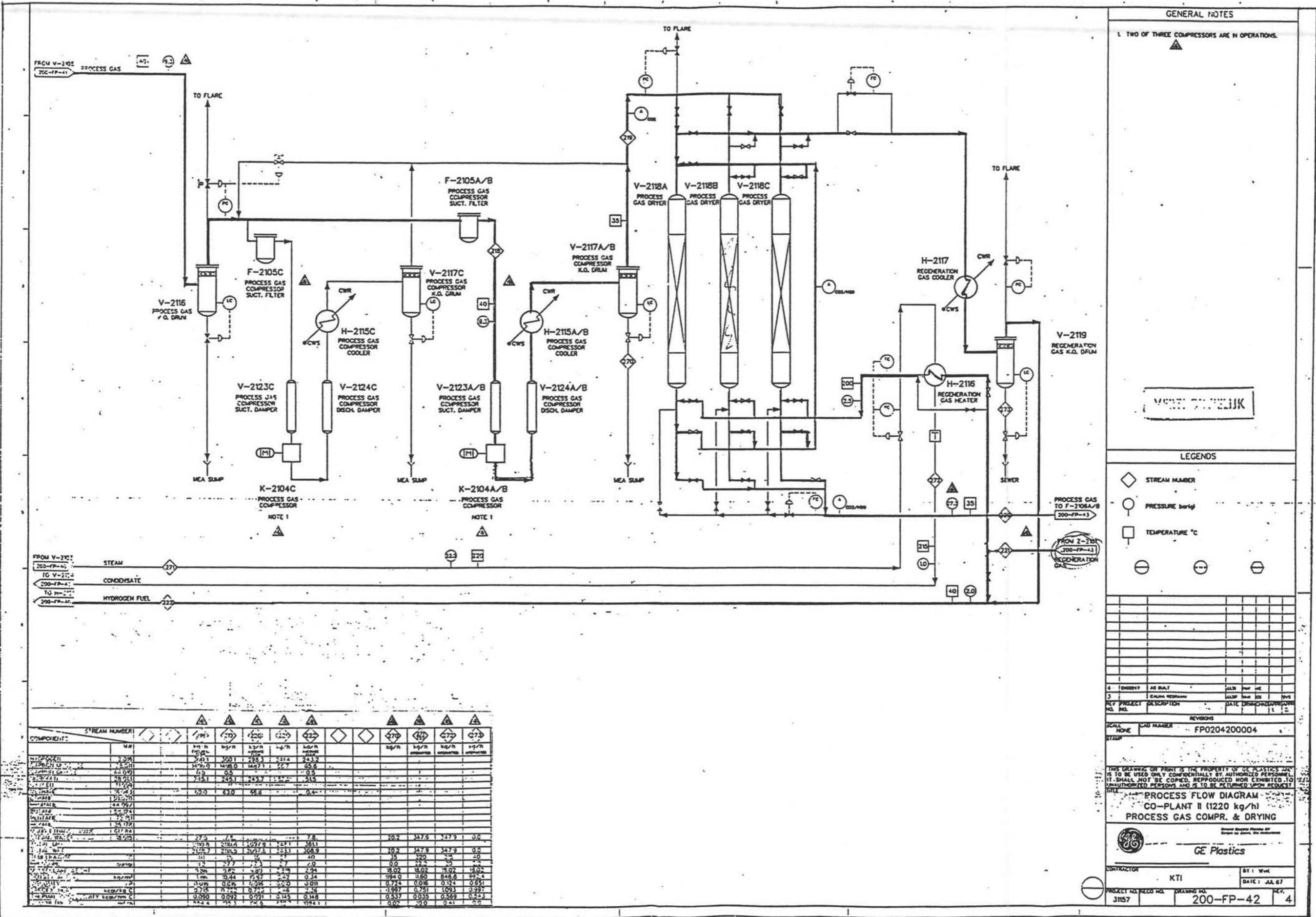
PROCESS FLOW DIAGRAM
CO-PLANT II (1220 kg/h)
CARBON DIOXIDE & COMPRESSION



CONTRACTOR: KTI
BY: WH
DATE: FEB/91
PROJECT NO. REC'D NO. DRAWING NO. REV. 1
3157 200-FP-45

COMPONENTS	STREAM NUMBER	Q17		Q27		Q68		Q88	
		kg/h	kg/h	kg/h	kg/h	kg/h	kg/h	kg/h	kg/h
HYDROGEN	2101	5.5	6.2	5.8					
CARBON MONOXIDE	2101	193.4	193.4	197.3					
CARBON DIOXIDE	44-1101	113.3	1313.5						
NITROGEN	28-011	22.2	29.8	29.4					
OXYGEN	31-011								
METHANE	16-041	25.1	68.7	64.8					
ETHANE	10-070								
PROPANE	44-097								
BUTANE	54-124								
PENTANE	17-121								
HEXANE	26-178								
WIND ETHANOL AMINE	11-024								
STEAM/WATER	3-021	25.5	1.7	2.7					
TOTAL (CV)		317.5	817.5	803.2					
TOTAL (W)		945.8	921.3	932.2					
TEMPERATURE		40	40	31	40	40			
PRESSURE		15	2	10	15	14			
MOLECULAR WEIGHT		16.5	35.25	37.4	30.2	28.7			
DENSITY ACTUAL		1.2	1.97	1.87	1.23	1.2			
VISCOSITY		0.09	0.019	0.08	0.55	0.4			
SPECIFIC HEAT		2.24	1.797	1.89	0.927	0.711			
THERMAL CAPACITY		0.72	0.123	0.137	0.543	0.415			
WILKINSON P.P.H.		1.1	1.14	1.15	1.017	1.017			





GENERAL NOTES

1. TWO OF THREE COMPRESSORS ARE IN OPERATIONS.

VENETIAN LIJK

LEGENDS

- ◇ STREAM NUMBER
- PRESSURE bar(g)
- TEMPERATURE °C

REV	DESCRIPTION	DATE	BY	CHK
1	ISSUED FOR AS BUILT	JUN 87	WVK	WVK
2	CHANGE REQUEST	JUN 87	WVK	WVK
3	PROJECT DESCRIPTION	DATE	BY	CHK

SCALE: NONE
JOB NUMBER: FP0204200004

STAMP

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PROCESS FLOW DIAGRAM

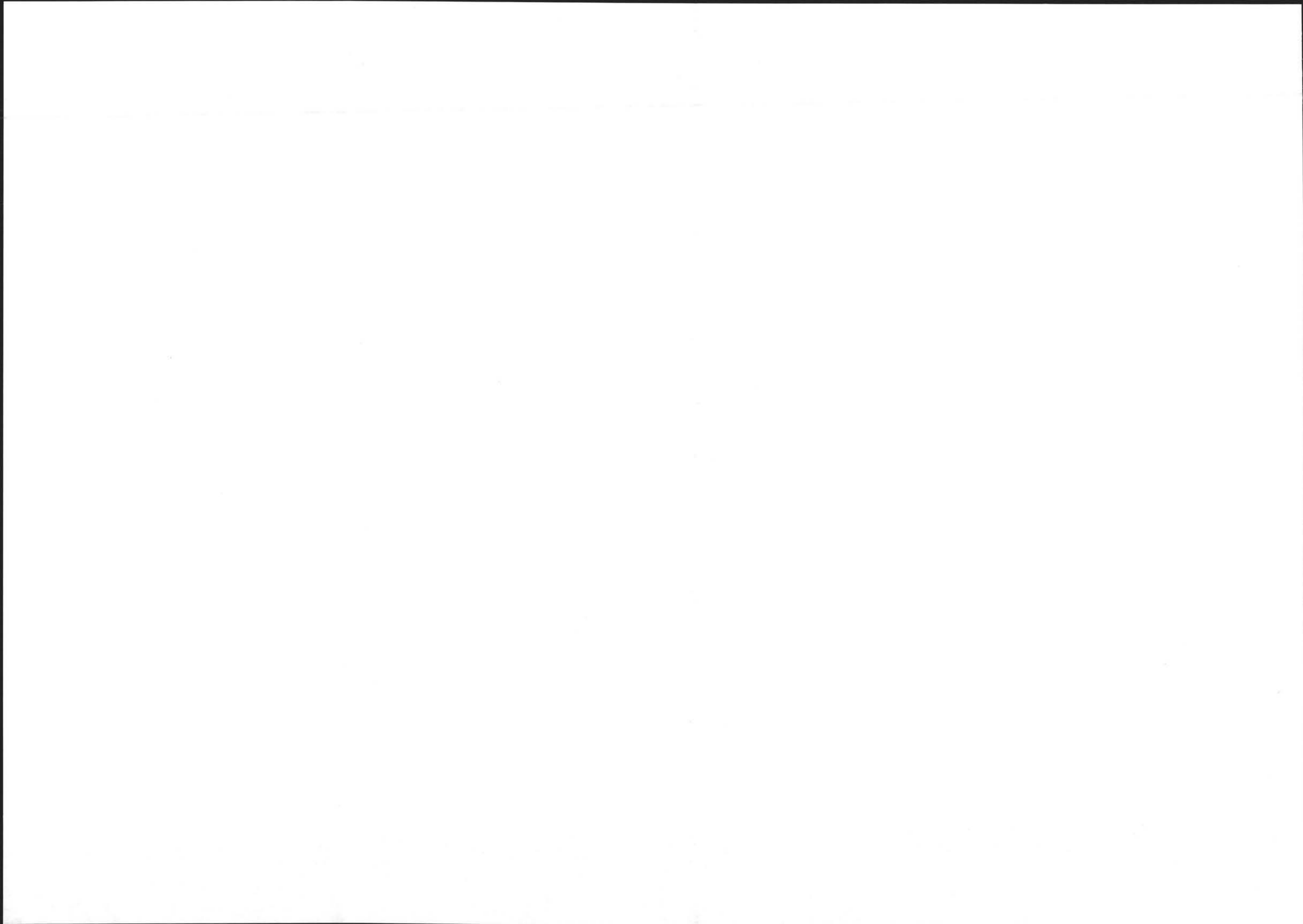
CO-PLANT II (1220 kg/h)

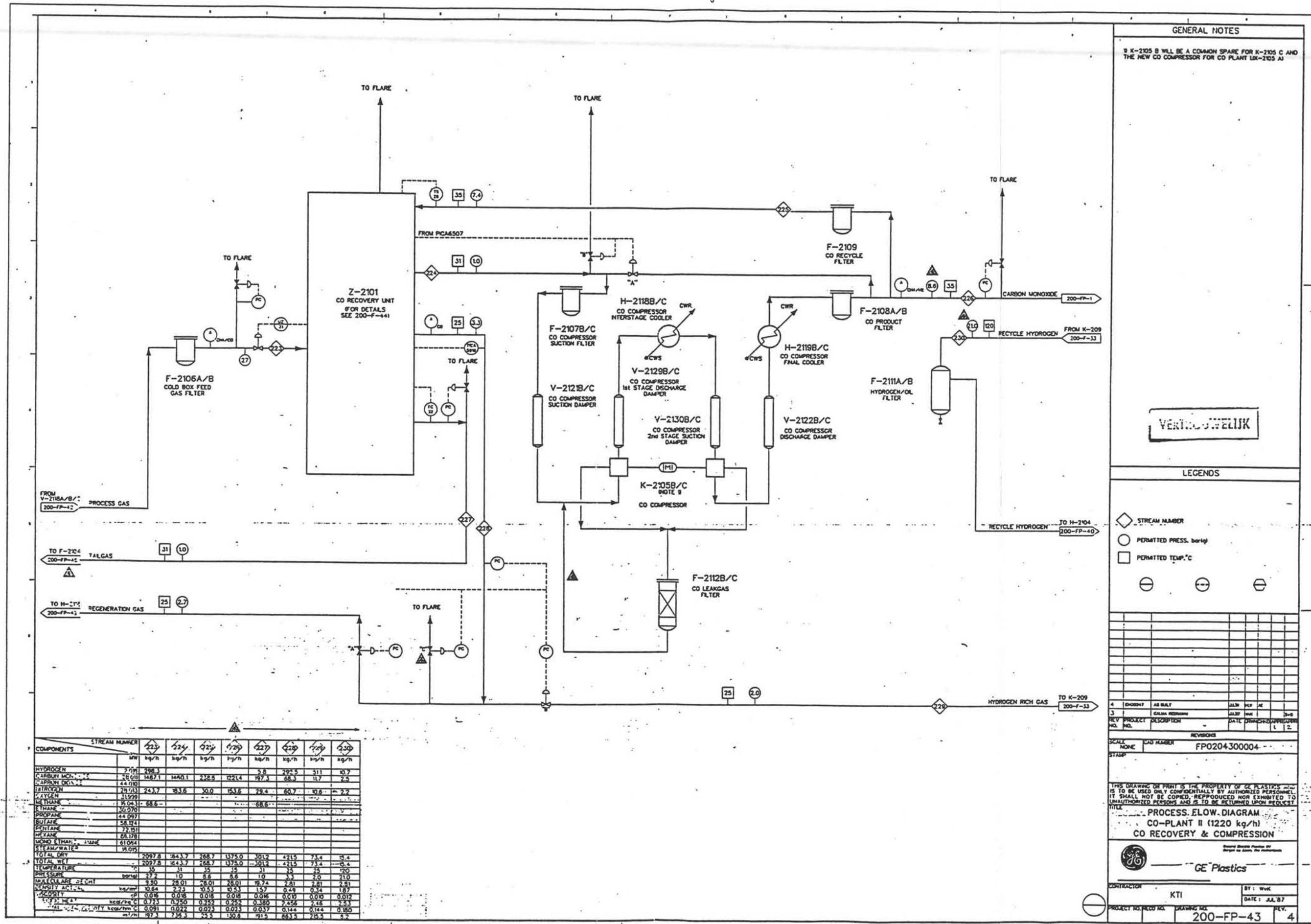
PROCESS GAS COMPR. & DRYING

GE Plastics

CONTRACTOR: KTI
BY: WVK
DATE: JUN 87

PROJECT NO./RECORD NO.: 3157
DRAWING NO.: 200-FP-42
4





GENERAL NOTES
 K-2105 B WILL BE A COMMON SPARE FOR K-2105 C AND THE NEW CO COMPRESSOR FOR CO PLANT LX-2105 A1

VERTINGWELIJK

LEGENDS

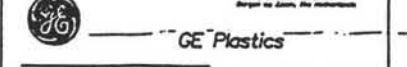
- ◇ STREAM NUMBER
- PERMITTED PRESS. bar(g)
- PERMITTED TEMP. °C

4	SHOWN	AS BUILT	JAN	1987	JE	
3		CHINA REVISION	JAN	1987	JE	2nd
REV	PROJECT NO.	DESCRIPTION	DATE	BY	APPROVED	

SCALE	NONE	DRAWING NO.	FP0204300004
STAMP			

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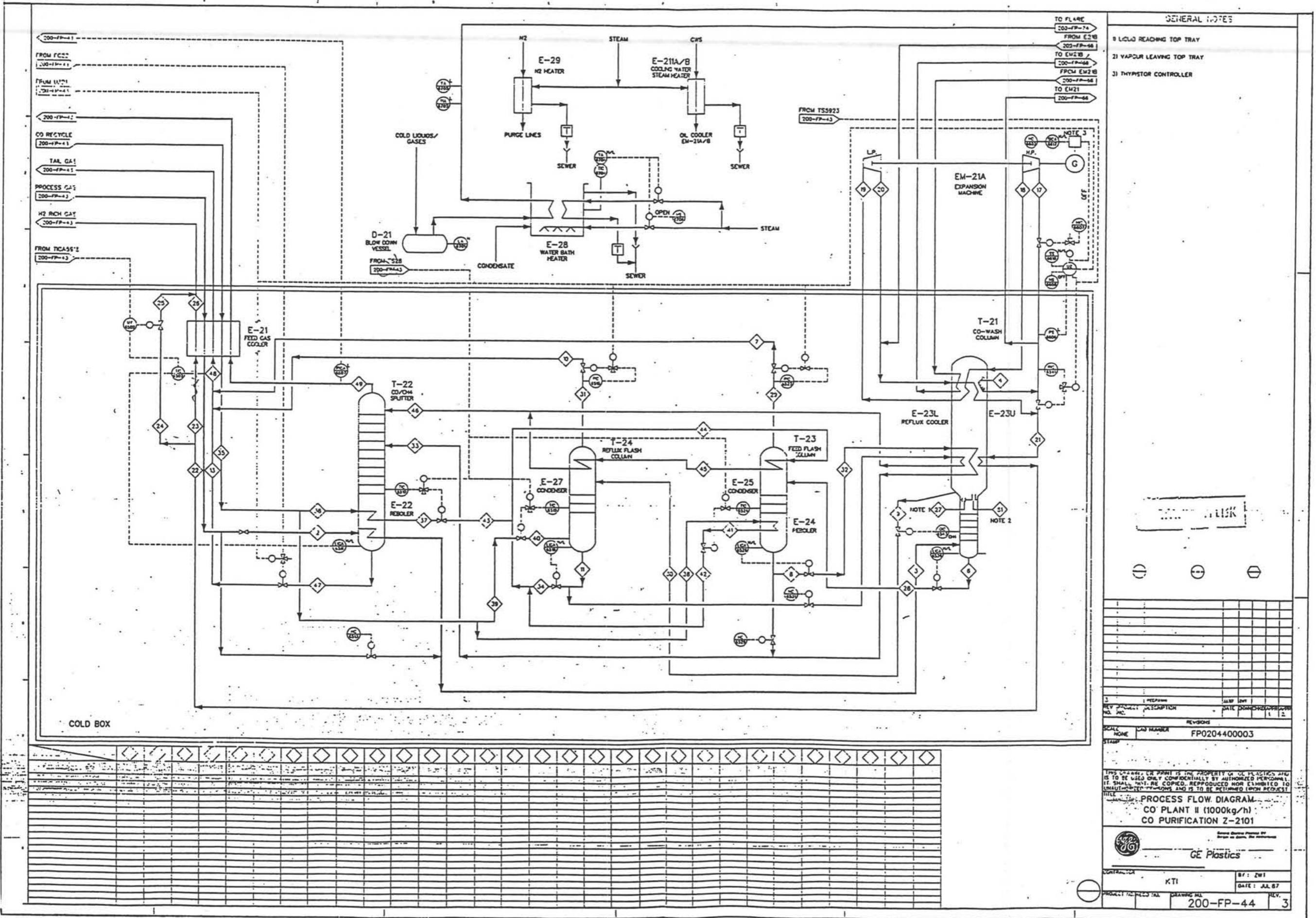
PROCESS FLOW DIAGRAM
 CO-PLANT II (1220 kg/h)
 CO RECOVERY & COMPRESSION

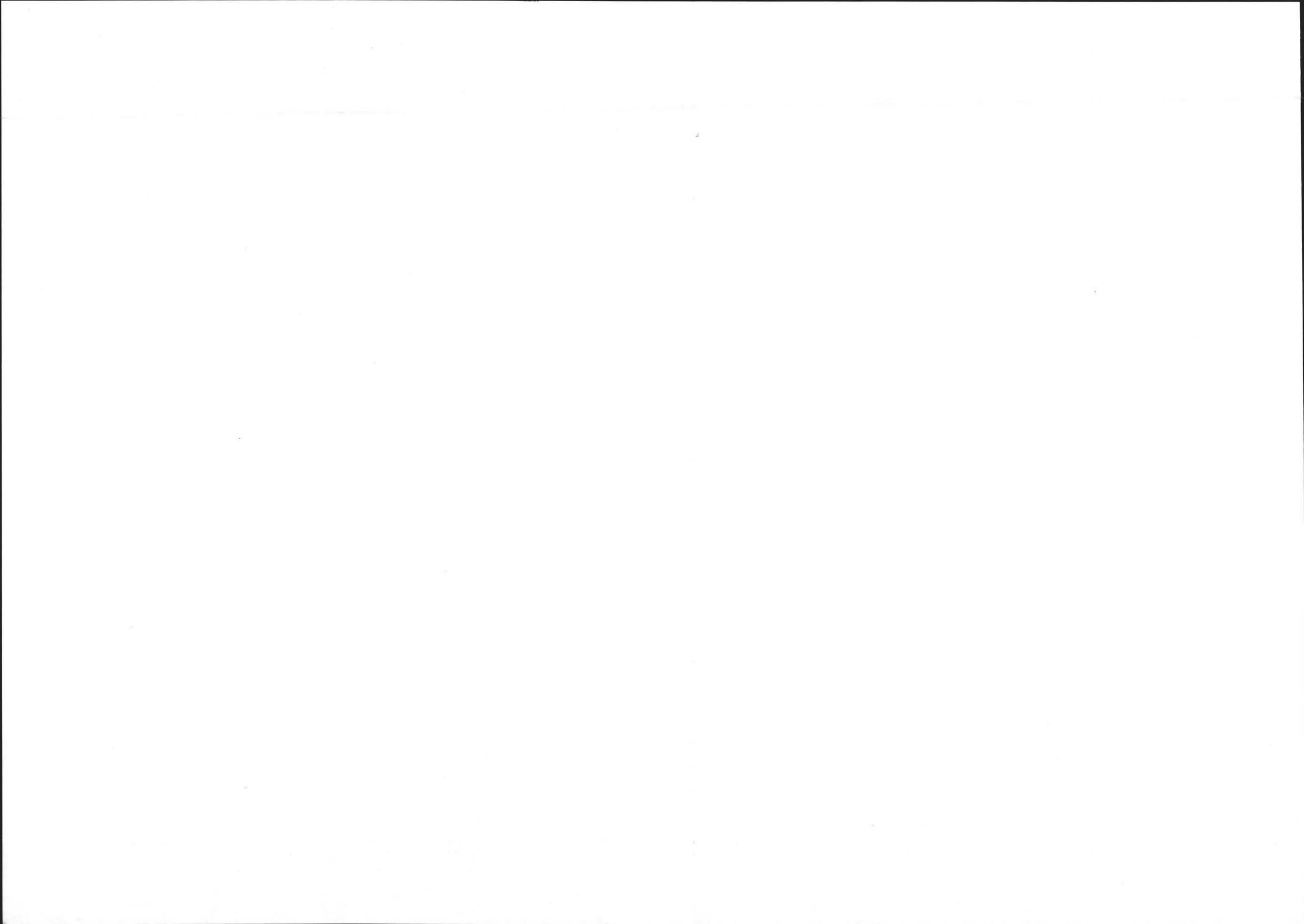


CONTRACTOR	KTJ	BY: WJK
PROJECT NO./REV. NO.	DRAWING NO.	DATE: AA.87
	200-FP-43	
		REV. 4

COMPONENTS	STREAM NUMBER	Flow Rates							
		223	224	225	226	227	228	229	230
	MW	kg/h	kg/h	kg/h	kg/h	kg/h	kg/h	kg/h	kg/h
HYDROGEN	770	288.3							
CARBON MONOXIDE	22	1487.1	1440.1	228.5	221.4	5.8	292.5	51.1	10.7
ETHANE	44	1170							
ETHYLENE	28	253.7	83.6	30.0	153.6	79.4	60.7	0.8	2.2
ETHANE	31	31599							
METHANE	8	68.6							
ETHANE	32								
PROPANE	44	997							
BUTANE	58	124							
PENTANE	72	51							
HEXANE	86	138							
HEPTANE	100	24							
STEAM/WATER	114	939							
TOTAL DRY		2097.8	2643.7	268.7	1375.0	301.2	421.5	73.4	14.4
TOTAL WET		2097.8	2643.7	268.7	1375.0	301.2	421.5	73.4	14.4
TEMPERATURE		35	31	35	35	31	35	35	30
PRESSURE		27.2	10	8.8	8.8	10	3.3	2.0	21.0
MOLE FRACTION		9.90	28.01	28.01	28.01	98.74	7.81	2.81	2.81
DENSITY		10.64	2.23	10.53	10.53	1.57	0.49	0.14	1.87
VELOCITY		0.016	0.018	0.018	0.018	0.018	0.018	0.018	0.012
WET WEIGHT		0.233	0.250	0.252	0.252	0.380	2.456	2.46	2.53
WET WEIGHT		0.093	0.092	0.092	0.092	0.037	0.144	0.144	0.180
WET WEIGHT		97.3	736.3	52.5	150.8	91.5	861.5	705.3	9.2







Appendix 3.1 Reformer heat duty calculations.

$$T1 := 487 + 273.15$$

$$p1 := 15$$

$$R := 8.314$$

$$T2 := 860 + 273.15$$

$$p2 := 11$$

$$Tst := 25 + 273.15$$

$$Tam := \frac{T1 + T2}{2}$$

$$M := \begin{bmatrix} 2.016 \\ 28.011 \\ 44.010 \\ 28.013 \\ 16.043 \\ 30.070 \\ 44.097 \\ 58.124 \\ 72.151 \\ 86.178 \\ 18.015 \end{bmatrix}$$

$$y_{in} := \begin{bmatrix} 8.38 \\ 7.21 \\ 30.32 \\ 8.76 \\ 47.46 \\ 1.51 \\ 0.20 \\ 0.07 \\ 0.02 \\ 0.03 \\ 151.49 \end{bmatrix}$$

$$y_{uit} := \begin{bmatrix} 149.06 \\ 53.48 \\ 31.37 \\ 8.76 \\ 4.31 \\ 0 \\ 0 \\ 0 \\ 0 \\ 0 \\ 103.13 \end{bmatrix}$$

$$y_{totin} := \sum y_{in}$$

$$y_{totuit} := \sum y_{uit}$$

$$y1 := \frac{y_{in}}{y_{totin}}$$

$$y2 := \frac{y_{uit}}{y_{totuit}}$$

$$T_{ci} := \begin{bmatrix} 33.2 \\ 132.9 \\ 304.2 \\ 126.2 \\ 190.6 \\ 305.4 \\ 369.8 \\ 425.2 \\ 469.6 \\ 507.4 \\ 647.3 \end{bmatrix}$$

$$w_i := \begin{bmatrix} -.22 \\ .049 \\ .225 \\ .040 \\ .008 \\ .098 \\ .152 \\ .193 \\ .251 \\ .296 \\ .344 \end{bmatrix}$$

$$p_{ci} := \begin{bmatrix} 13.0 \\ 35.0 \\ 73.8 \\ 33.9 \\ 46.0 \\ 48.8 \\ 42.5 \\ 38.0 \\ 33.7 \\ 29.7 \\ 220.5 \end{bmatrix}$$

	3.249	0.422 · 10 ⁻³	0	0.083 · 10 ⁵
	3.376	0.557 · 10 ⁻³	0	-0.031 · 10 ⁵
	5.457	1.045 · 10 ⁻³	0	-1.157 · 10 ⁵
	3.280	0.593 · 10 ⁻³	0	0.040 · 10 ⁵
	1.702	9.081 · 10 ⁻³	-2.164 · 10 ⁻⁶	0
Cpcoef :=	1.131	19.225 · 10 ⁻³	-5.561 · 10 ⁻⁶	0
	1.213	28.785 · 10 ⁻³	-8.824 · 10 ⁻⁶	0
	1.935	36.915 · 10 ⁻³	-11.402 · 10 ⁻⁶	0
	2.464	2.464 · 10 ⁻³	45.351 · 10 ⁻⁶	0
	3.025	53.722 · 10 ⁻³	-16.791 · 10 ⁻⁶	0
	3.470	1.450 · 10 ⁻³	0	0.212 · 10 ⁵

$$E := \begin{bmatrix} 1 \\ Tam \\ 2 \\ 4 \cdot Tam - T1 \cdot T2 \\ 3 \\ 1 \\ T1 \cdot T2 \end{bmatrix}$$

$$Cplos := R \cdot Cpcoef \cdot E$$

Cplos =	30.41362781
	32.42198198
	52.47736443
	31.97570426
	69.29036111
	118.74389363
	170.04141815
	220.57482461
	382.14034121
	321.24517664
	40.46635497

$$cpin := y1 \cdot Cplos$$

$$cput := y2 \cdot Cplos$$

Eerst ga ik ethaan en hoger (487gr C) laten reageren tot CH4

$$\text{yeerst} := \begin{bmatrix} \text{yin} \\ 5 \\ \text{yin} \\ 6 \\ \text{yin} \\ 7 \\ \text{yin} \\ 8 \\ \text{yin} \\ 9 \end{bmatrix} \quad \text{Tceerst} := \begin{bmatrix} \text{Tci} \\ 5 \\ \text{Tci} \\ 6 \\ \text{Tci} \\ 7 \\ \text{Tci} \\ 8 \\ \text{Tci} \\ 9 \end{bmatrix} \quad \text{weerst} := \begin{bmatrix} \text{wi} \\ 5 \\ \text{wi} \\ 6 \\ \text{wi} \\ 7 \\ \text{wi} \\ 8 \\ \text{wi} \\ 9 \end{bmatrix} \quad \text{pceerst} := \begin{bmatrix} \text{pci} \\ 5 \\ \text{pci} \\ 6 \\ \text{pci} \\ 7 \\ \text{pci} \\ 8 \\ \text{pci} \\ 9 \end{bmatrix}$$

$$\text{ytoteerst} := \Sigma \text{yeerst}$$

$$\text{yfrac} := \frac{\text{yeerst}}{\text{ytoteerst}}$$

$$\text{Tpc1} := \text{yfrac}^T \cdot \text{Tceerst}$$

$$\text{ppc1} := \text{yfrac}^T \cdot \text{pceerst}$$

$$\text{wi} := \text{yfrac}^T \cdot \text{weerst}$$

$$\text{ppr1} := \frac{\text{p1}}{\text{ppc1} + 0}$$

$$\text{Tpr1} := \frac{\text{T1}}{\text{Tpc1} + 0}$$

$$\text{B01} := 0.083 - \frac{0.422}{1.6 \cdot \text{Tpr1}}$$

$$\text{dB01} := \frac{0.675}{2.6 \cdot \text{Tpr1}}$$

$$\text{B11} := 0.139 - \frac{0.172}{4.2 \cdot \text{Tpr1}}$$

$$\text{dB11} := \frac{0.722}{5.2 \cdot \text{Tpr1}}$$

$$\text{Hr1} := \text{ytoteerst} \cdot \text{R} \cdot \text{Tpc1} \cdot \text{ppr1} \cdot (\text{B01} - \text{Tpr1} \cdot \text{dB01} + \text{wi} \cdot (\text{B11} - \text{Tpr1} \cdot \text{dB11}))$$

id --> 25 gr C

$$\begin{aligned}
 & \left[\begin{array}{c} \text{Cplos} \\ 5 \\ \text{Cplos} \\ 6 \\ \text{Cplos} \\ 7 \\ \text{Cplos} \\ 8 \\ \text{Cplos} \\ 9 \end{array} \right] & \text{ytoteerst} & := \sum \text{yeerst} \\
 \text{cpeerst} & := & \text{yfrac} & := \frac{\text{yeerst}}{\text{ytoteerst}} \\
 & & \text{cp} & := \text{yfrac} \cdot \text{cpeerst} \quad \text{cp} = 134.44369555
 \end{aligned}$$

$$\delta H_{\text{react}} := \text{ytoteerst} \cdot \text{cp} \cdot (298.15 - T1)$$

$$\begin{aligned}
 \text{Hr} & := \left[\begin{array}{c} -65.220 \\ -118.880 \\ -172.290 \\ -225.840 \\ -280.200 \end{array} \right] & j & := 0 \dots 4 \\
 \delta H_{\text{r}} & := \text{yeerst} \cdot \text{Hr} & \text{yCH4} & := \sum_j (j + 2) \cdot \text{yeerst}_j
 \end{aligned}$$

Nu is er alleen CH4 dus deze weer opwarmen tot 487 gr C

$$\delta H_{\text{prod}} := \text{yCH4} \cdot \text{Cplos}_4 \cdot (T1 - 298.15)$$

Even reeel gas van maken :

$$\text{Tr} := \frac{T1}{\text{Tci}_4} \quad \text{pr} := \frac{p1}{\text{pci}_4}$$

$$\text{B01} := 0.083 - \frac{0.422}{\text{Tr}} \quad \text{B11} := 0.139 - \frac{0.172}{\text{Tr}}$$

$$\text{dB01} := \frac{0.675}{\text{Tr}} \quad \text{dB11} := \frac{0.722}{\text{Tr}}$$

$$Hr2 := y_{CH4} \cdot R \cdot T_{ci} \cdot \ln \left[\frac{B_{01} - Tr \cdot dB_{01} + \omega_1 \cdot (B_{11} - Tr \cdot dB_{11})}{4} \right]$$

$$\delta H_{reactiel} := Hr1 + Hr2 + \delta Hr + \delta H_{prod} + \delta H_{react}$$

=====

Nu de hoofdreacties

$$y1 := \begin{bmatrix} yin \\ 0 \\ yin \\ 1 \\ yin \\ 2 \\ yin \\ 3 \\ [yin + y_{CH4}] \\ 4 \\ yin \\ 10 \end{bmatrix} \quad T_{ci} := \begin{bmatrix} T_{ci} \\ 0 \\ T_{ci} \\ 1 \\ T_{ci} \\ 2 \\ T_{ci} \\ 3 \\ T_{ci} \\ 4 \\ T_{ci} \\ 10 \end{bmatrix} \quad p_{ci} := \begin{bmatrix} p_{ci} \\ 0 \\ p_{ci} \\ 1 \\ p_{ci} \\ 2 \\ p_{ci} \\ 3 \\ p_{ci} \\ 4 \\ p_{ci} \\ 10 \end{bmatrix} \quad \omega_i := \begin{bmatrix} \omega_i \\ 0 \\ \omega_i \\ 1 \\ \omega_i \\ 2 \\ \omega_i \\ 3 \\ \omega_i \\ 4 \\ \omega_i \\ 10 \end{bmatrix}$$

$$y_{tot1} := \sum y1$$

$$y_{fract1} := \frac{y1}{y_{tot1}}$$

$$T_{pc1} := y_{fract1}^T \cdot T_{ci}$$

$$p_{pc1} := y_{fract1}^T \cdot p_{ci}$$

$$\omega_1 := y_{fract1}^T \cdot \omega_i$$

$$p_{pr1} := \frac{p1}{p_{pc1}} \quad 0$$

$$T_{pr1} := \frac{T1}{T_{pc1}} \quad 0$$

$$y2 := \begin{bmatrix} yuit \\ 0 \\ yuit \\ 1 \\ yuit \\ 2 \\ yuit \\ 3 \\ yuit \\ 4 \\ yuit \\ 10 \end{bmatrix}$$

$$y_{tot2} := \sum y2$$

$$y_{fract2} := \frac{y2}{y_{tot2}}$$

confidential

$$B01 := 0.083 - \frac{0.422}{1.6 \cdot Tpr1}$$

$$dB01 := \frac{0.675}{2.6 \cdot Tpr1}$$

$$B11 := 0.139 - \frac{0.172}{4.2 \cdot Tpr1}$$

$$dB11 := \frac{0.722}{5.2 \cdot Tpr1}$$

$$Hr1 := ytot1 \cdot R \cdot Tpc1 \cdot ppr1 \cdot (B01 - Tpr1 \cdot dB01 + \omega1 \cdot (B11 - Tpr1 \cdot dB11))$$

$$Tpc2 := yfract2^T \cdot Tci$$

$$ppc2 := yfract2^T \cdot pci$$

$$Tpr2 := \frac{T2}{Tpc2}$$

$$\omega2 := yfract2^T \cdot \omega_i$$

$$ppr2 := \frac{p2}{ppc2}$$

$$B02 := 0.083 - \frac{0.422}{1.6 \cdot Tpr2}$$

$$dB02 := \frac{0.675}{2.6 \cdot Tpr2}$$

$$B12 := 0.139 - \frac{0.172}{4.2 \cdot Tpr2}$$

$$dB12 := \frac{0.722}{5.2 \cdot Tpr2}$$

$$Hr2 := R \cdot ytot2 \cdot Tpc2 \cdot ppr2 \cdot (B02 - Tpr2 \cdot dB02 + \omega2 \cdot (B12 - Tpr2 \cdot dB12))$$

nu voor ideaal mengsel van 487 naar 25 gr C


```

      Cplos
      0
      Cplos
      1
      Cplos
      2
cplos := Cplos
      3
      Cplos
      4
      Cplos
      5]
      cp1 := yfract1T·cplos
      cp2 := yfract2T·cplos

```

$\delta H_{\text{reactant}} := y_{\text{tot1}} \cdot cp1 \cdot (298.15 - T1)$

$\delta H_{\text{prod}} := y_{\text{tot2}} \cdot cp2 \cdot (T2 - 298.15)$

X := 47.46

Y := 1.07

```

yout := yin + X - Y
  1   1
yout := yin + Y
  2   2
yout := yin
  3   3
yout := yin - X - Y
  10  10
yout := yin + 3·X + Y
  0   0
      yout := yCH4 - X
      4

```

```

yout =
      151.83
      53.6
      31.39
      8.76
      -43.28
      0
      0
      0
      0
      0
      102.96

```



$$\delta\text{Hst1} := 205813 \quad \delta\text{Hst2} := -41166$$

$$\delta\text{Hreactie} := X \cdot \delta\text{Hst1} + Y \cdot \delta\text{Hst2}$$

$$\delta\text{Hreactie} = 9.72383736 \cdot 10^6$$

$$\delta\text{Htotaal} := \delta\text{Hreactie} + \delta\text{Hprod} + \delta\text{Hreactant} + \text{Hr1} + \text{Hr2}$$

$$\delta\text{Htotaal} = 1.59444047 \cdot 10^7 \quad \text{kJ/h}$$

kWatt

$$Q := \delta\text{Htotaal} + \delta\text{Hreactie1}$$

$$Q_{\text{totaal}} := \frac{Q}{3600}$$

$$Q_{\text{totaal}} = 4.43445582 \cdot 10^3$$

Appendix 3.2. Prediction of the streams and calculation of the equilibrium constants.

To predict the composition of the flows, the following equations were solved with Mercury.

$$T =$$

$$T_{app3} =$$

$$T_{app4} =$$

$$P =$$

$$m =$$

$$H_{2f} = m^*$$

$$CO_f = m^*$$

$$CO_{2f} = m^*$$

$$N_{2f} = m^*$$

$$CH_4 = m^*$$

$$H_2O_f = m^*$$

$$H_2O_s =$$

$$K_3 = \exp\left[-\frac{27464}{T - T_{app3}} + 30.707\right]$$

$$K_4 = \exp\left[\frac{4578}{T - T_{app4}} - 4.33\right]$$

$$K_4 = \exp\left[\frac{4084}{T} - 3.765\right]$$

$$K_3 = \frac{P_{CO} * P_{H_2}^3}{P_{CH_4} * P_{H_2O}}$$

$$K_4 = \frac{P_{H_2} * P_{CO_2}}{P_{CO} * P_{H_2O}}$$

$$P_i = \frac{n_i}{n_{tot}} * P$$

$$H_2r = 0.0562 * H_2$$

$$CO_r = 0.1348 * CO$$

$$CO_{2r} = 0.951 * CO_2$$

$$H_2O_r = 0.002 * H_2O$$

$$CH_{4r} = 0.994 * CH_4$$

$$N_{2r} = 0.13 * N_2$$

$$H_2 = H_{2r} + H_{2f} + 3 * x - y$$

$$CO = CO_r + CO_f + x - y$$

$$H_2O = H_{2O_r} + H_{2O_f} - x - y$$

$$CH_4 = CH_{4r} + CH_{4f} - x$$

$$CO_2 = CO_{2r} + CO_{2f} + y$$

$$N_2 = N_{2r} + N_{2f}$$

$$n = H_2 + CO + CO_2 + CH_4 + H_2O + N_2$$

with

T : temperature (K).

T_{appj} : temperature approach for reaction j (K).

P : pressure (atm).

m : multiplication factor for the natural gas (-).

H_2O_s : molar flow of steam, added to the feed (kmol/h).

i_f : molar flow of component i in feed (kmol/h).

i_r : molar flow of component i in recycle (kmol/h).

i : amount of component i in equilibrium with the other components (kmol/h).

K_j : equilibrium constant of reaction j.

x : conversion of CH_4 (kmol/h).

y : production of CO_2 (kmol/h).

n : total molar flow (kmol/h).

Furthermore, the flows of the components are assumed to be positive.

The recycle ratio is calculated by using the design streams.

Appendix 3.3. Calculation of the design (KTI) overall heat transfer coefficient.

For the calculation of the overall heat transfer coefficient the following data was required [A1]:

- k_{st} : Conductivity of tube metal (W/m/K)
- D_{ti} : Internal tube diameter (m)
- D_{te} : External tube diameter (m)
- a_w : Wall heat transfer coefficient (W/m²/K)
- k_{er} : Effective conductivity of the catalyst bed (W/m/K)
- k_{er}° : Static contribution of the effective conductivity of the catalyst bed (W/m/K)
- k_g : Process gas conductivity (W/m/K)
- d_p : equivalent particle diameter (m)
- ρ_{og} : density of process gas (kg/m³)
- v : velocity of process gas (m/s)
- μ : viscosity of process gas (Pa * s)
- C_{pg} : Specific heat of process gas (J/kg/K)
- V_{cyl} : volume of a cylinder with same external diameter and height as the annular catalyst particle (m³)
- S_{cyl} : external surface area of a cylinder with same external diameter and height as the annular catalyst particle (m²)
- S_p : external surface area of the catalyst particle (m²)
- V_p : volume of the catalyst particle (m³)
- ϵ : void fraction of packing (-)
- D_t : inner catalyst diameter (m)

k_{st} (Conductivity of the tube metal)

The tubes are made of steel containing 30-35 % Ni, 19 - 23 % Cr and 1 - 1.5% Nb [A2].

$$k_{st} = 23.2 \text{ W/m/K}$$

D_{ti} (internal tube diameter)

Information given by KTI [A2].

$$D_{ti} = 0.10 \text{ m}$$

D_{te} (external tube diameter)

Information given by KTI [A2].

$$D_{te} = 0.12 \text{ m}$$

k_{er}^o (static contribution of the effective conductivity of the catalyst bed)

Relation found in literature [A3]. For (Raschig) Rings equation (A3.1) was found.

$$\frac{k_{er}^o}{k_g} = 3 \quad (A3.1)$$

k_g (process gas conductivity)

The conductivities of the components and of the process gas, under the given conditions, are mentioned in table A3.1 [A4] (except for CO [A5]).

Table A3.1. Conductivities of the components and the process gas.

Comp.	Conductivity (W/m/K)	Percentage (weight%)	Contribution
H ₂	42.91E-2	0.029	12.44E-3
CO ₂	6.48E-2	0.344	22.29E-3
CO	6.67E-2	0.083	5.54E-3
H ₂ O	9.04E-2	0.403	36.43E-3
N ₂	6.21E-2	0.046	2.86E-3
CH ₄	15.65E-2	0.095	14.87E-3
gas			kg = 94.43E-3

C_p_g (specific heat of process gas)

For the calculation of the specific heat of the components, equation (A3.2) is used [A3.6].

$$C_{p_g}^{ig} = R \cdot (A + B \cdot T + C \cdot T^2 + D \cdot T^{-2}) \quad (\text{J/mol/K}) \quad (A3.2)$$

The constants of the components, used in equation (A3.2) are mentioned in table A3.2.

Table A3.2. Constants for the calculation of the specific heats of the components.

component	A	B	C	D
CH ₄	1.702	9.081E-3	-2.164E-6	---
N ₂	3.280	0.593E-3	---	0.040E5
H ₂	3.249	0.422E-3	---	0.083E5
CO	3.376	0.557E-3	---	-0.031E5
CO ₂	5.457	1.045E-3	---	-1.157E5
H ₂ O	3.470	1.450E-3	---	0.121E5

With equation (A3.2), the specific heats of the components and of the process gas can be calculated. See for the results table A3.3.

Table 3.3. Specific heat of the components and of the process gas.

Comp.	Specific heat (J/kg/K)	Percentage (weight%)	Contribution
H ₂	15205.0	0.029	441.0
CO ₂	1193.6	0.344	410.6
CO	1157.9	0.083	96.1
H ₂ O	2247.8	0.403	905.9
N ₂	1141.8	0.046	52.5
CH ₄	4343.8	0.095	412.7
gas			C _{pg} = 2318.7

v (velocity of process gas)

Under the prevailing conditions the ideal gas law is valid (A3.3).

$$n = \frac{p \cdot V}{R \cdot T} \quad (\text{A3.3})$$

With the law, the process gas volume flow through the reformer can be calculated.

$$\frac{13.25 \cdot 10^5 \cdot \phi_v}{8.314 \cdot 946.65} = 296.1 \cdot \frac{1000}{3600}$$

$$\implies \Phi_v = 0.4886 \text{ m}^3/\text{s}$$

The reformer has 20 tubes, so the process gas volume flow per tube can be determined.

$$\implies \Phi_v = 0.0244 \text{ m}^3/\text{s}/\text{tube}$$

This means for the velocity of the process gas through the tubes

$$\phi_v = v * A = v * \pi * \frac{1}{4} * D_u^2$$

$$\implies v = 3.11 \text{ m/s}$$

dp (equivalent particle diameter)

The dimensions of the rings are shown in fig. A3.1.

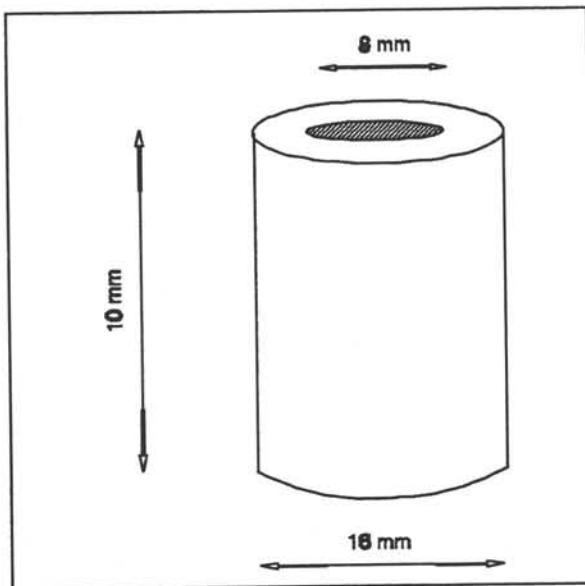


Figure A3.1. Dimensions of the catalyst rings.

The particle diameter d_p is defined as follows, to account for the shape of the annular catalyst particle.

$$d_p = 6 * \frac{V_{cyl}}{S_{cyl}} * E^n$$

with

$$E = \frac{V_p * S_{cyl}}{S_p * V_{cyl}}$$

and

$$n = \frac{\frac{D_d}{D_t}}{\left(\epsilon * \frac{D_d^2}{D_t^2}\right)^{0.4} + 0.01 * \left(\epsilon * \frac{D_d^2}{D_t^2}\right)^{0.75}}$$

$$\implies dp = 5.0E-3 \text{ m}$$

D_t (inner catalyst diameter)

Information given by KTI [A2]

$$D_t = 8E-3 \text{ m}$$

μ (viscosity of process gas)

The viscosities of the components and the process gas under the given conditions are mentioned in table A3.4.

Table A3.4. Viscosity of the components and process gas.

Comp.	Viscosity (Pa*s)	Percentage (weight%)	Contribution
H ₂	2.0E-5	0.029	0.6E-6
CO ₂	4.2E-5	0.344	14.0E-6
CO	4.1E-5	0.083	3.4E-6
H ₂ O	3.3E-5	0.403	13.0E-6
N ₂	4.1E-5	0.046	1.9E-6
CH ₄	2.9E-5	0.095	2.7E-6
			$\mu = 36.4E-6$

ϵ (void fraction of packing)

For the void fraction of the packing the following value is assumed.

$$\epsilon = 0.43 \text{ (-)}$$

ρ (density of process gas)

The density of the process gas can be calculated with equation (A3.4).

$$\rho_{\text{gas}} = \frac{\phi_m}{\phi_v} = \frac{5351.3}{1758.8} = 3.04 \frac{\text{kg}}{\text{m}^3} \quad (\text{A3.4})$$

Calculation of the dimensionless constants:

$$Re_p = \frac{\rho_{\text{gas}} * v * d_p}{\mu} = \frac{3.04 * 3.11 * 5.0E-3}{36.4E-6} = 1301$$

$$Pr = \frac{\mu * C_{p_g}}{k_g} = \frac{36.4E-6 * 2318.7}{94.43E-3} = 0.8931$$

Calculation of the overall heat transfer coefficient:

$$\alpha_w = \frac{8.694 * 0.283}{(0.1)^{\frac{4}{3}}} + \frac{0.512 * 94.43E-3 * 0.1 * 1301 * 0.9630}{5.0E-3}$$

$$= 53.06 + 1211.61$$

$$= 1264.67$$

$$k_{er} = 0.283 + \frac{0.111 * 94.43E-3 * 1301 * 0.9630}{1 + 46 * \frac{(5.0E-3)^2}{(0.1)^2}}$$

$$= 0.283 + 11.779$$

$$= 12.062$$

$$\frac{1}{U} = \frac{0.1}{2 * 23.2} * \ln(1.206) + \frac{1}{1264.67} + \frac{0.1}{8 * 12.062}$$

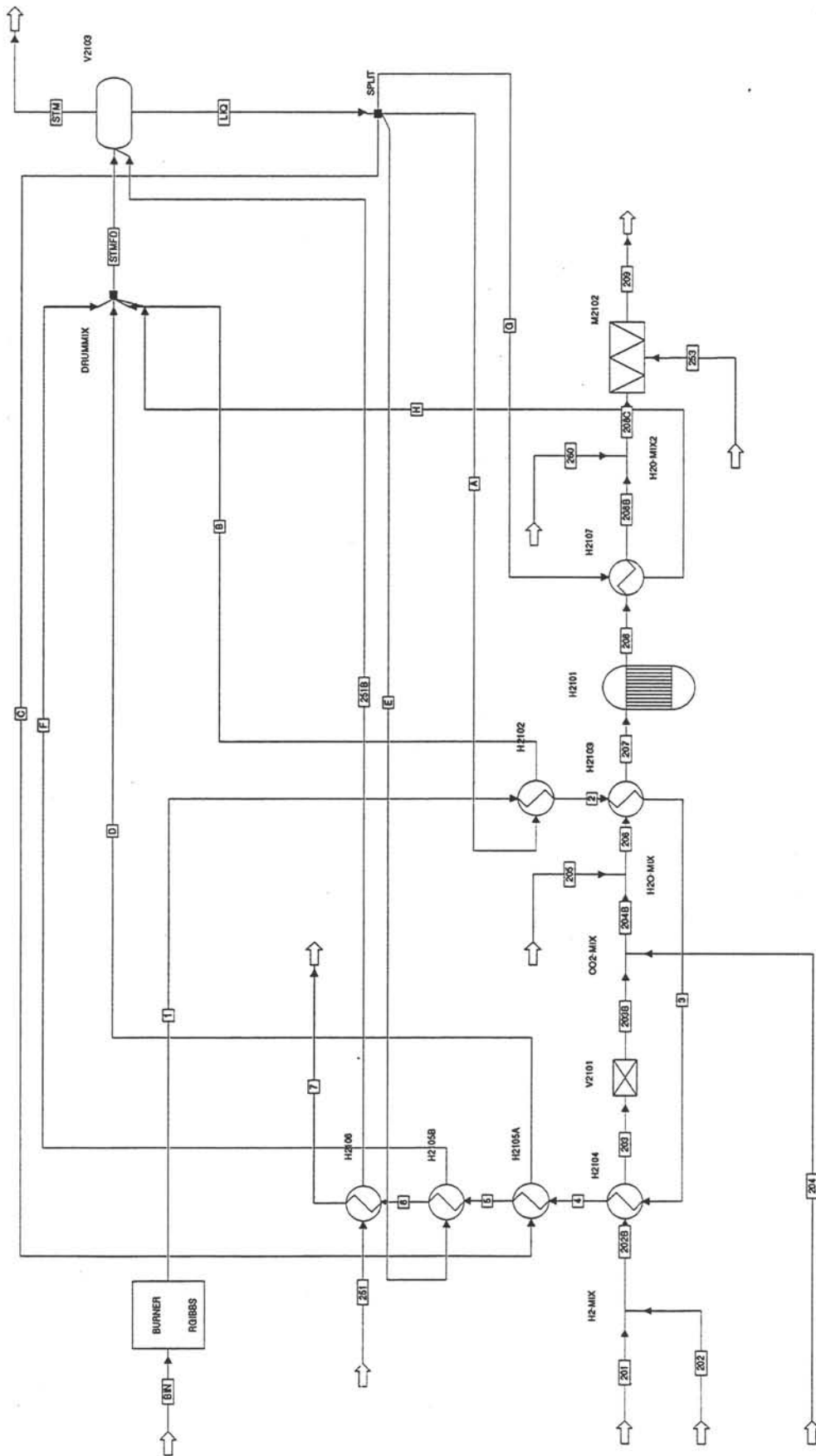
$$= 4.037E-4 + 7.907E-4 + 10.363E-4$$

$$\Longrightarrow U = 448 \text{ W/m}^2/\text{K}$$

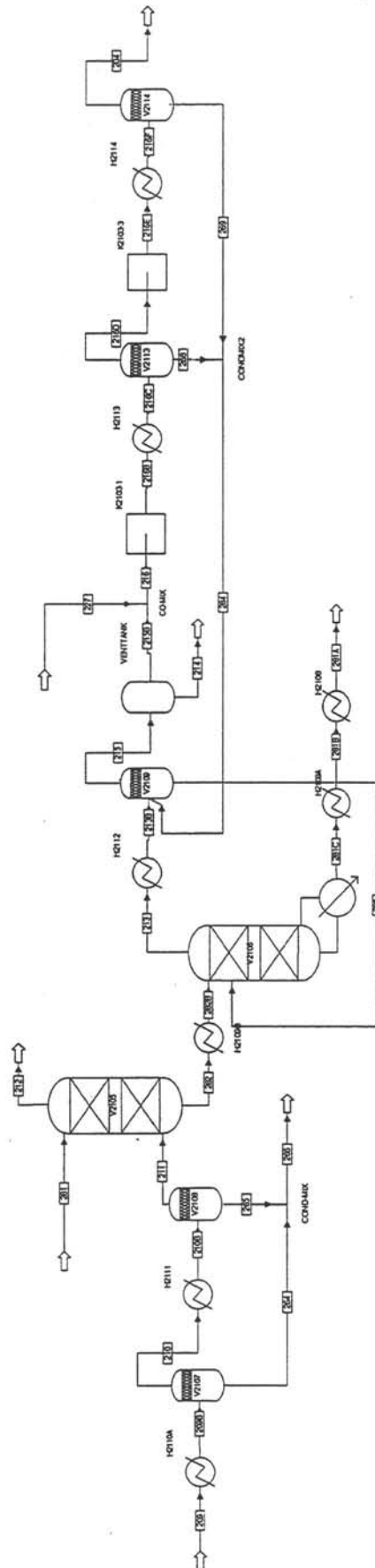
Literature

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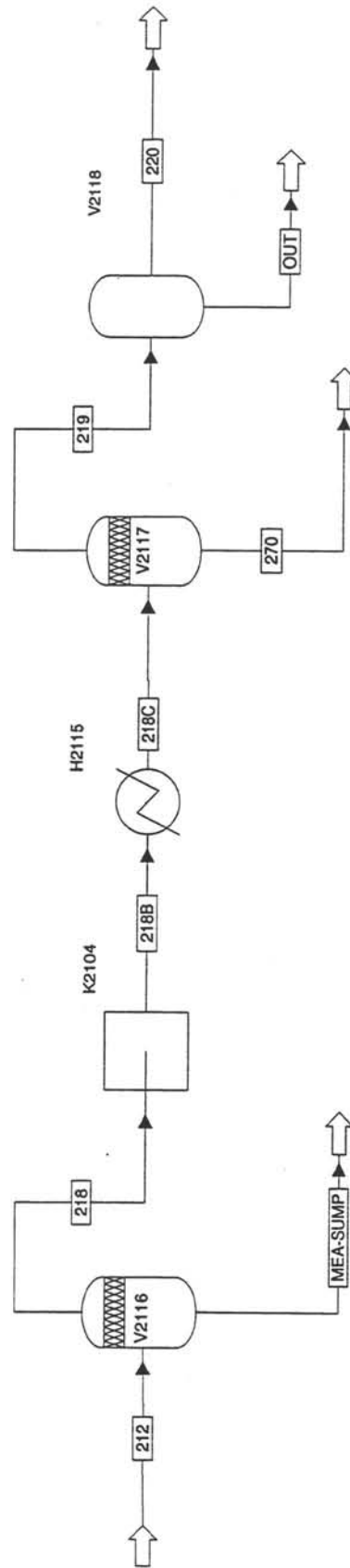
figuur A4.1 Aspen flowsheet reformer section



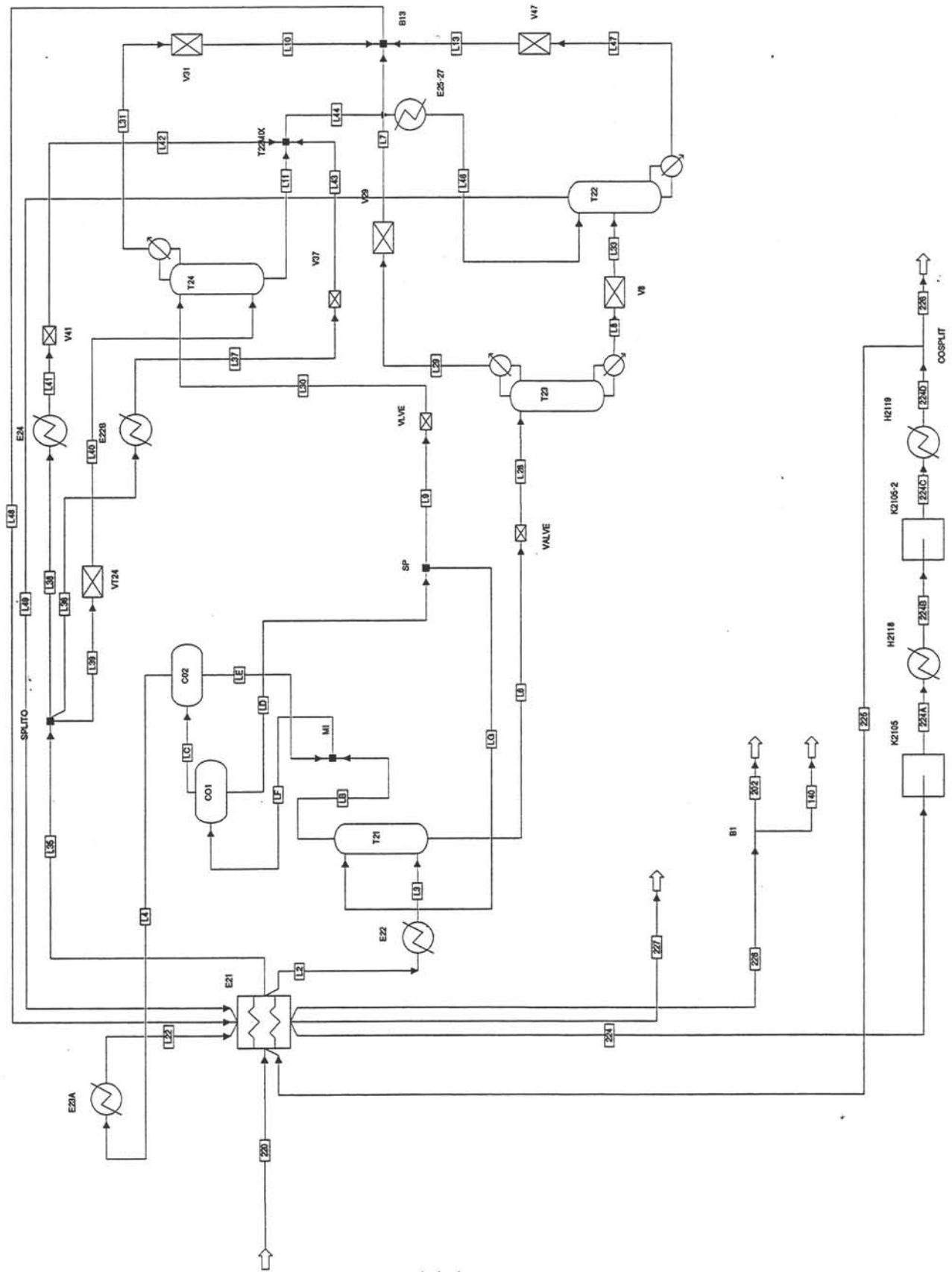
figuur A4.2 Aspen flowsheet CO₂ removal section



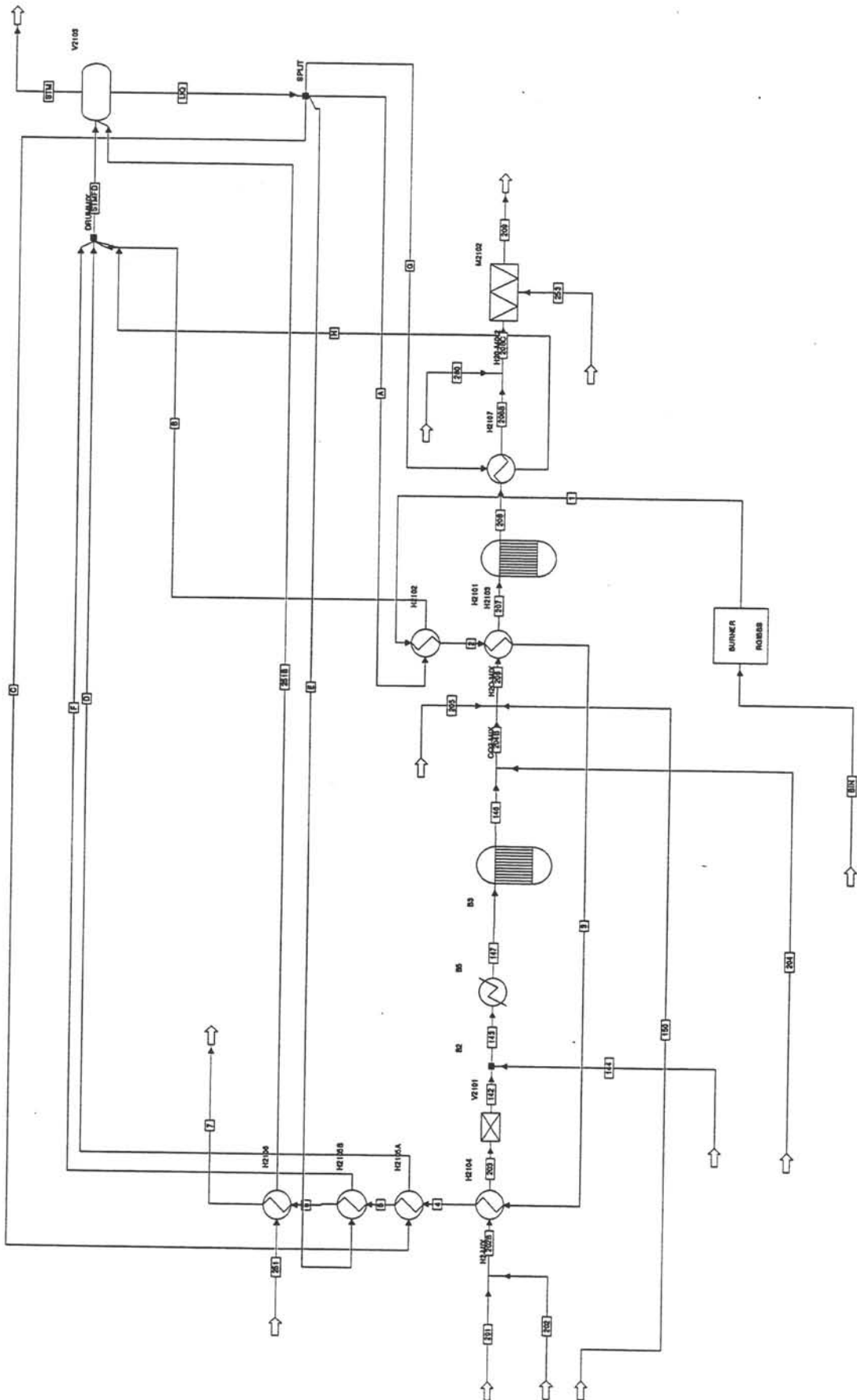
figuur A4.3 Aspen flowsheet drying section



figuur A4.4 Aspen flowsheet cryogenic separation section



figuur A4.5 reformer section with pre-reformer



APPENDIX 5

Block specifications of the reformer section used in Aspen

Table A5.1. The reactor (reformer).

Code	Block definition	Stream in	Stream out	T _{out} (°C)	p _{out} (bar)
H2101	RGIBBS	207	208	860	11.5

The Gibbs reactor has nine reactions, three of them have a temperature approach (chapter 5).

Table A5.2. The burners.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	-Δp (bar)
BURNER	RGIBBS	Bin	Bout	1005.0	0.5

Table A5.3. The heat exchangers.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	-Δp (bar)	Vapour fraction
H2102	HEATX	1 (h) A (c)	2 (h) B (c)		0 0	0.048
H2103	HEATX	2 (h) 206 (c)	3 (h) 207 (c)	487	0 0.5	
H2104	HEATX	3 (h) 202B (c)	4 (h) 203 (c)	400	0 0.3	
H2105A	HEATX	4 (h) C (c)	5 (h) D (c)		0 0	0.048
H2105B	HEATX	5 (h) E (c)	6 (h) F (c)		0 0	0.048
H2106	HEATX	6 (h) 251 (c)	7 (h) 251B (c)	180	0 0.2	
H2107	HEATX	208 (h) G (c)	208B (h) H (c)		0 0	0.048

'h' and 'c' stand for hot and cold stream.

Table A5.4. The desulphurizer.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	p _{out} (bar)
V2101	HEATER	203	203B	350.0	18.2

Table A5.5. The steamdrum.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	V _{frac}
V2103	FLASH2	STMFD 251B	STM (v) LIQ (l)	221	0.0421

'v' and 'l' stand for vapour and liquid.

Table A5.6. The mixing points.

Code	Block definition	Stream in	Stream out	P _{out} (bar)
M2102	MIXER	208C, 253	209	11.0
CO2-MIX	MIXER	203B, 204	204B	16.2
DRUMMIX	MIXER	B,D,F,H	STMFD	0
H2-MIX	MIXER	201, 202	202B	21.0
H2O-MIX	MIXER	204B, 205	206	15.6
H2O-MIX2	MIXER	208B, 260	208C	11.0

Table A5.7. The splitter.

Code	Block definition	Stream in	Stream out	Fraction of flow	-Δp (bar)
SPLIT	FSPLIT	LIQ	A C E G	0.13586 0.16845 0.56366	0

Block specifications of the coldbox section used in Aspen

Table A5.8. The heat exchangers and valves.

Code	Block definition	Stream in	Stream out	ΔT (°C)	T_{out} (°C)	$-\Delta p$ (bar)	p (bar)	Heat duty	
E21	MHEATX	220 (h)	L2 (h)	-192.6		0.2	3.9		
		225 (h)	L35 (h)	-190.1		0.2			
		L22 (c)	228 (c)						
		L49 (c)	224 (c)	205.5		0.2			
		L48 (c)	227 (c)	197.9		0.2			
E22	HEATER	L2	L3	- 7.2		0.1			
E22B	HEATER	L36	L37	- 5.7		0.1			
E23A	HEATER	L4	L22		-176.1				
E24	HEATER	L38	L41	- 2.5		0.1			
E25-27	HEATER	L44	L46		-184.5	0.18			
H2118	HEATER	224A	224B		30.0	0.2			
H2119	HEATER	224C	224D		35.0	0.2			
V29	HEATER	L29	L7			3.1	0		
V31	HEATER	L31	L10			5.3	0		
V37	HEATER	L37	L43			6.6	0		
V39	HEATER	L39	L40			1.25	0		
V41	HEATER	L41	L42			6.6	0		
V47	HEATER	L47	L13			0.1	0		
V6	HEATER	L6	L28			21.83	0		
V8	HEATER	L8	L33			3.02	0		
V9	HEATER	L9	L30			19.50	0		

Table A5.9. The splitters.

Code	Block definition	Stream in	Stream out	Fraction of flow	Flow (kg/h)
B1	FSPLIT	228	140		H ₂ : 10.7
			202		
CO-SPLIT	FSPLIT	224D	226		
			225	0.16347	
SPLIT1	FSPLIT	L35	L38	0.546	
			L36	0.156	
			L39		
SPLIT2	FSPLIT	LD	L9	0.25640	
			LG		

Table A5.10. The mixing points.

Code	Block definition	Stream in	Stream out	$-\Delta p$ (bar)
MIX1	MIXER	LB LE	LF	0
MIX2	MIXER	L13 L7 L10	L48	0
T22MIX	MIXER	L42 L43 L11	L44	0

Table A5.11. The flashers.

Code	Block definition	Stream in	Stream out	T_{out} ($^{\circ}C$)	p_{out} (bar)
CO1	FLASH2	LF	LC (v) LD (l)	-171.5	27.347
CO2	FLASH2	LC	L4 (v) LE (l)	-201.5	27.3

Table A5.12. The columns.

Code	Block definition	Stream in	Stream out	T ($^{\circ}C$)	p (bar)
T21	RADFRAC	L3 (st.20) LG (st.10)	LB (st.1) L6 (st.20)	-173.6 (st.1) -172.0 (st.20)	27.3 (st.1)
T22	RADFRAC	L33 (st.9) L46 (st.1)	L46 (st.1) L47 (st.12)		2.25 (st.1)
T23	RADFRAC	L28 (st.1)	L29 (st.1) L8 (st.4)		5.4 (st.1)
T24	RADFRAC	L40 (st.5) L30 (st.1)	L31 (st.1) L11 (st.5)		7.6 (st.1)

Code	Condensor duty	Reboiler duty	Molar boilup ratio	Molar reflux ratio	Mass reflux ratio
T21	0	0			
T22	0		7.51333		
T23	-4962.0			6.55	
T24		0			10.355

Table A5.13. The compressors.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	p _{out} (bar)	Mechanical efficiency
K2105-1	COMPR	224	224A	100.0	6.0	1.0
K2105-2	COMPR	224B	224C	90.0	9.6	1.0

The compressors are polytropic centrifugal compressors. 'st.' stands for stage, 'v' for vapour, 'l' for liquid, 'h' for hot streams and 'c' for cold streams.

Block specifications of the MEA section used in Aspen

Table A5.14. The heat exchangers.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	p _{out} (bar)	-Δp (bar)
H2108	HEATER	281B	281A	40.0	14.7	
H2109A	HEATER	281C	281B	88.0		0
H2109B	HEATER	282	282B	104.0		0
H2110A	HEATER	209	209B	132.0	10.7	
H2111	HEATER	210	210B	40.0	10.4	
H2112	HEATER	213	213B	40.0	1.5	
H2113	HEATER	216B	216C	40.0		0.2
H2114	HEATER	216E	216F	40.0	16.2	

Table A5.15. The separators.

Code	Block definition	Stream in	Stream out	-Δp (bar)	Fraction of flow
VENTTANK	SEP	215	215B	0	214: CO ₂ 0.047619 N ₂ 0.048326 MEA 1.0 H ₂ O 0.047620
			214	0	

Table A5.16. The mixing points.

Code	Block definition	Stream in	Stream out	p _{out} (bar)
CO-MIX	MIXER	215B	216	1.5
COND-MIX	MIXER	227	266	3.5
		264		
CONDMIX2	MIXER	265	284	1.5
		269		
		268		

Table A5.17. The flashers.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	P _{out} (bar)
V2107	FLASH2	209B	210 (v) 264 (l)	40.0	10.7
V2108	FLASH2	210B	211 (v) 265 (l)	40.0	10.4
V2109	FLASH2	213B 284	215 (v) 285 (l)	40.0	1.5
V2113	FLASH2	216C	216D (v) 268 (l)	40.0	9.6
V2114	FLASH2	216F	204 (v) 269 (l)	40.0	16.2

Table A5.18. The columns.

Code	Block definition	Stream in	Stream out	T _{spec} (°C)	P _{spec} (bar)
V2105	RADFRAC	281 (st.1)	212 (st.1)	40.0 (st.1)	10.3 (st.1)
		211 (st.20)	282 (st.20)	72.0 (st.20)	10.4 (st.20)
V2106	RADFRAC	285 (st.1)	213 (st.1)	102.0 (st.1)	1.7 (st.1)
		282B (st.5)	218C (st.18)	120.0 (st.16)	

Code	Condensor duty	Reboiler duty
V2105	0	0
V2106	0	2,850,000

V2105 has for component CO₂ a Murphree efficiency of 0.2 at stage 10. The reboiler duty of the V2106 column is an estimation, it is manipulated by the specification that the mass flow of CO₂ at stage 1 must be 1380 kg/h.

Table A5.19. The compressors.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	p _{out} (bar)	Mechanical efficiency
K2103-1	COMPR	216	216B	145.0	9.6	1.0
K2103-3	COMPR	216D	216E	145.0	15.4	1.0

'v' and 'l' stand for vapour and liquid, 'st.' stands for stage.

Block specifications process gas compression and drying section used in Aspen

Table A5.20. The heat exchanger.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	-Δp (bar)
H2115	HEATER	218B	218C	35.0	2 0.2

Table A5.21. The separator.

Code	Block definition	Stream in	Stream out	-Δp (bar)	Fractions of flow
V2118	SEP	219	220 OUT	0 0	220: H ₂ 0.994 CO 0.994 N ₂ 0.994 CH ₄ 0.994 rest 0

Table A5.22. The flashers.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	P _{out} (bar)
V2116	FLASH2	212	218 (v) MEASUMP (l)	40.0	10.3
V2117	FLASH2	218C	270 (l) 219 (v)	35.0	28.3

Table A5.23. The compressor.

Code	Block definition	Stream in	Stream out	T _{out} (°C)	P _{out} (bar)	Mechanical efficiency
K2104	COMPR	218	218B	145.0	28.5	1.0

'v' and 'l' stand for vapour and liquid.

Simulated design streams

STREAM ID	1	140	2	201	202
FROM :	BURNER	B1	H2102	----	B1
TO :	H2102	----	H2103	H2-MIX	H2-MIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	3.7117-05	139.4111	3.7117-05	0.0	5.3075
CO	1.2567-06	2.5668	1.2567-06	0.0	9.7723-02
CO2	2.3270	0.0	2.3270	0.4726	0.0
N2	260.6468	1.1188	260.6468	7.6178	4.2596-02
O2	9.2645	0.0	9.2645	0.0	0.0
CH4	7.8087-31	0.0	7.8087-31	43.1777	0.0
C2H6	0.0	0.0	0.0	1.5131	0.0
C3H8	0.0	0.0	0.0	0.1972	0.0
C4H10	0.0	0.0	0.0	7.3980-02	0.0
C5H12	0.0	0.0	0.0	2.0790-02	0.0
C6H14	0.0	0.0	0.0	2.6689-02	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	121.8739	0.0	121.8739	0.0	0.0
AR	3.1215	0.0	3.1215	0.0	0.0
TOTAL FLOW:					
KMOL/HR	397.2340	143.0969	397.2340	53.1001	5.4478
KG/HR	1.0021+04	384.2934	1.0021+04	989.2000	14.6304
CUM/SEC	7.8199	0.2718	7.0955	1.6236-02	1.0351-02
STATE VARIABLES:					
TEMP C	1005.0000	14.1653	886.5673	15.0000	14.1653
PRES BAR	1.5000	3.5000	1.5000	21.0000	3.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-4.3561+07	-2.2895+06	-4.7900+07	-6.8254+07	-2.2895+06
J/KG	-1.7268+06	-8.5254+05	-1.8988+06	-3.6638+06	-8.5254+05
WATT	-4.8066+06	-9.1007+04	-5.2854+06	-1.0067+06	-3464.7506
ENTROPY:					
J/KMOL-K	3.7538+04	-8631.7295	3.3977+04	-9.4630+04	-8631.7295
J/KG-K	1488.0866	-3214.1423	1346.8881	-5079.7284	-3214.1423
DENSITY:					
KMOL/CUM	1.4110-02	0.1461	1.5551-02	0.9084	0.1461
KG/CUM	0.3559	0.3926	0.3922	16.9236	0.3926
AVG MW	25.2260	2.6855	25.2260	18.6289	2.6855

Simulated design streams

STREAM ID	202B	203	203B	204	204B
FROM :	H2-MIX	H2104	V2101	V2114	CO2-MIX
TO :	H2104	V2101	CO2-MIX	CO2-MIX	H2O-MIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	5.3075	5.3075	5.3075	3.5315	8.8390
CO	9.7723-02	9.7723-02	9.7723-02	6.7552	6.8529
CO2	0.4726	0.4726	0.4726	29.8423	30.3149
N2	7.6604	7.6604	7.6604	0.9805	8.6410
O2	0.0	0.0	0.0	0.0	0.0
CH4	43.1777	43.1777	43.1777	4.5586	47.7363
C2H6	1.5131	1.5131	1.5131	1.0273-10	1.5131
C3H8	0.1972	0.1972	0.1972	0.0	0.1972
C4H10	7.3980-02	7.3980-02	7.3980-02	0.0	7.3980-02
C5H12	2.0790-02	2.0790-02	2.0790-02	0.0	2.0790-02
C6H14	2.6689-02	2.6689-02	2.6689-02	0.0	2.6689-02
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.1952	0.1952
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	58.5479	58.5479	58.5479	45.8634	104.4113
KG/HR	1003.8304	1003.8304	1003.8304	1613.8120	2617.6425
CUM/SEC	1.7996-02	4.4333-02	4.6625-02	1.9687-02	7.4631-02
STATE VARIABLES:					
TEMP C	14.2592	400.0000	350.0000	40.0000	226.9639
PRES BAR	21.0000	20.7000	18.2000	16.2000	16.2000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.2116+07	-4.5105+07	-4.7665+07	-2.8085+08	-1.5009+08
J/KG	-3.6229+06	-2.6307+06	-2.7801+06	-7.9815+06	-5.9868+06
WATT	-1.0102+06	-7.3355+05	-7.7520+05	-3.5780+06	-4.3532+06
ENTROPY:					
J/KMOL-K	-8.5515+04	-4.8826+04	-5.1700+04	-6218.8838	-2.5305+04
J/KG-K	-4987.6213	-2847.7721	-3015.3746	-176.7363	-1009.3497
DENSITY:					
KMOL/CUM	0.9037	0.3668	0.3488	0.6471	0.3886
KG/CUM	15.4942	6.2897	5.9805	22.7701	9.7429
AVG MW	17.1454	17.1454	17.1454	35.1873	25.0704

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Simulated design streams

STREAM ID	205	206	207	208	208B
FROM :	----	H2O-MIX	H2103	H2101	H2107
TO :	H2O-MIX	H2103	H2101	H2107	H2O-MIX2
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	0.0	8.8390	8.8390	149.1467	149.1467
CO	0.0	6.8529	6.8529	53.5113	53.5113
CO2	0.0	30.3149	30.3149	30.9846	30.9846
N2	0.0	8.6410	8.6410	8.6410	8.6410
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	47.7363	47.7363	4.5862	4.5862
C2H6	0.0	1.5131	1.5131	7.6396-05	7.6396-05
C3H8	0.0	0.1972	0.1972	3.1213-09	3.1213-09
C4H10	0.0	7.3980-02	7.3980-02	1.7555-13	1.7555-13
C5H12	0.0	2.0790-02	2.0790-02	6.6516-18	6.6516-18
C6H14	0.0	2.6689-02	2.6689-02	2.8846-22	2.8846-22
MEA	0.0	0.0	0.0	9.6476-31	9.6476-31
H2O	151.2739	151.4691	151.4691	103.4714	103.4714
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	151.2739	255.6853	255.6853	350.3414	350.3414
KG/HR	2725.2000	5342.8425	5342.8425	5342.8432	5342.8432
CUM/SEC	6.7293-02	0.1795	0.2964	0.7991	0.3815
STATE VARIABLES:					
TEMP C	220.0000	215.1403	487.0000	860.0000	269.2265
PRES BAR	23.2000	15.6000	15.1000	11.5000	11.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.3647+08	-2.0120+08	-1.8980+08	-9.5417+07	-1.1642+08
J/KG	-1.3126+07	-9.6284+06	-9.0830+06	-6.2567+06	-7.6339+06
WATT	-9.9366+06	-1.4290+07	-1.3480+07	-9.2857+06	-1.1330+07
ENTROPY:					
J/KMOL-K	-5.4783+04	-3.5298+04	-1.6547+04	3.5999+04	1.0004+04
J/KG-K	-3040.9421	-1689.2132	-791.8444	2360.5322	655.9938
DENSITY:					
KMOL/CUM	0.6244	0.3955	0.2396	0.1217	0.2550
KG/CUM	11.2492	8.2662	5.0071	1.8571	3.8901
AVG MW	18.0150	20.8961	20.8961	15.2503	15.2503

Simulated design streams

STREAM ID	208C	209	209B	210	210B
FROM :	H20-MIX2	M2102	H2110A	V2107	H2111
TO :	M2102	H2110A	V2107	H2111	V2108
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	MIXED	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	149.1467	149.1467	149.1467	149.1465	149.1465
CO	53.5113	53.5113	53.5113	53.5112	53.5112
CO2	30.9846	30.9846	30.9846	30.9640	30.9640
N2	8.6410	8.6410	8.6410	8.6410	8.6410
O2	0.0	0.0	0.0	0.0	0.0
CH4	4.5862	4.5862	4.5862	4.5862	4.5862
C2H6	7.6396-05	7.6396-05	7.6396-05	7.6396-05	7.6396-05
C3H8	3.1213-09	3.1213-09	3.1213-09	3.1213-09	3.1213-09
C4H10	1.7555-13	1.7555-13	1.7555-13	0.0	0.0
C5H12	6.6516-18	6.6516-18	6.6516-18	0.0	0.0
C6H14	2.8846-22	2.8846-22	2.8846-22	0.0	0.0
MEA	9.6476-31	9.6476-31	9.6476-31	0.0	0.0
H2O	205.9804	230.9596	230.9596	1.4203	1.4203
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	452.8505	477.8296	477.8296	248.2695	248.2695
KG/HR	7189.5432	7639.5432	7639.5432	3503.4858	3503.4858
CUM/SEC	0.4951	0.4461	0.2928	0.1683	0.1731
STATE VARIABLES:					
TEMP C	250.6157	177.3879	132.0000	40.0000	40.0000
PRES BAR	11.0000	11.0000	10.7000	10.7000	10.4000
VFRAC	1.0000	1.0000	0.7017	1.0000	1.0000
LFRAC	0.0	0.0	0.2982	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.4364+08	-1.5088+08	-1.6456+08	-7.5301+07	-7.5300+07
J/KG	-9.0475+06	-9.4369+06	-1.0293+07	-5.3361+06	-5.3360+06
WATT	-1.8069+07	-2.0026+07	-2.1842+07	-5.1930+06	-5.1930+06
ENTROPY:					
J/KMOL-K	-1294.8596	-8374.8101	-4.0828+04	9002.9636	9242.6793
J/KG-K	-81.5598	-523.8183	-2553.6608	637.9822	654.9693
DENSITY:					
KMOL/CUM	0.2540	0.2975	0.4533	0.4097	0.3983
KG/CUM	4.0337	4.7569	7.2473	5.7822	5.6206
AVG MW	15.8762	15.9880	15.9880	14.1116	14.1116

Simulated design streams

STREAM ID	211	212	213	213B	214
FROM :	V2108	V2105	V2106	H2112	VENTTANK
TO :	V2105	V2116	H2112	V2109	----
SUBSTREAM: MIXED					
PHASE:	VAPOR	MIXED	VAPOR	MIXED	MIXED
COMPONENTS: KMOL/HR					
H2	149.1465	149.1453	1.2347-03	1.2347-03	0.0
CO	53.5112	53.5103	9.0080-04	9.0080-04	0.0
CO2	30.9640	1.1361-02	31.3565	31.3565	1.4921
N2	8.6410	8.6409	5.3417-05	5.3417-05	2.5822-06
O2	0.0	0.0	0.0	0.0	0.0
CH4	4.5862	4.5862	1.9257-05	1.9257-05	0.0
C2H6	7.6396-05	7.6395-05	1.0253-10	1.0253-10	0.0
C3H8	3.1213-09	3.1213-09	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	4.0481-03	5.2707-05	5.2707-05	3.9385-05
H2O	1.4203	1.4521	222.5219	222.5219	6.1036-02
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	248.2695	217.3506	253.8807	253.8807	1.5532
KG/HR	3503.4858	2142.0500	5388.7656	5388.7656	66.7710
CUM/SEC	0.1731	0.1532	1.3112	0.1576	7.4384-03
STATE VARIABLES:					
TEMP C	40.0000	40.0442	110.9117	40.0000	40.0000
PRES BAR	10.4000	10.3000	1.7000	1.5000	1.5000
VFRAC	1.0000	0.9999	1.0000	0.1284	1.0000
LFRAC	0.0	2.4896-05	0.0	0.8715	2.0877-06
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-7.5300+07	-3.0027+07	-2.5786+08	-3.0052+08	-3.8730+08
J/KG	-5.3360+06	-3.0468+06	-1.2148+07	-1.4158+07	-9.0092+06
WATT	-5.1930+06	-1.8129+06	-1.8185+07	-2.1193+07	-1.6710+05
ENTROPY:					
J/KMOL-K	9242.6793	9210.8349	-3.1120+04	-1.4501+05	894.0483
J/KG-K	654.9693	934.6095	-1466.1721	-6831.6594	20.7971
DENSITY:					
KMOL/CUM	0.3983	0.3938	5.3781-02	0.4472	5.8003-02
KG/CUM	5.6206	3.8819	1.1415	9.4922	2.4934
AVG MW	14.1116	9.8552	21.2255	21.2255	42.9889

Simulated design streams

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STREAM ID          215          215B          216          216B          216C
FROM :            V2109          VENTTANK      CO-MIX      K2103-1      H2113
TO   :            VENTTANK      CO-MIX      K2103-1      H2113      V2113

SUBSTREAM: MIXED
PHASE:            VAPOR          VAPOR          VAPOR          VAPOR          MIXED
COMPONENTS: KMOL/HR
H2                1.2348-03    1.2348-03     3.5314        3.5314        3.5314
CO                9.0093-04    9.0093-04     6.7552        6.7552        6.7552
CO2               31.3349      29.8427       29.8427       29.8427       29.8427
N2               5.3433-05    5.0851-05     0.9805        0.9805        0.9805
O2                0.0          0.0           0.0           0.0           0.0
CH4              2.0125-05    2.0125-05     4.5587        4.5587        4.5587
C2H6             1.0253-10    1.0253-10     1.0253-10     1.0253-10     1.0253-10
C3H8             0.0          0.0           0.0           0.0           0.0
C4H10            0.0          0.0           0.0           0.0           0.0
C5H12            0.0          0.0           0.0           0.0           0.0
C6H14            0.0          0.0           0.0           0.0           0.0
MEA              3.9385-05    0.0           0.0           0.0           0.0
H2O              1.2817       1.2207        1.2207        1.2207        1.2207
AR               0.0          0.0           0.0           0.0           0.0
TOTAL FLOW:
KMOL/HR          32.6188      31.0656       46.8894       46.8894       46.8894
KG/HR            1402.1713    1335.4003     1632.3086     1632.3086     1632.3086
CUM/SEC          0.1562       0.1487        0.2202        4.6881-02     3.4589-02
STATE VARIABLES:
TEMP C           40.0000      40.0000       33.1462       145.0000      40.0000
PRES BAR         1.5000       1.5000        1.5000        9.6000        9.4000
VFRAC           1.0000       1.0000        1.0000        1.0000        0.9806
LFRAC           0.0          0.0           0.0           0.0           1.9364-02
SFRAC           0.0          0.0           0.0           0.0           0.0
ENTHALPY:
J/KMOL          -3.8728+08   -3.8728+08    -2.7988+08    -2.7584+08    -2.8072+08
J/KG            -9.0093+06   -9.0093+06    -8.0399+06    -7.9238+06    -8.0639+06
WATT            -3.5091+06   -3.3420+06    -3.6455+06    -3.5928+06    -3.6563+06
ENTROPY:
J/KMOL-K        916.9114     917.9789     1.3082+04     8963.4490    -4538.7397
J/KG-K          21.3302      21.3551      375.7963      257.4827     -130.3791
DENSITY:
KMOL/CUM        5.8003-02    5.8003-02     5.9145-02     0.2778        0.3765
KG/CUM          2.4933       2.4933        2.0589        9.6716        13.1089
AVG MW          42.9864      42.9863       34.8118       34.8118       34.8118
    
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Simulated design streams

STREAM ID	216D	216E	216F	218	218B
FROM :	V2113	K2103-3	H2114	V2116	K2104
TO :	K2103-3	H2114	V2114	K2104	H2115
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	MIXED	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	3.5315	3.5315	3.5315	149.1453	149.1453
CO	6.7552	6.7552	6.7552	53.5103	53.5103
CO2	29.8423	29.8423	29.8423	1.1361-02	1.1361-02
N2	0.9805	0.9805	0.9805	8.6409	8.6409
O2	0.0	0.0	0.0	0.0	0.0
CH4	4.5586	4.5586	4.5586	4.5862	4.5862
C2H6	1.0273-10	1.0273-10	1.0273-10	7.6395-05	7.6395-05
C3H8	0.0	0.0	0.0	3.1213-09	3.1213-09
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	4.0470-03	4.0470-03
H2O	0.3072	0.3072	0.3072	1.2700	1.2700
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	45.9755	45.9755	45.9755	217.1684	217.1684
KG/HR	1615.8331	1615.8331	1615.8331	2138.7677	2138.7677
CUM/SEC	3.3841-02	2.8583-02	1.9688-02	0.1533	7.4669-02
STATE VARIABLES:					
TEMP C	40.0000	145.0000	40.0000	40.0000	145.0000
PRES BAR	9.6000	15.4000	16.2000	10.3000	28.5000
VFRAC	1.0000	1.0000	0.9975	1.0000	1.0000
LFRAC	0.0	0.0	2.4378-03	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.8059+08	-2.7666+08	-2.8086+08	-2.9851+07	-2.6764+07
J/KG	-7.9837+06	-7.8719+06	-7.9915+06	-3.0310+06	-2.7176+06
WATT	-3.5834+06	-3.5333+06	-3.5869+06	-1.8007+06	-1.6145+06
ENTROPY:					
J/KMOL-K	-1525.2264	5401.2695	-6609.4827	9222.0813	9174.0573
J/KG-K	-43.3974	153.6829	-188.0604	936.4012	931.5248
DENSITY:					
KMOL/CUM	0.3773	0.4468	0.6486	0.3934	0.8078
KG/CUM	13.2631	15.7029	22.7977	3.8752	7.9564
AVG MW	35.1455	35.1455	35.1455	9.8484	9.8484

Simulated design streams

STREAM ID	218C	219	220	224	224A
FROM :	H2115	V2117	V2118	E21	K2105-1
TO :	V2117	V2118	E21	K2105-1	H2118
SUBSTREAM: MIXED					
PHASE:	MIXED	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	149.1453	149.1453	148.2504	1.5812-03	1.5812-03
CO	53.5103	53.5103	53.1893	52.3228	52.3228
CO2	1.1361-02	1.1361-02	0.0	0.0	0.0
N2	8.6409	8.6409	8.5891	7.7067	7.7067
O2	0.0	0.0	0.0	0.0	0.0
CH4	4.5862	4.5862	4.5586	6.5505-08	6.5505-08
C2H6	7.6395-05	7.6395-05	0.0	0.0	0.0
C3H8	3.1213-09	3.1213-09	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	4.0470-03	4.0328-03	0.0	0.0	0.0
H2O	1.2700	0.3701	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	217.1684	216.2685	214.5876	60.0311	60.0311
KG/HR	2138.7677	2122.5559	2102.4483	1681.4545	1681.4545
CUM/SEC	5.5039-02	5.5224-02	5.4804-02	0.1991	8.6447-02
STATE VARIABLES:					
TEMP C	35.0000	35.0000	35.0000	21.3379	100.0000
PRES BAR	28.4000	28.3000	28.3000	2.0500	6.0000
VFRAC	0.9958	1.0000	1.0000	1.0000	1.0000
LFrac	4.1490-03	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-3.0196+07	-2.9122+07	-2.8734+07	-9.6533+07	-9.4239+07
J/KG	-3.0660+06	-2.9673+06	-2.9328+06	-3.4464+06	-3.3645+06
WATT	-1.8215+06	-1.7495+06	-1.7128+06	-1.6097+06	-1.5715+06
ENTROPY:					
J/KMOL-K	-380.6202	347.4828	375.6344	7.5131+04	7.3097+04
J/KG-K	-38.6478	35.4052	38.3393	2682.3135	2609.6987
DENSITY:					
KMOL/CUM	1.0960	1.0878	1.0876	8.3727-02	0.1929
KG/CUM	10.7942	10.6764	10.6563	2.3451	5.4029
AVG MW	9.8484	9.8144	9.7976	28.0097	28.0097

Simulated design streams

STREAM ID	224B	224C	224D	225	226
FROM :	H2118	K2105-2	H2119	CO-SPLIT	CO-SPLIT
TO :	K2105-2	H2119	CO-SPLIT	E21	----
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	1.5812-03	1.5812-03	1.5812-03	2.5847-04	1.3227-03
CO	52.3228	52.3228	52.3228	8.5532	43.7696
CO2	0.0	0.0	0.0	0.0	0.0
N2	7.7067	7.7067	7.7067	1.2598	6.4469
O2	0.0	0.0	0.0	0.0	0.0
CH4	6.5505-08	6.5505-08	6.5505-08	1.0708-08	5.4797-08
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	60.0311	60.0311	60.0311	9.8132	50.2178
KG/HR	1681.4545	1681.4545	1681.4545	274.8673	1406.5871
CUM/SEC	7.2499-02	5.2650-02	4.5505-02	7.4387-03	3.8066-02
STATE VARIABLES:					
TEMP C	30.0000	90.0000	35.0000	35.0000	35.0000
PRES BAR	5.8000	9.6000	9.4000	9.4000	9.4000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-9.6303+07	-9.4546+07	-9.6178+07	-9.6178+07	-9.6178+07
J/KG	-3.4382+06	-3.3755+06	-3.4337+06	-3.4337+06	-3.4337+06
WATT	-1.6059+06	-1.5766+06	-1.6038+06	-2.6217+05	-1.3416+06
ENTROPY:					
J/KMOL-K	6.7251+04	6.8342+04	6.3645+04	6.3645+04	6.3645+04
J/KG-K	2400.9940	2439.9395	2272.2448	2272.2448	2272.2448
DENSITY:					
KMOL/CUM	0.2300	0.3167	0.3664	0.3664	0.3664
KG/CUM	6.4424	8.8711	10.2642	10.2642	10.2642
AVG MW	28.0097	28.0097	28.0097	28.0097	28.0097

Simulated design streams

STREAM ID	227	228	251	251B	253
FROM :	E21	E21	----	H2106	----
TO :	CO-MIX	B1	H2106	V2103	M2102
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	LIQUID	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	3.5302	144.7187	0.0	0.0	0.0
CO	6.7543	2.6645	0.0	0.0	0.0
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.9805	1.1614	0.0	0.0	0.0
O2	0.0	0.0	0.0	0.0	0.0
CH4	4.5586	0.0	0.0	0.0	0.0
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	343.2639	343.2639	24.9791
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	15.8237	148.5447	343.2639	343.2639	24.9791
KG/HR	296.9083	398.9239	6183.9000	6183.9000	450.0000
CUM/SEC	5.4401-02	0.2822	2.4282-03	2.6602-03	1.7670-04
STATE VARIABLES:					
TEMP C	17.2162	14.1653	104.0000	180.0000	104.0000
PRES BAR	1.9500	3.5000	27.0000	26.8000	27.0000
VFRAC	1.0000	1.0000	0.0	0.0	0.0
LFRAC	0.0	0.0	1.0000	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.9048+07	-2.2895+06	-2.8210+08	-2.7526+08	-2.8210+08
J/KG	-3.6799+06	-8.5254+05	-1.5659+07	-1.5279+07	-1.5659+07
WATT	-3.0350+05	-9.4472+04	-2.6899+07	-2.6246+07	-1.9574+06
ENTROPY:					
J/KMOL-K	1.8996+04	-8631.7295	-1.5042+05	-1.3390+05	-1.5042+05
J/KG-K	1012.4158	-3214.1423	-8349.6409	-7432.5015	-8349.6409
DENSITY:					
KMOL/CUM	8.0798-02	0.1461	39.2685	35.8435	39.2685
KG/CUM	1.5160	0.3926	707.4237	645.7212	707.4237
AVG MW	18.7634	2.6855	18.0150	18.0150	18.0150

Simulated design streams

STREAM ID	260	264	265	266	268
FROM :	----	V2107	V2108	COND-MIX	V2113
TO :	H2O-MIX2	COND-MIX	COND-MIX	----	CONDMIX2
SUBSTREAM: MIXED					
PHASE:	VAPOR	LIQUID	MIXED	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	0.0	1.9116-04	0.0	1.9116-04	8.9701-08
CO	0.0	4.1375-05	0.0	4.1375-05	1.0236-07
CO2	0.0	2.0550-02	0.0	2.0550-02	3.7708-04
N2	0.0	6.1456-06	0.0	6.1456-06	1.3683-08
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	3.7444-05	0.0	3.7444-05	7.2032-07
C2H6	0.0	9.3157-10	0.0	9.3157-10	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	102.5090	229.5393	0.0	229.5393	0.9134
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	102.5090	229.5601	0.0	229.5601	0.9138
KG/HR	1846.7000	4136.0574	0.0	4136.0574	16.4728
CUM/SEC	5.7348-02	1.5381-03	0.0	1.5384-03	6.1242-06
STATE VARIABLES:					
TEMP C	208.0000	40.0000	MISSING	40.1538	40.0000
PRES BAR	18.3000	10.7000	10.4000	3.5000	9.6000
VFRAC	1.0000	0.0	MISSING	0.0	0.0
LFRAC	0.0	1.0000	MISSING	1.0000	1.0000
SFRAC	0.0	0.0	MISSING	0.0	0.0
ENTHALPY:					
J/KMOL	-2.3667+08	-2.8771+08	MISSING	-2.8771+08	-2.8774+08
J/KG	-1.3137+07	-1.5968+07	MISSING	-1.5968+07	-1.5963+07
WATT	-6.7391+06	-1.8346+07	MISSING	-1.8346+07	-7.3043+04
ENTROPY:					
J/KMOL-K	-5.3384+04	-1.6654+05	MISSING	-1.6648+05	-1.6648+05
J/KG-K	-2963.3323	-9243.1010	MISSING	-9240.0219	-9235.5050
DENSITY:					
KMOL/CUM	0.4965	41.4588	MISSING	41.4489	41.4502
KG/CUM	8.9449	746.9772	MISSING	746.7984	747.1707
AVG MW	18.0150	18.0173	MISSING	18.0173	18.0257

Simulated design streams

STREAM ID	269	270	281	281A	281B
FROM :	V2114	V2117	----	H2108	H2109A
TO :	CONDMIX2	----	V2105	----	H2108
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	LIQUID	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	1.8952-08	1.8243-06	0.0	0.0	0.0
CO	2.1299-08	3.9548-07	0.0	0.0	0.0
CO2	7.5729-05	8.3739-08	9.5819	9.2001	9.2001
N2	2.8517-09	5.8657-08	0.0	0.0	0.0
O2	0.0	0.0	0.0	0.0	0.0
CH4	1.4756-07	3.6903-07	0.0	0.0	0.0
C2H6	0.0	8.8102-12	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	1.4206-05	87.1169	87.1128	87.1128
H2O	0.1120	0.8998	775.1928	774.9113	774.9113
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	0.1120	0.8998	871.8917	871.2243	871.2243
KG/HR	2.0210	16.2117	1.9708+04	1.9686+04	1.9686+04
CUM/SEC	7.5113-07	6.0054-06	6.0353-03	8.0918-03	8.4725-03
STATE VARIABLES:					
TEMP C	40.0000	35.0000	40.0000	40.0000	88.0000
PRES BAR	16.2000	28.3000	14.7000	14.7000	1.7000
VFRAC	0.0	0.0	0.0	0.0	0.0
LFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.8775+08	-2.8810+08	-2.8449+08	-2.8376+08	-2.7909+08
J/KG	-1.5958+07	-1.5991+07	-1.2586+07	-1.2558+07	-1.2351+07
WATT	-8958.7728	-7.2014+04	-6.8900+07	-6.8671+07	-6.7541+07
ENTROPY:					
J/KMOL-K	-1.6644+05	-1.6799+05	-2.4699+05	-2.2427+05	-2.1026+05
J/KG-K	-9230.1662	-9324.4646	-1.0927+04	-9925.2547	-9305.2783
DENSITY:					
KMOL/CUM	41.4486	41.6232	40.1289	29.9076	28.5637
KG/CUM	747.4247	749.8700	907.0601	675.7787	645.4138
AVG MW	18.0325	18.0156	22.6036	22.5955	22.5955

Simulated design streams

STREAM ID	281C	282	282B	284	285
FROM :	V2106	V2105	H2109B	CONDMIX2	V2109
TO :	H2109A	H2109B	V2106	V2109	V2106
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	MIXED	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	0.0	1.2347-03	1.2347-03	1.0865-07	1.6489-09
CO	0.0	9.0080-04	9.0080-04	1.2366-07	7.3046-10
CO2	9.2001	40.5346	40.5346	4.5281-04	2.2065-02
N2	0.0	5.3417-05	5.3417-05	1.6535-08	3.9826-11
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	1.9257-05	1.9257-05	8.6788-07	1.7323-10
C2H6	0.0	1.0253-10	1.0253-10	0.0	0.0
C3H8	0.0	7.3884-16	7.3884-16	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	87.1128	87.1128	87.1128	0.0	1.3322-05
H2O	774.9113	775.1610	775.1610	1.0254	222.2656
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	871.2243	902.8106	902.8106	1.0259	222.2877
KG/HR	1.9686+04	2.1069+04	2.1069+04	18.4939	4005.0882
CUM/SEC	6.6315-03	6.5955-03	4.1647-02	6.8773-06	1.4896-03
STATE VARIABLES:					
TEMP C	118.0213	68.9435	104.0000	40.1884	40.0000
PRES BAR	1.7000	10.4000	10.4000	1.5000	1.5000
VFRAC	0.0	0.0	4.5200-02	0.0	0.0
LFRAC	1.0000	1.0000	0.9548	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7789+08	-2.8822+08	-2.8116+08	-2.8774+08	-2.8772+08
J/KG	-1.2298+07	-1.2350+07	-1.2047+07	-1.5962+07	-1.5969+07
WATT	-6.7251+07	-7.2280+07	-7.0509+07	-8.2001+04	-1.7766+07
ENTROPY:					
J/KMOL-K	-2.0053+05	-2.2534+05	-1.9835+05	-1.6641+05	-1.6652+05
J/KG-K	-8874.7396	-9655.7333	-8499.1980	-9231.1477	-9241.9675
DENSITY:					
KMOL/CUM	36.4936	38.0229	6.0215	41.4379	41.4520
KG/CUM	824.5923	887.3604	140.5274	746.9790	746.8649
AVG MW	22.5955	23.3374	23.3374	18.0264	18.0175

Simulated design streams

STREAM ID	3	4	5	6	7
FROM :	H2103	H2104	H2105A	H2105B	H2106
TO :	H2104	H2105A	H2105B	H2106	----
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	3.7117-05	3.7117-05	3.7117-05	3.7117-05	3.7117-05
CO	1.2567-06	1.2567-06	1.2567-06	1.2567-06	1.2567-06
CO2	2.3270	2.3270	2.3270	2.3270	2.3270
N2	260.6468	260.6468	260.6468	260.6468	260.6468
O2	9.2645	9.2645	9.2645	9.2645	9.2645
CH4	7.8087-31	7.8087-31	7.8087-31	7.8087-31	7.8087-31
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	121.8739	121.8739	121.8739	121.8739	121.8739
AR	3.1215	3.1215	3.1215	3.1215	3.1215
TOTAL FLOW:					
KMOL/HR	397.2340	397.2340	397.2340	397.2340	397.2340
KG/HR	1.0021+04	1.0021+04	1.0021+04	1.0021+04	1.0021+04
CUM/SEC	5.8412	5.4002	4.5950	3.5573	2.4016
STATE VARIABLES:					
TEMP C	681.5454	609.4574	477.8817	308.3998	120.0029
PRES BAR	1.5000	1.5000	1.5000	1.5000	1.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-5.5236+07	-5.7744+07	-6.2209+07	-6.7745+07	-7.3661+07
J/KG	-2.1897+06	-2.2891+06	-2.4661+06	-2.6855+06	-2.9201+06
WATT	-6.0949+06	-6.3716+06	-6.8643+06	-7.4751+06	-8.1280+06
ENTROPY:					
J/KMOL-K	2.7019+04	2.4288+04	1.8813+04	1.0467+04	-1814.9320
J/KG-K	1071.0806	962.8275	745.7787	414.9155	-71.9468
DENSITY:					
KMOL/CUM	1.8890-02	2.0433-02	2.4013-02	3.1018-02	4.5946-02
KG/CUM	0.4765	0.5154	0.6057	0.7824	1.1590
AVG MW	25.2260	25.2260	25.2260	25.2260	25.2260

Simulated design streams

STREAM ID	A	B	BIN	C	D
FROM :	SPLIT	H2102	----	SPLIT	H2105A
TO :	H2102	DRUMMIX	BURNER	H2105A	DRUMMIX
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	VAPOR	LIQUID	MIXED
COMPONENTS: KMOL/HR					
H2	0.0	0.0	121.8254	0.0	0.0
CO	0.0	0.0	2.3027	0.0	0.0
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.0	0.0	260.6468	0.0	0.0
O2	0.0	0.0	71.3772	0.0	0.0
CH4	0.0	0.0	2.4310-02	0.0	0.0
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	1031.1889	1031.1889	0.0	1061.1022	1061.1022
AR	0.0	0.0	3.1215	0.0	0.0
TOTAL FLOW:					
KMOL/HR	1031.1889	1031.1889	459.2980	1061.1022	1061.1022
KG/HR	1.8577+04	1.8577+04	1.0021+04	1.9116+04	1.9116+04
CUM/SEC	8.5677-03	2.9407-02	1.6619	8.8163-03	3.0260-02
STATE VARIABLES:					
TEMP C	221.0000	221.0000	40.0000	221.0000	221.0000
PRES BAR	24.0094	24.0094	2.0000	24.0094	24.0094
VFRAC	0.0	4.8000-02	1.0000	0.0	4.8000-02
LFRAC	1.0000	0.9520	0.0	1.0000	0.9520
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7130+08	-2.6962+08	-1.3052+05	-2.7130+08	-2.6962+08
J/KG	-1.5059+07	-1.4967+07	-5982.4219	-1.5059+07	-1.4967+07
WATT	-7.7710+07	-7.7232+07	-1.6652+04	-7.9965+07	-7.9472+07
ENTROPY:					
J/KMOL-K	-1.2552+05	-1.2213+05	4700.2516	-1.2552+05	-1.2213+05
J/KG-K	-6967.2642	-6779.5164	215.4359	-6967.2642	-6779.5164
DENSITY:					
KMOL/CUM	33.4325	9.7406	7.6766-02	33.4325	9.7406
KG/CUM	602.2870	175.4783	1.6748	602.2870	175.4783
AVG MW	18.0150	18.0150	21.8174	18.0150	18.0150

Simulated design streams

STREAM ID	E	F	G	H	L10
FROM :	SPLIT	H2105B	SPLIT	H2107	V31
TO :	H2105B	DRUMMIX	H2107	DRUMMIX	MIX2
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	LIQUID	MIXED	VAPOR
COMPONENTS: KMOL/HR					
H2	0.0	0.0	0.0	0.0	1.2259
CO	0.0	0.0	0.0	0.0	1.3294
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.0	0.0	0.0	0.0	0.3848
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	0.0	0.0	0.0	1.1418-18
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	1315.6387	1315.6387	4402.3326	4402.3326	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	1315.6387	1315.6387	4402.3326	4402.3326	2.9402
KG/HR	2.3701+04	2.3701+04	7.9308+04	7.9308+04	50.4898
CUM/SEC	1.0931-02	3.7518-02	3.6577-02	0.1255	2.5762-03
STATE VARIABLES:					
TEMP C	221.0000	221.0000	221.0000	221.0000	-183.0864
PRES BAR	24.0094	24.0094	24.0094	24.0094	2.3000
VFRAC	0.0	4.8000-02	0.0	4.8000-02	1.0000
LFRAC	1.0000	0.9520	1.0000	0.9520	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7130+08	-2.6962+08	-2.7130+08	-2.6962+08	-5.5917+07
J/KG	-1.5059+07	-1.4967+07	-1.5059+07	-1.4967+07	-3.2563+06
WATT	-9.9147+07	-9.8536+07	-3.3176+08	-3.2972+08	-4.5669+04
ENTROPY:					
J/KMOL-K	-1.2552+05	-1.2213+05	-1.2552+05	-1.2213+05	8253.5354
J/KG-K	-6967.2642	-6779.5164	-6967.2642	-6779.5158	480.6349
DENSITY:					
KMOL/CUM	33.4325	9.7406	33.4325	9.7406	0.3170
KG/CUM	602.2870	175.4783	602.2870	175.4783	5.4440
AVG MW	18.0150	18.0150	18.0150	18.0150	17.1721

Simulated design streams

STREAM ID	L11	L13	L2	L22	L28
FROM :	T24	V47	E21	E23A	V6
TO :	T22MIX	MIX2	E22	E21	T23
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	VAPOR	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	4.1183-04	1.6328-15	148.2504	144.7187	2.3053
CO	17.1293	2.0378	53.1893	2.6645	34.6148
CO2	0.0	0.0	0.0	0.0	0.0
N2	3.3475	4.0712-02	8.5891	1.1614	4.0705
O2	0.0	0.0	0.0	0.0	0.0
CH4	3.1910-09	4.5204	4.5586	0.0	4.5586
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	20.4772	6.5990	214.5876	148.5447	45.5493
KG/HR	573.5680	130.7428	2102.4483	398.9239	1161.3720
CUM/SEC	2.4084-04	1.0299-04	1.8868-02	8.5078-02	3.0026-03
STATE VARIABLES:					
TEMP C	-169.3914	-172.9837	-157.6000	-176.1000	-177.3833
PRES BAR	7.6000	2.1500	28.1000	3.9000	5.4700
VFRAC	0.0	5.2909-03	1.0000	1.0000	0.1535
LFRAC	1.0000	0.9947	0.0	0.0	0.8464
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0348+08	-9.9581+07	-3.4366+07	-7.3772+06	-1.0262+08
J/KG	-3.6945+06	-5.0262+06	-3.5076+06	-2.7470+06	-4.0247+06
WATT	-5.8862+05	-1.8254+05	-2.0485+06	-3.0440+05	-1.2984+06
ENTROPY:					
J/KMOL-K	-1.8338+04	-1.3264+05	-2.8241+04	-3.8018+04	-2.9782+04
J/KG-K	-654.6788	-6694.9005	-2882.4528	-1.4157+04	-1168.0410
DENSITY:					
KMOL/CUM	23.6182	17.7987	3.1591	0.4849	4.2138
KG/CUM	661.5463	352.6361	30.9522	1.3024	107.4414
AVG MW	28.0099	19.8124	9.7976	2.6855	25.4970

Simulated design streams

STREAM ID	L29	L3	L30	L31	L33
FROM :	T23	E22	V9	T24	V8
TO :	V29	T21	T24	V31	T22
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	MIXED	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	2.3043	148.2504	1.2262	1.2259	9.8799-04
CO	3.3870	53.1893	15.9099	1.3294	31.2277
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.5549	8.5891	3.3569	0.3848	3.5156
O2	0.0	0.0	0.0	0.0	0.0
CH4	3.8212-02	4.5586	0.0	1.1418-18	4.5204
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	6.2845	214.5876	20.4931	2.9402	39.2648
KG/HR	115.6755	2102.4483	542.1473	50.4898	1045.6965
CUM/SEC	2.3454-03	1.7354-02	8.9800-04	7.7472-04	3.9269-03
STATE VARIABLES:					
TEMP C	-178.8131	-164.8000	-176.7005	-177.7584	-182.1491
PRES BAR	5.4000	28.0000	7.8500	7.6000	2.3830
VFRAC	1.0000	1.0000	0.1341	1.0000	0.1107
LFRAC	0.0	0.0	0.8659	0.0	0.8892
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.5993+07	-3.4585+07	-9.6519+07	-5.5918+07	-1.0828+08
J/KG	-3.5853+06	-3.5299+06	-3.6484+06	-3.2563+06	-4.0657+06
WATT	-1.1520+05	-2.0615+06	-5.4944+05	-4.5670+04	-1.1810+06
ENTROPY:					
J/KMOL-K	8440.6659	-3.0173+04	-2.0389+04	-1127.7300	-3.3430+04
J/KG-K	458.5718	-3079.5921	-770.7210	-65.6720	-1255.2716
DENSITY:					
KMOL/CUM	0.7443	3.4349	6.3391	1.0542	2.7774
KG/CUM	13.7003	33.6539	167.7024	18.1033	73.9687
AVG MW	18.4064	9.7976	26.4550	17.1721	26.6318

Simulated design streams

STREAM ID	L35	L36	L37	L38	L39
FROM :	E21	SPLIT1	E22B	SPLIT1	SPLIT1
TO :	SPLIT1	E22B	V37	E24	V39
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	2.5847-04	4.0322-05	4.0322-05	1.4113-04	7.7025-05
CO	8.5532	1.3343	1.3343	4.6700	2.5488
CO2	0.0	0.0	0.0	0.0	0.0
N2	1.2598	0.1965	0.1965	0.6878	0.3754
O2	0.0	0.0	0.0	0.0	0.0
CH4	1.0708-08	1.6705-09	1.6705-09	5.8466-09	3.1910-09
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	9.8132	1.5308	1.5308	5.3580	2.9243
KG/HR	274.8673	42.8793	42.8793	150.0775	81.9104
CUM/SEC	2.5055-03	3.9086-04	3.6609-04	1.3680-03	7.4664-04
STATE VARIABLES:					
TEMP C	-155.1000	-155.1000	-160.8000	-155.1000	-155.1000
PRES BAR	9.2000	9.2000	9.1000	9.2000	9.2000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0205+08	-1.0205+08	-1.0225+08	-1.0205+08	-1.0205+08
J/KG	-3.6434+06	-3.6434+06	-3.6506+06	-3.6434+06	-3.6434+06
WATT	-2.7818+05	-4.3397+04	-4.3482+04	-1.5189+05	-8.2899+04
ENTROPY:					
J/KMOL-K	3.3838+04	3.3838+04	3.2169+04	3.3838+04	3.3838+04
J/KG-K	1208.0928	1208.0928	1148.4985	1208.0928	1208.0928
DENSITY:					
KMOL/CUM	1.0879	1.0879	1.1615	1.0879	1.0879
KG/CUM	30.4738	30.4738	32.5353	30.4738	30.4738
AVG MW	28.0097	28.0097	28.0097	28.0097	28.0097

Simulated design streams

STREAM ID	L4	L40	L41	L42	L43
FROM :	CO2	V39	E24	V41	V37
TO :	E23A	T24	V41	T22MIX	T22MIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	144.7187	7.7025-05	1.4113-04	1.4113-04	4.0322-05
CO	2.6645	2.5488	4.6700	4.6700	1.3343
CO2	0.0	0.0	0.0	0.0	0.0
N2	1.1614	0.3754	0.6878	0.6878	0.1965
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	3.1910-09	5.8466-09	5.8466-09	1.6705-09
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	148.5447	2.9243	5.3580	5.3580	1.5308
KG/HR	398.9239	81.9104	150.0775	150.0775	42.8793
CUM/SEC	8.3025-03	8.6639-04	1.3406-03	4.9807-03	1.3647-03
STATE VARIABLES:					
TEMP C	-201.5000	-156.8238	-157.6000	-167.4827	-171.2600
PRES BAR	27.3000	7.9500	9.1000	2.5000	2.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-8.1343+06	-1.0205+08	-1.0214+08	-1.0214+08	-1.0225+08
J/KG	-3.0289+06	-3.6434+06	-3.6464+06	-3.6464+06	-3.6506+06
WATT	-3.3564+05	-8.2898+04	-1.5201+05	-1.5201+05	-4.3482+04
ENTROPY:					
J/KMOL-K	-6.2953+04	3.4897+04	3.3193+04	4.2971+04	4.1847+04
J/KG-K	-2.3441+04	1245.8932	1185.0432	1534.1477	1494.0189
DENSITY:					
KMOL/CUM	4.9698	0.9375	1.1102	0.2988	0.3116
KG/CUM	13.3468	26.2616	31.0968	8.3699	8.7280
AVG MW	2.6855	28.0097	28.0097	28.0097	28.0097

Simulated design streams

STREAM ID	L44	L46	L47	L48	L49
FROM :	T22MIX	E25-27	T22	MIX2	T22
TO :	E25-27	T22	V47	E21	E21
SUBSTREAM: MIXED					
PHASE:	MIXED	LIQUID	LIQUID	MIXED	VAPOR
COMPONENTS: KMOL/HR					
H2	5.9328-04	5.9328-04	1.6328-15	3.5302	1.5812-03
CO	23.1337	23.1337	2.0378	6.7543	52.3228
CO2	0.0	0.0	0.0	0.0	0.0
N2	4.2319	4.2319	4.0712-02	0.9805	7.7067
O2	0.0	0.0	0.0	0.0	0.0
CH4	1.0708-08	1.0708-08	4.5204	4.5586	6.5505-08
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	27.3662	27.3662	6.5990	15.8237	60.0311
KG/HR	766.5249	766.5249	130.7428	296.9083	1681.4545
CUM/SEC	8.4820-03	2.8213-04	6.7915-05	9.1534-03	5.1031-02
STATE VARIABLES:					
TEMP C	-183.2570	-184.5000	-172.4673	-180.6837	-184.1620
PRES BAR	2.5000	2.3250	2.2500	2.1500	2.2500
VFRAC	0.3952	0.0	0.0	0.5971	1.0000
LFRAC	0.6047	1.0000	1.0000	0.4028	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0315+08	-1.0546+08	-9.9581+07	-7.8128+07	-1.0264+08
J/KG	-3.6826+06	-3.7649+06	-5.0262+06	-4.1638+06	-3.6645+06
WATT	-7.8411+05	-8.0165+05	-1.8254+05	-3.4341+05	-1.7116+06
ENTROPY:					
J/KMOL-K	-2121.9147	-2.7792+04	-1.3265+05	-4.7083+04	3.8596+04
J/KG-K	-75.7558	-992.2272	-6695.2605	-2509.2788	1377.9520
DENSITY:					
KMOL/CUM	0.8962	26.9439	26.9907	0.4802	0.3267
KG/CUM	25.1029	754.6960	534.7511	9.0103	9.1527
AVG MW	28.0099	28.0099	19.8124	18.7634	28.0097

Simulated design streams

STREAM ID	L6	L7	L8	L9	LB
FROM :	T21	V29	T23	SPLIT2	T21
TO :	V6	MIX2	V8	V9	MIX1
SUBSTREAM: MIXED					
PHASE:	LIQUID	VAPOR	LIQUID	LIQUID	VAPOR
COMPONENTS: KMOL/HR					
H2	2.3053	2.3043	9.8799-04	1.2262	149.5015
CO	34.6148	3.3870	31.2277	15.9099	64.7157
CO2	0.0	0.0	0.0	0.0	0.0
N2	4.0705	0.5549	3.5156	3.3569	14.2542
O2	0.0	0.0	0.0	0.0	0.0
CH4	4.5586	3.8212-02	4.5204	0.0	5.2456-13
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	45.5493	6.2845	39.2648	20.4931	228.4715
KG/HR	1161.3720	115.6755	1045.6965	542.1473	2513.3874
CUM/SEC	5.1430-04	5.5388-03	4.3456-04	2.3577-04	1.7132-02
STATE VARIABLES:					
TEMP C	-169.6737	-182.2501	-172.1917	-171.5000	-171.2848
PRES BAR	27.3000	2.3000	5.4000	27.3470	27.3000
VFRAC	0.0	1.0000	0.0	0.0	1.0000
LFRAC	1.0000	0.0	1.0000	1.0000	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0262+08	-6.5992+07	-1.0828+08	-9.6519+07	-3.7133+07
J/KG	-4.0247+06	-3.5853+06	-4.0657+06	-3.6484+06	-3.3754+06
WATT	-1.2984+06	-1.1520+05	-1.1810+06	-5.4944+05	-2.3566+06
ENTROPY:					
J/KMOL-K	-3.1517+04	1.5173+04	-3.3914+04	-2.1784+04	-2.7210+04
J/KG-K	-1236.1190	824.3378	-1273.4399	-823.4156	-2473.4417
DENSITY:					
KMOL/CUM	24.6015	0.3151	25.0988	24.1446	3.7044
KG/CUM	627.2653	5.8012	668.4296	638.7475	40.7524
AVG MW	25.4970	18.4064	26.6318	26.4550	11.0008

Simulated design streams

STREAM ID	LC	LD	LE	LF	LG
FROM :	CO1	CO1	CO2	MIX1	SPLIT2
TO :	CO2	SPLIT2	MIX1	CO1	T21
SUBSTREAM: MIXED					
PHASE:	VAPOR	LIQUID	LIQUID	MIXED	LIQUID
COMPONENTS: KMOL/HR					
H2	148.3965	4.7826	3.6776	153.1792	3.5563
CO	61.1690	62.0511	58.5043	123.2201	46.1412
CO2	0.0	0.0	0.0	0.0	0.0
N2	16.7122	13.0926	15.5506	29.8049	9.7357
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	0.0	0.0	5.2456-13	0.0
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	226.2778	79.9264	77.7327	306.2042	59.4333
KG/HR	2480.6710	2114.4593	2081.7430	4595.1304	1572.3119
CUM/SEC	1.6903-02	9.1953-04	7.1708-04	1.6170-02	6.8377-04
STATE VARIABLES:					
TEMP C	-171.5000	-171.5000	-201.5000	-174.9585	-171.5000
PRES BAR	27.3470	27.3470	27.3000	27.3000	27.3470
VFRAC	1.0000	0.0	0.0	0.6639	0.0
LFRAC	0.0	1.0000	1.0000	0.3360	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-3.5707+07	-9.6519+07	-9.5757+07	-5.2015+07	-9.6519+07
J/KG	-3.2571+06	-3.6484+06	-3.5756+06	-3.4661+06	-3.6484+06
WATT	-2.2444+06	-2.1429+06	-2.0676+06	-4.4242+06	-1.5935+06
ENTROPY:					
J/KMOL-K	-2.8312+04	-2.1784+04	-4.4861+04	-3.0948+04	-2.1784+04
J/KG-K	-2582.4909	-823.4156	-1675.1372	-2062.2539	-823.4156
DENSITY:					
KMOL/CUM	3.7186	24.1446	30.1117	5.2602	24.1446
KG/CUM	40.7669	638.7475	806.4147	78.9395	638.7475
AVG MW	10.9629	26.4550	26.7807	15.0067	26.4550

Simulated design streams

STREAM ID	LIQ	MEASUMP	OUT	STM	STMFD
FROM :	V2103	V2116	V2118	V2103	DRUMMIX
TO :	SPLIT	----	----	----	V2103
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	MIXED	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	0.0	1.6653-07	0.8948	0.0	0.0
CO	0.0	3.6150-08	0.3210	0.0	0.0
CO2	0.0	6.6366-09	1.1361-02	0.0	0.0
N2	0.0	5.3679-09	5.1846-02	0.0	0.0
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	3.2820-08	2.7517-02	0.0	0.0
C2H6	0.0	0.0	7.6395-05	0.0	0.0
C3H8	0.0	0.0	3.1213-09	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	1.1001-06	4.0328-03	0.0	0.0
H2O	7810.2630	0.1822	0.3701	343.2634	7810.2625
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	7810.2630	0.1822	1.6809	343.2634	7810.2625
KG/HR	1.4070+05	3.2823	20.1075	6183.8913	1.4070+05
CUM/SEC	6.4892-02	1.2207-06	3.3726-04	0.1473	0.2227
STATE VARIABLES:					
TEMP C	221.0000	40.0000	35.0000	221.0000	221.0000
PRES BAR	24.0094	10.3000	28.3000	24.0094	24.0094
VFRAC	0.0	0.0	0.7805	1.0000	4.8000-02
LFRAC	1.0000	1.0000	0.2194	0.0	0.9520
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7130+08	-2.8770+08	-8.8685+07	-2.3648+08	-2.6962+08
J/KG	-1.5059+07	-1.5970+07	-7.4137+06	-1.3127+07	-1.4967+07
WATT	-5.8858+08	-1.4560+04	-4.1409+04	-2.2548+07	-5.8496+08
ENTROPY:					
J/KMOL-K	-1.2552+05	-1.6656+05	-3.7930+04	-5.5051+04	-1.2213+05
J/KG-K	-6967.2642	-9245.2471	-3170.7566	-3055.8504	-6779.5160
DENSITY:					
KMOL/CUM	33.4325	41.4597	1.3844	0.6470	9.7406
KG/CUM	602.2870	746.9080	16.5613	11.6559	175.4781
AVG MW	18.0150	18.0152	11.9623	18.0150	18.0150

APPENDIX 5. Most important simulated in and out going actual streams

Table A5.1 Most important streams entering the plant

STREAM	201	205	251	253	260	BIN	281
Comp.							
H ₂	0.0	0.0	0.0	0.0	0.0	125.347	0.0
CO	0.0	0.0	0.0	0.0	0.0	2.371	0.0
CO ₂	0.486	0.0	0.0	0.0	0.0	0.0	9.859
N ₂	7.839	0.0	0.0	0.0	0.0	268.204	0.0
O ₂	0.0	0.0	0.0	0.0	0.0	73.446	0.0
CH ₄	44.431	0.0	0.0	0.0	0.0	0.025	0.0
C ₂ H ₆	1.556	0.0	0.0	0.0	0.0	0.0	0.0
C ₃ H ₈	0.204	0.0	0.0	0.0	0.0	0.0	0.0
C ₄ H ₁₀	0.076	0.0	0.0	0.0	0.0	0.0	0.0
C ₅ H ₁₂	0.021	0.0	0.0	0.0	0.0	0.0	0.0
C ₆ H ₁₄	0.028	0.0	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0	0.0	89.643
H ₂ O	0.0	155.659	343.264	25.762	105.484	0.0	797.674
AR	0.0	0.0	0.0	0.0	0.0	3.212	0.0
kmol/h	54.641	155.659	343.264	25.762	105.484	472.605	897.176
kg/h	1017.9	2804.2	6183.9	464.1	1900.3	10311.0	20279.0
T (°C)	15.0	220.0	104.0	104.0	208.0	40.0	40.0
P (bar)	19.0	23.2	27.0	27.0	18.3	2.0	14.7

APPENDIX 5. Most important simulated in and out going actual streams

Table A5.2 Most important streams going out the plant

STREAM	226	STM	266	7	140	281A
Comp.						
H ₂	0.001	0.0	0.0	0.0	143.472	0.0
CO	45.024	0.0	0.0	0.0	2.6401	0.0
CO ₂	0.0	0.0	0.021	2.396	0.0	9.450
N ₂	6.632	0.0	0.0	268.204	1.154	0.0
O ₂	0.0	0.0	0.0	9.537	0.0	0.0
CH ₄	0.0	0.0	0.0	0.0	0.0	0.0
C ₂ H ₆	0.0	0.0	0.0	0.0	0.0	0.0
C ₃ H ₈	0.0	0.0	0.0	0.0	0.0	0.0
C ₄ H ₁₀	0.0	0.0	0.0	0.0	0.0	0.0
C ₅ H ₁₂	0.0	0.0	0.0	0.0	0.0	0.0
C ₆ H ₁₄	0.0	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0	89.631
H ₂ O	0.0	343.264	236.238	125.397	0.0	797.674
AR	0.0	0.0	0.0	3.2112	0.0	0.0
kmol/h	51.057	343.264	236.259	408.746	147.267	897.176
kg/h	1446.9	6183.9	4256.8	10311.0	395.5	20279.0
T (°C)	90.0	221.0	40.2	138.5	13.8	40.0
P (bar)	9.6	24.0	3.5	1.5	3.5	14.7

APPENDIX 5. Most important simulated in and out going streams with less N₂Table A5.3 Most important entering streams (less N₂ simulation)

STREAM	201	205	251	253	260	BIN	281
Comp.							
H ₂	0.0	0.0	0.0	0.0	0.0	125.347	0.0
CO	0.0	0.0	0.0	0.0	0.0	2.371	0.0
CO ₂	0.486	0.0	0.0	0.0	0.0	0.0	9.859
N ₂	1.556	0.0	0.0	0.0	0.0	268.204	0.0
O ₂	0.0	0.0	0.0	0.0	0.0	73.446	0.0
CH ₄	44.431	0.0	0.0	0.0	0.0	0.025	0.0
C ₂ H ₆	1.556	0.0	0.0	0.0	0.0	0.0	0.0
C ₃ H ₈	0.204	0.0	0.0	0.0	0.0	0.0	0.0
C ₄ H ₁₀	0.076	0.0	0.0	0.0	0.0	0.0	0.0
C ₅ H ₁₂	0.021	0.0	0.0	0.0	0.0	0.0	0.0
C ₆ H ₁₄	0.028	0.0	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0	0.0	89.643
H ₂ O	0.0	155.659	343.264	25.762	105.484	0.0	797.674
AR	0.0	0.0	0.0	0.0	0.0	3.212	0.0
kmol/h	54.641	155.659	343.264	25.762	105.484	472.605	897.176
kg/h	1017.9	2804.2	6183.9	464.1	1900.3	10311.0	20279.0
T (°C)	15.0	220.0	104.0	104.0	208.0	40.0	40.0
P (bar)	19.0	23.2	27.0	27.0	18.3	2.0	14.7

APPENDIX 5. Most important simulated in and out going streams with less N₂Table A5.4 Most important out going streams (less N₂ simulation)

STREAM	226	STM	266	7	140	281A
Comp.						
H ₂	0.001	0.0	0.0	0.0	143.227	0.0
CO	44.764	0.0	0.0	0.0	3.154	0.0
CO ₂	0.0	0.0	0.022	2.396	0.0	9.205
N ₂	1.279	0.0	0.0	268.204	0.267	0.0
O ₂	0.0	0.0	0.0	9.537	0.0	0.0
CH ₄	0.0	0.0	0.0	0.0	0.0	0.0
C ₂ H ₆	0.0	0.0	0.0	0.0	0.0	0.0
C ₃ H ₈	0.0	0.0	0.0	0.0	0.0	0.0
C ₄ H ₁₀	0.0	0.0	0.0	0.0	0.0	0.0
C ₅ H ₁₂	0.0	0.0	0.0	0.0	0.0	0.0
C ₆ H ₁₄	0.0	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0	89.643
H ₂ O	0.0	343.264	236.529	125.397	0.0	797.674
AR	0.0	0.0	0.0	3.2112	0.0	0.0
kmol/h	46.043	343.264	236.551	408.746	146.648	897.176
kg/h	1289.7	6183.9	4262.0	10311.0	384.6	20279.0
T (°C)	90.0	221.0	40.2	148.8	15.2	40.0
P (bar)	9.6	24.0	3.5	1.5	3.5	14.7

Simulated capacity increase by using CO₂.

S'TREAM ID	1	140	2	201	202
FROM :	BURNER	B1	H2102	----	B1
TO :	H2102	----	H2103	H2-MIX	H2-MIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	3.8182-05	117.1113	3.8182-05	0.0	48.1150
CO	1.2933-06	2.2539	1.2933-06	0.0	0.9260
CO2	2.3955	0.0	2.3955	14.6080	0.0
N2	268.2040	0.7733	268.2040	7.1788	0.3177
O2	9.5373	0.0	9.5373	0.0	0.0
CH4	8.0301-31	0.0	8.0301-31	40.7093	0.0
C2H6	0.0	0.0	0.0	1.4266	0.0
C3H8	0.0	0.0	0.0	0.1848	0.0
C4H10	0.0	0.0	0.0	6.9851-02	0.0
C5H12	0.0	0.0	0.0	1.9404-02	0.0
C6H14	0.0	0.0	0.0	2.4600-02	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	125.3970	0.0	125.3970	0.0	0.0
AR	3.2116	0.0	3.2116	0.0	0.0
TOTAL FLOW:					
KMOL/HR	408.7456	120.1386	408.7456	64.2215	49.3588
KG/HR	1.0311+04	320.8922	1.0311+04	1555.7300	131.8382
CUM/SEC	8.0465	0.2279	7.6872	2.1459-02	9.3644-02
STATE VARIABLES:					
TEMP C	1005.0000	13.7301	947.9141	15.0000	13.7301
PRES BAR	1.5000	3.5000	1.5000	19.0000	3.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-4.3557+07	-2.3930+06	-4.5655+07	-1.4026+08	-2.3930+06
J/KG	-1.7266+06	-8.9590+05	-1.8098+06	-5.7901+06	-8.9590+05
WATT	-4.9455+06	-7.9858+04	-5.1837+06	-2.5022+06	-3.2809+04
ENTROPY:					
J/KMOL-K	3.7540+04	-8630.1778	3.5860+04	-7.4323+04	-8630.1778
J/KG-K	1488.1121	-3231.0463	1421.5486	-3068.0872	-3231.0463
DENSITY:					
KMOL/CUM	1.4110-02	0.1464	1.4770-02	0.8313	0.1464
KG/CUM	0.3559	0.3910	0.3725	20.1380	0.3910
AVG MW	25.2263	2.6710	25.2263	24.2244	2.6710

Simulated capacity increase by using CO₂.

STREAM ID	202B	203	203B	204	204B
FROM :	H2-MIX	H2104	V2101	V2114	CO2-MIX
TO :	H2104	V2101	CO2-MIX	CO2-MIX	H2O-MIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	48.1150	48.1150	48.1150	4.2859	52.4010
CO	0.9260	0.9260	0.9260	8.0093	8.9353
CO2	14.6080	14.6080	14.6080	35.9402	50.5482
N2	7.4965	7.4965	7.4965	0.9671	8.4636
O2	0.0	0.0	0.0	0.0	0.0
CH4	40.7093	40.7093	40.7093	5.9294	46.6388
C2H6	1.4266	1.4266	1.4266	1.5236-10	1.4266
C3H8	0.1848	0.1848	0.1848	0.0	0.1848
C4H10	6.9851-02	6.9851-02	6.9851-02	0.0	6.9851-02
C5H12	1.9404-02	1.9404-02	1.9404-02	0.0	1.9404-02
C6H14	2.4600-02	2.4600-02	2.4600-02	0.0	2.4600-02
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.2357	0.2357
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	113.5803	113.5803	113.5803	55.3678	168.9482
KG/HR	1687.5682	1687.5682	1687.5682	1941.1781	3628.7465
CUM/SEC	3.5311-02	8.5969-02	9.0440-02	2.3763-02	0.1263
STATE VARIABLES:					
TEMP C	12.5537	400.0000	350.0000	40.0000	249.0306
PRES BAR	21.0000	20.7000	18.2000	16.2000	16.2000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-8.0348+07	-6.5514+07	-6.7641+07	-2.8052+08	-1.3741+08
J/KG	-5.4077+06	-4.4094+06	-4.5525+06	-8.0012+06	-6.3974+06
WATT	-2.5350+06	-2.0670+06	-2.1341+06	-4.3144+06	-6.4485+06
ENTROPY:					
J/KMOL-K	-4.7244+04	-1.4902+04	-1.7106+04	-7101.7192	-9427.6211
J/KG-K	-3179.7127	-1002.9696	-1151.2780	-202.5611	-438.9339
DENSITY:					
KMOL/CUM	0.8935	0.3670	0.3488	0.6472	0.3714
KG/CUM	13.2755	5.4528	5.1831	22.6910	7.9776
AVG MW	14.8579	14.8579	14.8579	35.0596	21.4784

Simulated capacity increase by using CO₂.

STREAM ID	205	206	207	208	208B
FROM :	----	H2O-MIX	H2103	H2101	H2107
TO :	H2O-MIX	H2103	H2101	H2107	H2O-MIX2
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	0.0	52.4010	52.4010	170.5384	170.5384
CO	0.0	8.9353	8.9353	67.0124	67.0124
CO2	0.0	50.5482	50.5482	37.0764	37.0764
N2	0.0	8.4636	8.4636	8.4636	8.4636
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	46.6388	46.6388	5.9652	5.9652
C2H6	0.0	1.4266	1.4266	1.1303-04	1.1303-04
C3H8	0.0	0.1848	0.1848	5.2530-09	5.2530-09
C4H10	0.0	6.9851-02	6.9851-02	3.3607-13	3.3607-13
C5H12	0.0	1.9404-02	1.9404-02	1.4485-17	1.4485-17
C6H14	0.0	2.4600-02	2.4600-02	7.1454-22	7.1454-22
MEA	0.0	0.0	0.0	1.2575-30	1.2575-30
H2O	143.9411	144.1768	144.1768	113.0435	113.0435
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	143.9411	312.8894	312.8894	402.0998	402.0998
KG/HR	2593.1000	6221.8465	6221.8465	6221.8331	6221.8331
CUM/SEC	6.4031-02	0.2289	0.3636	0.9172	0.4518
STATE VARIABLES:					
TEMP C	220.0000	228.9851	487.0000	860.0000	286.1904
PRES BAR	23.2000	15.6000	15.1000	11.5000	11.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.3647+08	-1.8298+08	-1.7251+08	-9.5110+07	-1.1556+08
J/KG	-1.3126+07	-9.2018+06	-8.6751+06	-6.1467+06	-7.4685+06
WATT	-9.4550+06	-1.5903+07	-1.4993+07	-1.0623+07	-1.2908+07
ENTROPY:					
J/KMOL-K	-5.4783+04	-2.3071+04	-6026.3997	3.7849+04	1.2863+04
J/KG-K	-3040.9421	-1160.2344	-303.0606	2446.0770	831.3111
DENSITY:					
KMOL/CUM	0.6244	0.3796	0.2389	0.1217	0.2471
KG/CUM	11.2492	7.5484	4.7522	1.8842	3.8246
AVG MW	18.0150	19.8851	19.8851	15.4733	15.4733

Simulated capacity increase by using CO₂.

STREAM ID	208C	209	209B	210	210B
FROM :	H20-MIX2	M2102	H2110A	V2107	H2111
TO :	M2102	H2110A	V2107	H2111	V2108
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	MIXED	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	170.5384	170.5384	170.5384	170.5382	170.5382
CO	67.0124	67.0124	67.0124	67.0124	67.0124
CO2	37.0764	37.0764	37.0764	37.0518	37.0518
N2	8.4636	8.4636	8.4636	8.4636	8.4636
O2	0.0	0.0	0.0	0.0	0.0
CH4	5.9652	5.9652	5.9652	5.9651	5.9651
C2H6	1.1303-04	1.1303-04	1.1303-04	1.1303-04	1.1303-04
C3H8	5.2530-09	5.2530-09	5.2530-09	5.2530-09	5.2530-09
C4H10	3.3607-13	3.3607-13	3.3607-13	0.0	0.0
C5H12	1.4485-17	1.4485-17	1.4485-17	0.0	0.0
C6H14	7.1454-22	7.1454-22	7.1454-22	0.0	0.0
MEA	1.2575-30	1.2575-30	1.2575-30	0.0	0.0
H2O	231.8889	269.7185	269.7185	1.6645	1.6645
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	520.9452	558.7748	558.7748	290.6959	290.6959
KG/HR	8362.8331	9044.3331	9044.3331	4214.2579	4214.2579
CUM/SEC	0.5839	0.5098	0.3428	0.1970	0.2027
STATE VARIABLES:					
TEMP C	263.3128	167.7082	132.0000	40.0000	40.0000
PRES BAR	11.0000	11.0000	10.7000	10.7000	10.4000
VFRAC	1.0000	1.0000	0.7026	1.0000	1.0000
LFRAC	0.0	0.0	0.2973	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.4319+08	-1.5260+08	-1.6591+08	-7.8193+07	-7.8193+07
J/KG	-8.9198+06	-9.4276+06	-1.0250+07	-5.3937+06	-5.3937+06
WATT	-2.0721+07	-2.3685+07	-2.5752+07	-6.3140+06	-6.3140+06
ENTROPY:					
J/KMOL-K	912.0006	-8393.1186	-4.0026+04	1.0266+04	1.0506+04
J/KG-K	56.8111	-518.5416	-2472.8814	708.1352	724.6790
DENSITY:					
KMOL/CUM	0.2478	0.3044	0.4527	0.4098	0.3983
KG/CUM	3.9782	4.9276	7.3282	5.9412	5.7751
AVG MW	16.0531	16.1860	16.1860	14.4971	14.4971

Simulated capacity increase by using CO₂.

STREAM ID	211	212	213	213B	214
FROM :	V2108	V2105	V2106	H2112	VENTTANK
TO :	V2105	V2116	H2112	V2109	----
SUBSTREAM: MIXED					
PHASE:	VAPOR	MIXED	VAPOR	MIXED	MIXED
COMPONENTS: KMOL/HR					
H2	170.5382	170.5367	1.4075-03	1.4075-03	0.0
CO	67.0124	67.0112	1.1254-03	1.1254-03	0.0
CO2	37.0518	1.1361-02	37.7641	37.7641	1.7970
N2	8.4636	8.4636	5.2169-05	5.2169-05	2.5219-06
O2	0.0	0.0	0.0	0.0	0.0
CH4	5.9651	5.9651	2.5065-05	2.5065-05	0.0
C2H6	1.1303-04	1.1303-04	1.5235-10	1.5235-10	0.0
C3H8	5.2530-09	5.2530-09	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	4.7331-03	6.3647-05	6.3647-05	4.7432-05
H2O	1.6645	1.6962	270.8446	270.8446	7.3508-02
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	290.6959	253.6892	308.6114	308.6114	1.8706
KG/HR	4214.2579	2584.9283	6541.3056	6541.3056	80.4149
CUM/SEC	0.2027	0.1788	1.5941	0.1899	8.9584-03
STATE VARIABLES:					
TEMP C	40.0000	40.0419	110.9508	40.0000	40.0000
PRES BAR	10.4000	10.3000	1.7000	1.5000	1.5000
VFRAC	1.0000	0.9999	1.0000	0.1272	1.0000
LFRAC	0.0	2.3858-05	0.0	0.8727	2.0498-06
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-7.8193+07	-3.2192+07	-2.5768+08	-3.0040+08	-3.8730+08
J/KG	-5.3937+06	-3.1594+06	-1.2157+07	-1.4172+07	-9.0092+06
WATT	-6.3140+06	-2.2686+06	-2.2090+07	-2.5752+07	-2.0124+05
ENTROPY:					
J/KMOL-K	1.0506+04	1.0685+04	-3.1191+04	-1.4520+05	894.0208
J/KG-K	724.6790	1048.6483	-1471.5545	-6850.6006	20.7965
DENSITY:					
KMOL/CUM	0.3983	0.3939	5.3776-02	0.4513	5.8003-02
KG/CUM	5.7751	4.0140	1.1398	9.5665	2.4934
AVG MW	14.4971	10.1893	21.1959	21.1959	42.9889

Simulated capacity increase by using CO₂.

STREAM ID	215	215B	216	216B	216C
FROM :	V2109	VENTTANK	CO-MIX	K2103-1	H2113
TO :	VENTTANK	CO-MIX	K2103-1	H2113	V2113
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	1.4076-03	1.4076-03	4.2860	4.2860	4.2860
CO	1.1256-03	1.1256-03	8.0094	8.0094	8.0094
CO2	37.7378	35.9407	35.9407	35.9407	35.9407
N2	5.2185-05	4.9663-05	0.9671	0.9671	0.9671
O2	0.0	0.0	0.0	0.0	0.0
CH4	2.6190-05	2.6190-05	5.9293	5.9293	5.9293
C2H6	1.5235-10	1.5235-10	1.5235-10	1.5235-10	1.5235-10
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	4.7432-05	0.0	0.0	0.0	0.0
H2O	1.5436	1.4701	1.4701	1.4701	1.4701
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	39.2841	37.4135	56.6029	56.6029	56.6029
KG/HR	1688.6899	1608.2749	1963.4422	1963.4422	1963.4422
CUM/SEC	0.1881	0.1791	0.2660	5.6590-02	4.1753-02
STATE VARIABLES:					
TEMP C	40.0000	40.0000	33.4351	145.0000	40.0000
PRES BAR	1.5000	1.5000	1.5000	9.6000	9.4000
VFRAC	1.0000	1.0000	1.0000	1.0000	0.9807
LFRAC	0.0	0.0	0.0	0.0	1.9302-02
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-3.8728+08	-3.8728+08	-2.7955+08	-2.7551+08	-2.8040+08
J/KG	-9.0093+06	-9.0093+06	-8.0591+06	-7.9427+06	-8.0834+06
WATT	-4.2261+06	-4.0249+06	-4.3955+06	-4.3319+06	-4.4087+06
ENTROPY:					
J/KMOL-K	916.8939	917.9625	1.2255+04	8120.0928	-5391.0193
J/KG-K	21.3297	21.3547	353.2864	234.0894	-155.4145
DENSITY:					
KMOL/CUM	5.8003-02	5.8003-02	5.9089-02	0.2778	0.3765
KG/CUM	2.4933	2.4933	2.0496	9.6376	13.0626
AVG MW	42.9865	42.9864	34.6879	34.6879	34.6879

Simulated capacity increase by using CO₂.

STREAM ID	216D	216E	216F	218	218B
FROM :	V2113	K2103-3	H2114	V2116	K2104
TO :	K2103-3	H2114	V2114	K2104	H2115
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	MIXED	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	4.2859	4.2859	4.2859	170.5367	170.5367
CO	8.0093	8.0093	8.0093	67.0112	67.0112
CO2	35.9403	35.9403	35.9403	1.1361-02	1.1361-02
N2	0.9671	0.9671	0.9671	8.4636	8.4636
O2	0.0	0.0	0.0	0.0	0.0
CH4	5.9294	5.9294	5.9294	5.9651	5.9651
C2H6	1.5236-10	1.5236-10	1.5236-10	1.1303-04	1.1303-04
C3H8	0.0	0.0	0.0	5.2530-09	5.2530-09
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	4.7318-03	4.7318-03
H2O	0.3709	0.3709	0.3709	1.4833	1.4833
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	55.5031	55.5031	55.5031	253.4764	253.4764
KG/HR	1943.6180	1943.6180	1943.6180	2581.0934	2581.0934
CUM/SEC	4.0851-02	3.4504-02	2.3764-02	0.1789	8.7153-02
STATE VARIABLES:					
TEMP C	40.0000	145.0000	40.0000	40.0000	145.0000
PRES BAR	9.6000	15.4000	16.2000	10.3000	28.5000
VFRAC	1.0000	1.0000	0.9975	1.0000	1.0000
LFRAC	0.0	0.0	2.4377-03	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.8026+08	-2.7633+08	-2.8054+08	-3.2018+07	-2.8928+07
J/KG	-8.0034+06	-7.8910+06	-8.0112+06	-3.1443+06	-2.8409+06
WATT	-4.3210+06	-4.2603+06	-4.3252+06	-2.2544+06	-2.0368+06
ENTROPY:					
J/KMOL-K	-2405.2237	4538.4885	-7490.1506	1.0697+04	1.0658+04
J/KG-K	-68.6850	129.6039	-213.8934	1050.5261	1046.6972
DENSITY:					
KMOL/CUM	0.3774	0.4468	0.6487	0.3935	0.8078
KG/CUM	13.2162	15.6471	22.7187	4.0070	8.2265
AVG MW	35.0181	35.0181	35.0181	10.1827	10.1827

Simulated capacity increase by using CO₂.

STREAM ID	218C	219	220	224	224A
FROM :	H2115	V2117	V2118	E21	K2105-1
TO :	V2117	V2118	E21	K2105-1	H2118
SUBSTREAM: MIXED					
PHASE:	MIXED	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	170.5367	170.5367	169.5135	2.4996-03	2.4996-03
CO	67.0112	67.0112	66.6092	66.2501	66.2501
CO2	1.1361-02	1.1361-02	0.0	0.0	0.0
N2	8.4636	8.4636	8.4128	7.5964	7.5964
O2	0.0	0.0	0.0	0.0	0.0
CH4	5.9651	5.9651	5.9293	1.0916-07	1.0916-07
C2H6	1.1303-04	1.1303-04	0.0	0.0	0.0
C3H8	5.2530-09	5.2530-09	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	4.7318-03	4.7153-03	0.0	0.0	0.0
H2O	1.4833	0.4329	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	253.4764	252.4259	250.4649	73.8490	73.8490
KG/HR	2581.0934	2562.1681	2538.2566	2068.4694	2068.4694
CUM/SEC	6.4229-02	6.4446-02	6.3956-02	0.2450	0.1063
STATE VARIABLES:					
TEMP C	35.0000	35.0000	35.0000	21.4353	100.0000
PRES BAR	28.4000	28.3000	28.3000	2.0500	6.0000
VFRAC	0.9958	1.0000	1.0000	1.0000	1.0000
LFRAC	4.1497-03	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-3.2365+07	-3.1300+07	-3.0918+07	-9.9352+07	-9.7060+07
J/KG	-3.1784+06	-3.0837+06	-3.0508+06	-3.5471+06	-3.4653+06
WATT	-2.2788+06	-2.1947+06	-2.1511+06	-2.0381+06	-1.9911+06
ENTROPY:					
J/KMOL-K	1089.6573	1824.0174	1855.3902	7.6998+04	7.4954+04
J/KG-K	107.0098	179.7029	183.0824	2748.9953	2676.0391
DENSITY:					
KMOL/CUM	1.0962	1.0880	1.0878	8.3700-02	0.1929
KG/CUM	11.1627	11.0436	11.0243	2.3443	5.4029
AVG MW	10.1827	10.1501	10.1341	28.0094	28.0094

Simulated capacity increase by using CO₂.

STREAM ID	224B	224C	224D	225	226
FROM :	H2118	K2105-2	H2119	CO-SPLIT	CO-SPLIT
TO :	K2105-2	H2119	CO-SPLIT	E21	----
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	2.4996-03	2.4996-03	2.4996-03	4.0861-04	2.0910-03
CO	66.2501	66.2501	66.2501	10.8299	55.4202
CO2	0.0	0.0	0.0	0.0	0.0
N2	7.5964	7.5964	7.5964	1.2417	6.3546
O2	0.0	0.0	0.0	0.0	0.0
CH4	1.0916-07	1.0916-07	1.0916-07	1.7844-08	9.1314-08
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	73.8490	73.8490	73.8490	12.0721	61.7769
KG/HR	2068.4694	2068.4694	2068.4694	338.1327	1730.3367
CUM/SEC	8.9185-02	6.4769-02	5.5978-02	9.1507-03	4.6827-02
STATE VARIABLES:					
TEMP C	30.0000	90.0000	35.0000	35.0000	35.0000
PRES BAR	5.8000	9.6000	9.4000	9.4000	9.4000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-9.9125+07	-9.7368+07	-9.8999+07	-9.8999+07	-9.8999+07
J/KG	-3.5390+06	-3.4762+06	-3.5345+06	-3.5345+06	-3.5345+06
WATT	-2.0334+06	-1.9974+06	-2.0308+06	-3.3198+05	-1.6989+06
ENTROPY:					
J/KMOL-K	6.9108+04	7.0199+04	6.5502+04	6.5502+04	6.5502+04
J/KG-K	2467.3191	2506.2708	2338.5616	2338.5616	2338.5616
DENSITY:					
KMOL/CUM	0.2300	0.3167	0.3664	0.3664	0.3664
KG/CUM	6.4424	8.8711	10.2643	10.2643	10.2643
AVG MW	28.0094	28.0094	28.0094	28.0094	28.0094

Simulated capacity increase by using CO₂.

STREAM ID	227	228	251	251B	253
FROM :	E21	E21	----	H2106	----
TO :	CO-MIX	B1	H2106	V2103	M2102
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	LIQUID	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	4.2846	165.2264	0.0	0.0	0.0
CO	8.0083	3.1799	0.0	0.0	0.0
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.9671	1.0910	0.0	0.0	0.0
O2	0.0	0.0	0.0	0.0	0.0
CH4	5.9293	0.0	0.0	0.0	0.0
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	343.2639	343.2639	37.8295
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	19.1894	169.4974	343.2639	343.2639	37.8295
KG/HR	355.1672	452.7304	6183.9000	6183.9000	681.5000
CUM/SEC	6.6232-02	0.3215	2.4282-03	2.6602-03	2.6760-04
STATE VARIABLES:					
TEMP C	18.3697	13.7301	104.0000	180.0000	104.0000
PRES BAR	1.9500	3.5000	27.0000	26.8000	27.0000
VFRAC	1.0000	1.0000	0.0	0.0	0.0
LFRAC	0.0	0.0	1.0000	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.9526+07	-2.3930+06	-2.8210+08	-2.7526+08	-2.8210+08
J/KG	-3.7564+06	-8.9590+05	-1.5659+07	-1.5279+07	-1.5659+07
WATT	-3.7060+05	-1.1267+05	-2.6899+07	-2.6246+07	-2.9644+06
ENTROPY:					
J/KMOL-K	1.6443+04	-8630.1778	-1.5042+05	-1.3390+05	-1.5042+05
J/KG-K	888.4260	-3231.0463	-8349.6409	-7432.5015	-8349.6409
DENSITY:					
KMOL/CUM	8.0481-02	0.1464	39.2685	35.8435	39.2685
KG/CUM	1.4895	0.3910	707.4237	645.7212	707.4237
AVG MW	18.5085	2.6710	18.0150	18.0150	18.0150

Simulated capacity increase by using CO₂.

STREAM ID	260	264	265	266	268
FROM :	----	V2107	V2108	COND-MIX	V2113
TO :	H2O-MIX2	COND-MIX	COND-MIX	----	CONDMIX2
SUBSTREAM: MIXED					
PHASE:	VAPOR	LIQUID	MIXED	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	0.0	2.1804-04	0.0	2.1804-04	1.0852-07
CO	0.0	5.1676-05	0.0	5.1676-05	1.2097-07
CO2	0.0	2.4515-02	0.0	2.4515-02	4.5265-04
N2	0.0	6.0036-06	0.0	6.0036-06	1.3453-08
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	4.8563-05	0.0	4.8563-05	9.3388-07
C2H6	0.0	1.3738-09	0.0	1.3738-09	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	118.8454	268.0540	0.0	268.0540	1.0992
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	118.8454	268.0788	0.0	268.0788	1.0996
KG/HR	2141.0000	4830.0751	0.0	4830.0751	19.8221
CUM/SEC	6.6487-02	1.7962-03	0.0	1.7966-03	7.3693-06
STATE VARIABLES:					
TEMP C	208.0000	40.0000	MISSING	40.1538	40.0000
PRES BAR	18.3000	10.7000	10.4000	3.5000	9.6000
VFRAC	1.0000	0.0	MISSING	0.0	0.0
LFRAC	0.0	1.0000	MISSING	1.0000	1.0000
SFRAC	0.0	0.0	MISSING	0.0	0.0
ENTHALPY:					
J/KMOL	-2.3667+08	-2.8771+08	MISSING	-2.8771+08	-2.8774+08
J/KG	-1.3137+07	-1.5968+07	MISSING	-1.5968+07	-1.5963+07
WATT	-7.8130+06	-2.1424+07	MISSING	-2.1424+07	-8.7893+04
ENTROPY:					
J/KMOL-K	-5.3384+04	-1.6654+05	MISSING	-1.6648+05	-1.6648+05
J/KG-K	-2963.3323	-9243.0553	MISSING	-9239.9762	-9235.5276
DENSITY:					
KMOL/CUM	0.4965	41.4588	MISSING	41.4488	41.4502
KG/CUM	8.9449	746.9784	MISSING	746.7996	747.1700
AVG MW	18.0150	18.0173	MISSING	18.0173	18.0257

Simulated capacity increase by using CO₂.

STREAM ID	269	270	281	281A	281B
FROM :	V2114	V2117	----	H2108	H2109A
TO :	CONDMIX2	----	V2105	----	H2108
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	LIQUID	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	2.3003-08	2.0866-06	0.0	0.0	0.0
CO	2.5254-08	4.9521-07	0.0	0.0	0.0
CO2	9.1194-05	8.3660-08	11.2110	10.5143	10.5143
N2	2.8128-09	5.7450-08	0.0	0.0	0.0
O2	0.0	0.0	0.0	0.0	0.0
CH4	1.9193-07	4.7975-07	0.0	0.0	0.0
C2H6	0.0	1.3018-11	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	1.6526-05	101.9269	101.9222	101.9222
H2O	0.1352	1.0504	906.9775	906.6546	906.6546
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	0.1353	1.0504	1020.1155	1019.0911	1019.0911
KG/HR	2.4398	18.9252	2.3058+04	2.3022+04	2.3022+04
CUM/SEC	9.0675-07	7.0106-06	7.0614-03	9.4647-03	9.9099-03
STATE VARIABLES:					
TEMP C	40.0000	35.0000	40.0000	40.0000	88.0000
PRES BAR	16.2000	28.3000	14.7000	14.7000	1.7000
VFRAC	0.0	0.0	0.0	0.0	0.0
LFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.8775+08	-2.8810+08	-2.8449+08	-2.8373+08	-2.7906+08
J/KG	-1.5958+07	-1.5991+07	-1.2586+07	-1.2560+07	-1.2353+07
WATT	-1.0815+04	-8.4067+04	-8.0614+07	-8.0318+07	-7.8996+07
ENTROPY:					
J/KMOL-K	-1.6644+05	-1.6799+05	-2.4699+05	-2.2431+05	-2.1030+05
J/KG-K	-9230.2033	-9324.4639	-1.0927+04	-9929.6771	-9309.5128
DENSITY:					
KMOL/CUM	41.4486	41.6232	40.1289	29.9089	28.5655
KG/CUM	747.4236	749.8700	907.0602	675.6515	645.3032
AVG MW	18.0325	18.0156	22.6036	22.5902	22.5902

Simulated capacity increase by using CO₂.

STREAM ID	281C	282	282B	284	285
FROM :	V2106	V2105	H2109B	CONDMIX2	V2109
TO :	H2109A	H2109B	V2106	V2109	V2106
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	MIXED	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	0.0	1.4075-03	1.4075-03	1.3152-07	1.8998-09
CO	0.0	1.1254-03	1.1254-03	1.4622-07	9.2232-10
CO2	10.5143	48.2516	48.2516	5.4384-04	2.6856-02
N2	0.0	5.2169-05	5.2169-05	1.6266-08	3.9310-11
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	2.5065-05	2.5065-05	1.1258-06	2.2784-10
C2H6	0.0	1.5235-10	1.5235-10	0.0	0.0
C3H8	0.0	1.2490-15	1.2490-15	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	101.9222	101.9222	101.9222	0.0	1.6215-05
H2O	906.6546	906.9458	906.9458	1.2344	270.5353
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	1019.0911	1057.1222	1057.1222	1.2349	270.5622
KG/HR	2.3022+04	2.4688+04	2.4688+04	22.2619	4874.8776
CUM/SEC	7.7557-03	7.7228-03	4.9501-02	8.2785-06	1.8131-03
STATE VARIABLES:					
TEMP C	118.0284	68.5646	104.0000	40.1884	40.0000
PRES BAR	1.7000	10.4000	10.4000	1.5000	1.5000
VFRAC	0.0	0.0	4.6068-02	0.0	0.0
LFRAC	1.0000	1.0000	0.9539	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7784+08	-2.8830+08	-2.8124+08	-2.8774+08	-2.8772+08
J/KG	-1.2299+07	-1.2345+07	-1.2043+07	-1.5962+07	-1.5969+07
WATT	-7.8651+07	-8.4659+07	-8.2584+07	-9.8708+04	-2.1624+07
ENTROPY:					
J/KMOL-K	-2.0053+05	-2.2528+05	-1.9819+05	-1.6641+05	-1.6652+05
J/KG-K	-8876.6332	-9646.6502	-8486.3451	-9231.1693	-9241.9675
DENSITY:					
KMOL/CUM	36.4995	38.0232	5.9320	41.4379	41.4520
KG/CUM	824.5345	887.9808	138.5349	746.9783	746.8649
AVG MW	22.5902	23.3536	23.3536	18.0264	18.0175

Simulated capacity increase by using CO₂.

STREAM ID	3	4	5	6	7
FROM :	H2103	H2104	H2105A	H2105B	H2106
TO :	H2104	H2105A	H2105B	H2106	----
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	3.8182-05	3.8182-05	3.8182-05	3.8182-05	3.8182-05
CO	1.2933-06	1.2933-06	1.2933-06	1.2933-06	1.2933-06
CO2	2.3955	2.3955	2.3955	2.3955	2.3955
N2	268.2040	268.2040	268.2040	268.2040	268.2040
O2	9.5373	9.5373	9.5373	9.5373	9.5373
CH4	8.0301-31	8.0301-31	8.0301-31	8.0301-31	8.0301-31
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	125.3970	125.3970	125.3970	125.3970	125.3970
AR	3.2116	3.2116	3.2116	3.2116	3.2116
TOTAL FLOW:					
KMOL/HR	408.7456	408.7456	408.7456	408.7456	408.7456
KG/HR	1.0311+04	1.0311+04	1.0311+04	1.0311+04	1.0311+04
CUM/SEC	6.2891	5.5467	4.7415	3.7050	2.5517
STATE VARIABLES:					
TEMP C	725.7996	607.8656	480.0039	315.4778	132.7314
PRES BAR	1.5000	1.5000	1.5000	1.5000	1.5000
VFRAC :	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-5.3673+07	-5.7795+07	-6.2134+07	-6.7514+07	-7.3263+07
J/KG	-2.1276+06	-2.2911+06	-2.4631+06	-2.6763+06	-2.9042+06
WATT	-6.0940+06	-6.5621+06	-7.0547+06	-7.6656+06	-8.3183+06
ENTROPY:					
J/KMOL-K	2.8617+04	2.4227+04	1.8908+04	1.0855+04	-827.4121
J/KG-K	1134.4139	960.3928	749.5480	430.3124	-32.7995
DENSITY:					
KMOL/CUM	1.8053-02	2.0470-02	2.3946-02	3.0645-02	4.4496-02
KG/CUM	0.4554	0.5163	0.6040	0.7730	1.1224
AVG MW	25.2263	25.2263	25.2263	25.2263	25.2263

Simulated capacity increase by using CO₂.

STREAM ID	A	B	BIN	C	D
FROM :	SPLIT	H2102	----	SPLIT	H2105A
TO :	H2102	DRUMMIX	BURNER	H2105A	DRUMMIX
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	VAPOR	LIQUID	MIXED
COMPONENTS: KMOL/HR					
H2	0.0	0.0	125.3472	0.0	0.0
CO	0.0	0.0	2.3705	0.0	0.0
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.0	0.0	268.2040	0.0	0.0
O2	0.0	0.0	73.4460	0.0	0.0
CH4	0.0	0.0	2.4933-02	0.0	0.0
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	513.0561	513.0561	0.0	1061.1022	1061.1022
AR	0.0	0.0	3.2116	0.0	0.0
TOTAL FLOW:					
KMOL/HR	513.0561	513.0561	472.6045	1061.1022	1061.1022
KG/HR	9242.7065	9242.7065	1.0311+04	1.9116+04	1.9116+04
CUM/SEC	4.2628-03	1.4631-02	1.7101	8.8163-03	3.0260-02
STATE VARIABLES:					
TEMP C	221.0000	221.0000	40.0000	221.0000	221.0000
PRES BAR	24.0094	24.0094	2.0000	24.0094	24.0094
VFRAC	0.0	4.8000-02	1.0000	0.0	4.8000-02
LFRAC	1.0000	0.9520	0.0	1.0000	0.9520
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7130+08	-2.6962+08	-1.3077+05	-2.7130+08	-2.6962+08
J/KG	-1.5059+07	-1.4967+07	-5993.7285	-1.5059+07	-1.4967+07
WATT	-3.8664+07	-3.8426+07	-1.7167+04	-7.9965+07	-7.9472+07
ENTROPY:					
J/KMOL-K	-1.2552+05	-1.2213+05	4700.4527	-1.2552+05	-1.2213+05
J/KG-K	-6967.2642	-6779.5164	215.4409	-6967.2642	-6779.5164
DENSITY:					
KMOL/CUM	33.4325	9.7406	7.6766-02	33.4325	9.7406
KG/CUM	602.2870	175.4783	1.6748	602.2870	175.4783
AVG MW	18.0150	18.0150	21.8178	18.0150	18.0150

Simulated capacity increase by using CO₂.

STREAM ID	E	F	G	H	L10
FROM :	SPLIT	H2105B	SPLIT	H2107	V31
TO :	H2105B	DRUMMIX	H2107	DRUMMIX	MIX2
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	LIQUID	MIXED	VAPOR
COMPONENTS: KMOL/HR					
H2	0.0	0.0	0.0	0.0	1.4513
CO	0.0	0.0	0.0	0.0	1.6665
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.0	0.0	0.0	0.0	0.3796
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	0.0	0.0	0.0	2.5452-18
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	1315.6387	1315.6387	4920.4654	4920.4654	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	1315.6387	1315.6387	4920.4654	4920.4654	3.4975
KG/HR	2.3701+04	2.3701+04	8.8642+04	8.8642+04	60.2423
CUM/SEC	1.0931-02	3.7518-02	4.0882-02	0.1403	3.0713-03
STATE VARIABLES:					
TEMP C	221.0000	221.0000	221.0000	221.0000	-182.8828
PRES BAR	24.0094	24.0094	24.0094	24.0094	2.3000
VFRAC	0.0	4.8000-02	0.0	4.8000-02	1.0000
LFRAC	1.0000	0.9520	1.0000	0.9520	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7130+08	-2.6962+08	-2.7130+08	-2.6962+08	-5.8607+07
J/KG	-1.5059+07	-1.4967+07	-1.5059+07	-1.4967+07	-3.4026+06
WATT	-9.9147+07	-9.8536+07	-3.7081+08	-3.6852+08	-5.6940+04
ENTROPY:					
J/KMOL-K	-1.2552+05	-1.2213+05	-1.2552+05	-1.2213+05	1.0236+04
J/KG-K	-6967.2642	-6779.5164	-6967.2642	-6779.5158	594.2923
DENSITY:					
KMOL/CUM	33.4325	9.7406	33.4325	9.7406	0.3163
KG/CUM	602.2870	175.4783	602.2870	175.4783	5.4485
AVG MW	18.0150	18.0150	18.0150	18.0150	17.2240

Simulated capacity increase by using CO₂.

STREAM ID	L11	L13	L2	L22	L28
FROM :	T24	V47	E21	E23A	V6
TO :	T22MIX	MIX2	E22	E21	T23
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	VAPOR	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	4.0934-04	2.5451-15	169.5135	165.2264	2.8351
CO	21.1709	2.0282	66.6092	3.1799	43.8190
CO2	0.0	0.0	0.0	0.0	0.0
N2	3.2435	3.1783-02	8.4128	1.0910	4.0686
O2	0.0	0.0	0.0	0.0	0.0
CH4	5.3175-09	5.8788	5.9293	0.0	5.9293
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	24.4149	7.9388	250.4649	169.4974	56.6521
KG/HR	683.8605	152.0155	2538.2566	452.7304	1442.1882
CUM/SEC	2.8736-04	1.2453-04	2.1890-02	9.7081-02	3.8080-03
STATE VARIABLES:					
TEMP C	-169.2201	-170.9661	-157.6000	-176.1000	-177.0264
PRES BAR	7.6000	2.1500	28.1000	3.9000	5.4700
VFRAC	0.0	5.2010-03	1.0000	1.0000	0.1566
LFRAC	1.0000	0.9948	0.0	0.0	0.8433
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0687+08	-9.7950+07	-3.6570+07	-7.4679+06	-1.0445+08
J/KG	-3.8156+06	-5.1153+06	-3.6085+06	-2.7959+06	-4.1029+06
WATT	-7.2481+05	-2.1600+05	-2.5443+06	-3.5161+05	-1.6437+06
ENTROPY:					
J/KMOL-K	-1.6003+04	-1.4198+05	-2.6884+04	-3.7971+04	-2.8932+04
J/KG-K	-571.3501	-7414.9639	-2652.8225	-1.4216+04	-1136.5184
DENSITY:					
KMOL/CUM	23.6009	17.7077	3.1782	0.4849	4.1325
KG/CUM	661.0611	339.0736	32.2092	1.2954	105.2017
AVG MW	28.0099	19.1482	10.1341	2.6710	25.4568

Simulated capacity increase by using CO₂.

STREAM ID	L29	L3	L30	L31	L33
FROM :	T23	E22	V9	T24	V8
TO :	V29	T21	T24	V31	T22
SUBSTREAM: MIXED					
PHASE:	VAPOR	MIXED	MIXED	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	2.8332	169.5135	1.4516	1.4513	1.8036-03
CO	4.3134	66.6092	19.6102	1.6665	39.5056
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.5556	8.4128	3.2531	0.3796	3.5129
O2	0.0	0.0	0.0	0.0	0.0
CH4	5.0512-02	5.9293	0.0	2.5452-18	5.8788
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	7.7529	250.4649	24.3149	3.4975	48.8992
KG/HR	142.9094	2538.2566	643.3393	60.2423	1299.2787
CUM/SEC	2.8999-03	2.0072-02	1.0574-03	9.2342-04	4.8931-03
STATE VARIABLES:					
TEMP C	-178.6051	-164.8000	-176.6117	-177.5456	-181.9861
PRES BAR	5.4000	28.0000	7.8500	7.6000	2.3830
VFRAC	1.0000	0.9967	0.1325	1.0000	0.1106
LFRAC	0.0	3.2713-03	0.8674	0.0	0.8894
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.7947+07	-3.6808+07	-9.9871+07	-5.8608+07	-1.1007+08
J/KG	-3.6862+06	-3.6320+06	-3.7746+06	-3.4027+06	-4.1426+06
WATT	-1.4633+05	-2.5608+06	-6.7455+05	-5.6940+04	-1.4951+06
ENTROPY:					
J/KMOL-K	9766.8538	-2.8989+04	-1.8197+04	852.7922	-3.2996+04
J/KG-K	529.8614	-2860.4830	-687.7553	49.5117	-1241.8411
DENSITY:					
KMOL/CUM	0.7426	3.4662	6.3874	1.0521	2.7759
KG/CUM	13.6892	35.1278	169.0016	18.1217	73.7585
AVG MW	18.4328	10.1341	26.4585	17.2240	26.5705

Simulated capacity increase by using CO₂.

STREAM ID	L35	L36	L37	L38	L39
FROM :	E21	SPLIT1	E22B	SPLIT1	SPLIT1
TO :	SPLIT1	E22B	V37	E24	V39
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	4.0861-04	6.3744-05	6.3744-05	2.2310-04	1.2177-04
CO	10.8299	1.6894	1.6894	5.9131	3.2273
CO2	0.0	0.0	0.0	0.0	0.0
N2	1.2417	0.1937	0.1937	0.6780	0.3700
O2	0.0	0.0	0.0	0.0	0.0
CH4	1.7844-08	2.7837-09	2.7837-09	9.7428-09	5.3175-09
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	12.0721	1.8832	1.8832	6.5913	3.5974
KG/HR	338.1327	52.7487	52.7487	184.6204	100.7635
CUM/SEC	3.0805-03	4.8055-04	4.5005-04	1.6819-03	9.1798-04
STATE VARIABLES:					
TEMP C	-155.1000	-155.1000	-160.8000	-155.1000	-155.1000
PRES BAR	9.2000	9.2000	9.1000	9.2000	9.2000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0487+08	-1.0487+08	-1.0508+08	-1.0487+08	-1.0487+08
J/KG	-3.7442+06	-3.7442+06	-3.7514+06	-3.7442+06	-3.7442+06
WATT	-3.5168+05	-5.4862+04	-5.4967+04	-1.9202+05	-1.0480+05
ENTROPY:					
J/KMOL-K	3.5689+04	3.5689+04	3.4018+04	3.5689+04	3.5689+04
J/KG-K	1274.1718	1274.1718	1214.5273	1274.1718	1274.1718
DENSITY:					
KMOL/CUM	1.0885	1.0885	1.1623	1.0885	1.0885
KG/CUM	30.4907	30.4907	32.5573	30.4907	30.4907
AVG MW	28.0094	28.0094	28.0094	28.0094	28.0094

Simulated capacity increase by using CO₂.

STREAM ID	L4	L40	L41	L42	L43
FROM :	CO2	V39	E24	V41	V37
TO :	E23A	T24	V41	T22MIX	T22MIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	165.2264	1.2177-04	2.2310-04	2.2310-04	6.3744-05
CO	3.1799	3.2273	5.9131	5.9131	1.6894
CO2	0.0	0.0	0.0	0.0	0.0
N2	1.0910	0.3700	0.6780	0.6780	0.1937
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	5.3175-09	9.7428-09	9.7428-09	2.7837-09
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	169.4974	3.5974	6.5913	6.5913	1.8832
KG/HR	452.7304	100.7635	184.6204	184.6204	52.7487
CUM/SEC	9.4768-03	1.0652-03	1.6482-03	6.1241-03	1.6779-03
STATE VARIABLES:					
TEMP C	-201.5000	-156.8294	-157.6000	-167.5142	-171.2942
PRES BAR	27.3000	7.9500	9.1000	2.5000	2.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-8.2245+06	-1.0487+08	-1.0496+08	-1.0496+08	-1.0508+08
J/KG	-3.0792+06	-3.7442+06	-3.7473+06	-3.7473+06	-3.7514+06
WATT	-3.8723+05	-1.0480+05	-1.9217+05	-1.9217+05	-5.4967+04
ENTROPY:					
J/KMOL-K	-6.2901+04	3.6747+04	3.5043+04	4.4817+04	4.3692+04
J/KG-K	-2.3549+04	1311.9515	1251.1048	1600.0856	1559.9079
DENSITY:					
KMOL/CUM	4.9682	0.9381	1.1108	0.2989	0.3117
KG/CUM	13.2701	26.2757	31.1154	8.3739	8.7328
AVG MW	2.6710	28.0094	28.0094	28.0094	28.0094

Simulated capacity increase by using CO₂.

STREAM ID	L44	L46	L47	L48	L49
FROM :	T22MIX	E25-27	T22	MIX2	T22
TO :	E25-27	T22	V47	E21	E21
SUBSTREAM: MIXED					
PHASE:	MIXED	LIQUID	LIQUID	MIXED	VAPOR
COMPONENTS: KMOL/HR					
H2	6.9619-04	6.9619-04	2.5451-15	4.2846	2.4996-03
CO	28.7735	28.7735	2.0282	8.0083	66.2501
CO2	0.0	0.0	0.0	0.0	0.0
N2	4.1152	4.1152	3.1783-02	0.9671	7.5964
O2	0.0	0.0	0.0	0.0	0.0
CH4	1.7844-08	1.7844-08	5.8788	5.9293	1.0916-07
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	32.8895	32.8895	7.9388	19.1894	73.8490
KG/HR	921.2297	921.2297	152.0155	355.1672	2068.4694
CUM/SEC	1.0336-02	3.3895-04	8.2147-05	1.1266-02	6.2846-02
STATE VARIABLES:					
TEMP C	-183.1202	-184.5000	-170.4541	-179.5302	-184.0646
PRES BAR	2.5000	2.3250	2.2500	2.1500	2.2500
VFRAC	0.4003	0.0	0.0	0.5982	1.0000
LFRAC	0.5996	1.0000	1.0000	0.4017	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0639+08	-1.0874+08	-9.7950+07	-7.8657+07	-1.0546+08
J/KG	-3.7982+06	-3.8821+06	-5.1154+06	-4.2498+06	-3.7651+06
WATT	-9.7195+05	-9.9341+05	-2.1600+05	-4.1927+05	-2.1634+06
ENTROPY:					
J/KMOL-K	438.0986	-2.5661+04	-1.4199+05	-4.9479+04	4.0485+04
J/KG-K	15.6408	-916.1387	-7415.3313	-2673.3087	1445.4213
DENSITY:					
KMOL/CUM	0.8839	26.9535	26.8451	0.4731	0.3264
KG/CUM	24.7581	754.9654	514.0389	8.7567	9.1425
AVG MW	28.0098	28.0098	19.1482	18.5085	28.0094

Simulated capacity increase by using CO₂.

STREAM ID	L6	L7	L8	L9	LB
FROM :	T21	V29	T23	SPLIT2	T21
TO :	V6	MIX2	V8	V9	MIX1
SUBSTREAM: MIXED					
PHASE:	LIQUID	VAPOR	LIQUID	LIQUID	VAPOR
COMPONENTS: KMOL/HR					
H2	2.8351	2.8332	1.8036-03	1.4516	170.8884
CO	43.8190	4.3134	39.5056	19.6102	79.6631
CO2	0.0	0.0	0.0	0.0	0.0
N2	4.0686	0.5556	3.5129	3.2531	13.7788
O2	0.0	0.0	0.0	0.0	0.0
CH4	5.9293	5.0512-02	5.8788	0.0	6.0613-13
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	56.6521	7.7529	48.8992	24.3149	264.3303
KG/HR	1442.1882	142.9094	1299.2787	643.3393	2961.8613
CUM/SEC	6.4231-04	6.8485-03	5.4112-04	2.7941-04	1.9874-02
STATE VARIABLES:					
TEMP C	-169.0234	-182.0469	-172.0224	-171.5000	-170.7580
PRES BAR	27.3000	2.3000	5.4000	27.3470	27.3000
VFRAC	0.0	1.0000	0.0	0.0	1.0000
LFRAC	1.0000	0.0	1.0000	1.0000	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0445+08	-6.7946+07	-1.1007+08	-9.9871+07	-3.9135+07
J/KG	-4.1029+06	-3.6862+06	-4.1426+06	-3.7746+06	-3.4926+06
WATT	-1.6437+06	-1.4633+05	-1.4951+06	-6.7455+05	-2.8735+06
ENTROPY:					
J/KMOL-K	-3.0682+04	1.6499+04	-3.3479+04	-1.9585+04	-2.5598+04
J/KG-K	-1205.2466	895.1133	-1260.0177	-740.2158	-2284.4901
DENSITY:					
KMOL/CUM	24.5003	0.3144	25.1016	24.1733	3.6945
KG/CUM	623.7030	5.7964	666.9655	639.5910	41.3977
AVG MW	25.4568	18.4328	26.5705	26.4585	11.2051

Simulated capacity increase by using CO₂.

STREAM ID	LC	LD	LE	LF	LG
FROM :	CO1	CO1	CO2	MIX1	SPLIT2
TO :	CO2	SPLIT2	MIX1	CO1	T21
SUBSTREAM: MIXED					
PHASE:	VAPOR	LIQUID	LIQUID	MIXED	LIQUID
COMPONENTS: KMOL/HR					
H2	169.3174	5.6615	4.0906	174.9790	4.2099
CO	71.9319	76.4829	68.7517	148.4148	56.8727
CO2	0.0	0.0	0.0	0.0	0.0
N2	15.4632	12.6877	14.3722	28.1510	9.4346
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	0.0	0.0	6.0613-13	0.0
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	256.7126	94.8322	87.2145	351.5449	70.5172
KG/HR	2789.3300	2509.1236	2336.5923	5298.4536	1865.7843
CUM/SEC	1.9209-02	1.0897-03	8.0462-04	1.8751-02	8.1032-04
STATE VARIABLES:					
TEMP C	-171.5000	-171.5000	-201.5000	-174.3183	-171.5000
PRES BAR	27.3470	27.3470	27.3000	27.3000	27.3470
VFRAC	1.0000	0.0	0.0	0.6687	0.0
LFRAC	0.0	1.0000	1.0000	0.3312	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-3.6794+07	-9.9871+07	-9.9723+07	-5.4166+07	-9.9871+07
J/KG	-3.3863+06	-3.7746+06	-3.7222+06	-3.5938+06	-3.7746+06
WATT	-2.6238+06	-2.6308+06	-2.4159+06	-5.2894+06	-1.9563+06
ENTROPY:					
J/KMOL-K	-2.7578+04	-1.9585+04	-4.2191+04	-2.8965+04	-1.9585+04
J/KG-K	-2538.0718	-740.2158	-1574.8191	-1921.8016	-740.2158
DENSITY:					
KMOL/CUM	3.7123	24.1733	30.1089	5.2077	24.1733
KG/CUM	40.3366	639.5910	806.6591	78.4902	639.5910
AVG MW	10.8655	26.4585	26.7913	15.0719	26.4585

Simulated capacity increase by using CO₂.

STREAM ID	LIQ	MEASUMP	OUT	STM	STMFD
FROM :	V2103	V2116	V2118	V2103	DRUMMIX
TO :	SPLIT	----	----	----	V2103
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	MIXED	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	0.0	1.9061-07	1.0232	0.0	0.0
CO	0.0	4.5312-08	0.4020	0.0	0.0
CO2	0.0	6.6406-09	1.1361-02	0.0	0.0
N2	0.0	5.2626-09	5.0782-02	0.0	0.0
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	4.2721-08	3.5791-02	0.0	0.0
C2H6	0.0	0.0	1.1303-04	0.0	0.0
C3H8	0.0	0.0	5.2530-09	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	1.2851-06	4.7153-03	0.0	0.0
H2O	7810.2630	0.2128	0.4329	343.2634	7810.2625
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	7810.2630	0.2128	1.9609	343.2634	7810.2625
KG/HR	1.4070+05	3.8348	23.9115	6183.8913	1.4070+05
CUM/SEC	6.4892-02	1.4262-06	3.9312-04	0.1473	0.2227
STATE VARIABLES:					
TEMP C	221.0000	40.0000	35.0000	221.0000	221.0000
PRES BAR	24.0094	10.3000	28.3000	24.0094	24.0094
VFRAC	0.0	0.0	0.7800	1.0000	4.8000-02
LFRAC	1.0000	1.0000	0.2199	0.0	0.9520
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7130+08	-2.8770+08	-9.0162+07	-2.3648+08	-2.6962+08
J/KG	-1.5059+07	-1.5970+07	-7.3940+06	-1.3127+07	-1.4967+07
WATT	-5.8858+08	-1.7012+04	-4.9112+04	-2.2548+07	-5.8496+08
ENTROPY:					
J/KMOL-K	-1.2552+05	-1.6656+05	-3.6902+04	-5.5051+04	-1.2213+05
J/KG-K	-6967.2642	-9245.2469	-3026.2792	-3055.8504	-6779.5160
DENSITY:					
KMOL/CUM	33.4325	41.4597	1.3856	0.6470	9.7406
KG/CUM	602.2870	746.9080	16.8958	11.6559	175.4781
AVG MW	18.0150	18.0152	12.1938	18.0150	18.0150

Capacity increase of 20% by using a pre-reformer and extra CO₂.

146 147 150 2 201

STREAM ID	146	147	150	2	201
FROM :	B3	B5	B1	H2102	----
TO :	CO2-MIX	B3	H2O-MIX	H2103	H2-MIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	8.9420	5.3075	39.6825	3.7117-05	0.0
CO	0.1000	0.1120	0.7553	1.2567-06	1.1067-02
CO2	1.4746	0.0	0.0	2.3270	0.0
N2	7.3121	7.3121	0.2763	260.6468	7.2751
O2	3.1858-28	0.0	0.0	9.2645	0.0
CH4	43.6486	41.1394	0.0	7.8087-31	41.1394
C2H6	2.0503-03	1.4433	0.0	0.0	1.4433
C3H8	3.5404-07	0.1882	0.0	0.0	0.1882
C4H10	1.0303-10	7.0539-02	0.0	0.0	7.0539-02
C5H12	1.7900-14	1.9958-02	0.0	0.0	1.9958-02
C6H14	3.6504-18	2.3788-02	0.0	0.0	2.3788-02
MEA	1.5541-35	0.0	0.0	0.0	0.0
H2O	22.0418	24.9791	0.0	121.8739	0.0
AR	0.0	0.0	0.0	3.1215	0.0
TOTAL FLOW:					
KMOL/HR	83.5214	80.5962	40.7142	397.2340	50.1715
KG/HR	1387.9667	1387.9652	108.8988	1.0021+04	923.4000
CUM/SEC	9.5991-02	7.9312-02	7.7149-02	7.3419	1.5351-02
STATE VARIABLES:					
TEMP C	471.1761	500.0000	13.3824	926.8516	15.0000
PRES BAR	15.0000	18.2000	3.5000	1.5000	21.0000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-9.1835+07	-9.5168+07	-2.3797+06	-4.6430+07	-6.5305+07
J/KG	-5.5262+06	-5.5262+06	-8.8970+05	-1.8406+06	-3.5483+06
WATT	-2.1306+06	-2.1306+06	-2.6913+04	-5.1232+06	-9.1013+05
ENTROPY:					
J/KMOL-K	-2.9470+04	-3.3797+04	-8675.6728	3.5222+04	-9.5587+04
J/KG-K	-1773.3499	-1962.5487	-3243.5914	1396.2718	-5193.5551
DENSITY:					
KMOL/CUM	0.2416	0.2822	0.1465	1.5029-02	0.9078
KG/CUM	4.0164	4.8611	0.3920	0.3791	16.7091
AVG MW	16.6180	17.2212	2.6747	25.2260	18.4048

Capacity increase of 20% by using a pre-reformer and extra CO₂.

1 140 142 143 144

STREAM ID	1	140	142	143	144
FROM :	BURNER	B1	V2101	B2	----
TO :	H2102	----	B2	B5	B2
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	3.7117-05	120.4674	5.3075	5.3075	0.0
CO	1.2567-06	2.2930	0.1120	0.1120	0.0
CO2	2.3270	0.0	0.0	0.0	0.0
N2	260.6468	0.8390	7.3121	7.3121	0.0
O2	9.2645	0.0	0.0	0.0	0.0
CH4	7.8087-31	0.0	41.1394	41.1394	0.0
C2H6	0.0	0.0	1.4433	1.4433	0.0
C3H8	0.0	0.0	0.1882	0.1882	0.0
C4H10	0.0	0.0	7.0539-02	7.0539-02	0.0
C5H12	0.0	0.0	1.9958-02	1.9958-02	0.0
C6H14	0.0	0.0	2.3788-02	2.3788-02	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	121.8739	0.0	0.0	24.9791	24.9791
AR	3.1215	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	397.2340	123.5994	55.6170	80.5962	24.9791
KG/HR	1.0021+04	330.5929	937.9652	1387.9652	450.0000
CUM/SEC	7.8199	0.2342	5.4951-02	7.0931-02	1.1112-02
STATE VARIABLES:					
TEMP C	1005.0000	13.3824	500.0000	430.4931	220.0000
PRES BAR	1.5000	3.5000	18.2000	18.5000	23.2000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-4.3561+07	-2.3797+06	-3.6735+07	-9.8639+07	-2.3647+08
J/KG	-1.7268+06	-8.8970+05	-2.1782+06	-5.7278+06	-1.3126+07
WATT	-4.8066+06	-8.1702+04	-5.6753+05	-2.2083+06	-1.6408+06
ENTROPY:					
J/KMOL-K	3.7538+04	-8675.6728	-4.0735+04	-3.8637+04	-5.4783+04
J/KG-K	1488.0866	-3243.5914	-2415.3734	-2243.5567	-3040.9421
DENSITY:					
KMOL/CUM	1.4110-02	0.1465	0.2811	0.3156	0.6244
KG/CUM	0.3559	0.3920	4.7413	5.4355	11.2492
AVG MW	25.2260	2.6747	16.8647	17.2212	18.0150

Capacity increase of 20% by using a pre-reformer and extra CO₂.

202 202B 203 204 204B

STREAM ID	202	202B	203	204	204B
FROM :	B1	H2-MIX	H2104	V2114	CO2-MIX
TO :	H2-MIX	H2104	V2101	CO2-MIX	H2O-MIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	5.3075	5.3075	5.3075	4.0486	12.9907
CO	0.1010	0.1120	0.1120	6.0642	6.1643
CO2	0.0	0.0	0.0	25.9516	27.4263
N2	3.6965-02	7.3121	7.3121	1.0047	8.3169
O2	0.0	0.0	0.0	0.0	3.1858-28
CH4	0.0	41.1394	41.1394	8.1415	51.7901
C2H6	0.0	1.4433	1.4433	2.3864-10	2.0503-03
C3H8	0.0	0.1882	0.1882	0.0	3.5404-07
C4H10	0.0	7.0539-02	7.0539-02	0.0	1.0303-10
C5H12	0.0	1.9958-02	1.9958-02	0.0	1.7900-14
C6H14	0.0	2.3788-02	2.3788-02	0.0	3.6504-18
MEA	0.0	0.0	0.0	0.0	1.5541-35
H2O	0.0	0.0	0.0	0.1923	22.2342
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	5.4455	55.6170	55.6170	45.4033	128.9248
KG/HR	14.5652	937.9652	937.9652	1482.3828	2870.3495
CUM/SEC	1.0319-02	1.7106-02	5.1478-02	1.9573-02	0.1112
STATE VARIABLES:					
TEMP C	13.3824	14.1760	550.0000	40.0000	331.2364
PRES BAR	3.5000	21.0000	20.7000	16.2000	16.2000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.3797+06	-5.9144+07	-3.3863+07	-2.5414+08	-1.4899+08
J/KG	-8.8970+05	-3.5070+06	-2.0079+06	-7.7838+06	-6.6922+06
WATT	-3599.6273	-9.1373+05	-5.2316+05	-3.2052+06	-5.3358+06
ENTROPY:					
J/KMOL-K	-8675.6728	-8.5949+04	-3.8214+04	-1.3017+04	-1.8187+04
J/KG-K	-3243.5914	-5096.3944	-2265.9007	-398.6928	-816.8994
DENSITY:					
KMOL/CUM	0.1465	0.9031	0.3001	0.6443	0.3218
KG/CUM	0.3920	15.2312	5.0613	21.0382	7.1649
AVG MW	2.6747	16.8647	16.8647	32.6492	22.2637

Capacity increase of 20% by using a pre-reformer and extra CO₂.

	205	206	207	208	208B

STREAM ID	205	206	207	208	208B
FROM :	----	H2O-MIX	H2103	H2101	H2107
TO :	H2O-MIX	H2103	H2101	H2107	H2O-MI
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	0.0	52.6732	52.6732	170.5304	170.5304
CO	0.0	6.9196	6.9196	63.4034	63.4034
CO2	13.8604	41.2868	41.2868	28.3888	28.3888
N2	0.0	8.5933	8.5933	8.5933	8.5933
O2	0.0	3.1858-28	3.1858-28	0.0	0.0
CH4	0.0	51.7901	51.7901	8.2081	8.2081
C2H6	0.0	2.0503-03	2.0503-03	2.1401-04	2.1401-04
C3H8	0.0	3.5404-07	3.5404-07	1.3686-08	1.3686-08
C4H10	0.0	1.0303-10	1.0303-10	1.2049-12	1.2049-12
C5H12	0.0	1.7900-14	1.7900-14	7.1458-17	7.1458-17
C6H14	0.0	3.6504-18	3.6504-18	4.8507-21	4.8507-21
MEA	0.0	1.5541-35	1.5541-35	2.1063-30	2.1063-30
H2O	99.9167	122.1509	122.1509	91.4633	91.4633
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	113.7772	283.4162	283.4162	370.5876	370.5876
KG/HR	2410.0000	5389.2484	5389.2484	5389.2355	5389.2355
CUM/SEC	5.1371-02	0.2162	0.3296	0.8454	0.4220
STATE VARIABLES:					
TEMP C	220.0000	248.1458	487.0000	860.0000	293.1850
PRES BAR	23.2000	15.6000	15.1000	11.5000	11.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.5467+08	-1.7036+08	-1.6066+08	-8.1999+07	-1.0198+08
J/KG	-1.2023+07	-8.9589+06	-8.4489+06	-5.6386+06	-7.0128+06
WATT	-8.0489+06	-1.3412+07	-1.2648+07	-8.4411+06	-1.0498+07
ENTROPY:					
J/KMOL-K	-4.5384+04	-2.2420+04	-6881.3373	3.8598+04	1.4313+04
J/KG-K	-2142.6200	-1179.0385	-361.8840	2654.2028	984.2019
DENSITY:					
KMOL/CUM	0.6152	0.3640	0.2388	0.1217	0.2438
KG/CUM	13.0316	6.9218	4.5417	1.7707	3.5466
AVG MW	21.1817	19.0153	19.0153	14.5424	14.5424

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Capacity increase of 20% by using a pre-reformer and extra CO₂.

208C 209 209B 210 210B

STREAM ID	208C	209	209B	210	210B
FROM :	H20-MIX2	M2102	H2110A	V2107	H2111
TO :	M2102	H2110A	V2107	H2111	V2108
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	MIXED	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	170.5304	170.5304	170.5304	170.5302	170.5302
CO	63.4034	63.4034	63.4034	63.4033	63.4033
CO2	28.3888	28.3888	28.3888	28.3724	28.3724
N2	8.5933	8.5933	8.5933	8.5933	8.5933
O2	0.0	0.0	0.0	0.0	0.0
CH4	8.2081	8.2081	8.2081	8.2081	8.2081
C2H6	2.1401-04	2.1401-04	2.1401-04	2.1401-04	2.1401-04
C3H8	1.3686-08	1.3686-08	1.3686-08	1.3686-08	1.3686-08
C4H10	1.2049-12	1.2049-12	1.2049-12	0.0	0.0
C5H12	7.1458-17	7.1458-17	7.1458-17	0.0	0.0
C6H14	4.8507-21	4.8507-21	4.8507-21	0.0	0.0
MEA	2.1063-30	2.1063-30	2.1063-30	0.0	0.0
H2O	196.9309	227.4611	227.4611	1.6028	1.6028
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	476.0553	506.5854	506.5854	280.7105	280.7105
KG/HR	7289.2355	7839.2355	7839.2355	3769.6748	3769.6748
CUM/SEC	0.5400	0.4799	0.3308	0.1903	0.1958
STATE VARIABLES:					
TEMP C	269.0083	182.8193	132.0000	40.0000	40.0000
PRES BAR	11.0000	11.0000	10.7000	10.7000	10.4000
VFRAC	1.0000	1.0000	0.7482	1.0000	1.0000
LFRAC	0.0	0.0	0.2517	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.3182+08	-1.4088+08	-1.5283+08	-6.7935+07	-6.7934+07
J/KG	-8.6092+06	-9.1038+06	-9.8765+06	-5.0588+06	-5.0588+06
WATT	-1.7432+07	-1.9824+07	-2.1507+07	-5.2972+06	-5.2972+06
ENTROPY:					
J/KMOL-K	2584.2472	-5638.3418	-3.4038+04	8938.4811	9177.9829
J/KG-K	168.7755	-364.3597	-2199.5920	665.6079	683.4425
DENSITY:					
KMOL/CUM	0.2448	0.2931	0.4252	0.4095	0.3981
KG/CUM	3.7492	4.5368	6.5812	5.5002	5.3465
AVG MW	15.3117	15.4746	15.4746	13.4290	13.4290

Capacity increase of 20% by using a pre-reformer and extra CO₂.

	211	212	213	213B	214

STREAM ID	211	212	213	213B	214
FROM :	V2108	V2105	V2106	H2112	VENTTAR
TO :	V2105	V2116	H2112	V2109	----
SUBSTREAM: MIXED					
PHASE:	VAPOR	MIXED	VAPOR	MIXED	MIXED
COMPONENTS: KMOL/HR					
H2	170.5302	170.5290	1.2014-03	1.2014-03	0.0
CO	63.4033	63.4024	9.0682-04	9.0682-04	0.0
CO2	28.3724	1.1361-02	27.2665	27.2665	1.2976
N2	8.5933	8.5932	4.5131-05	4.5131-05	2.1817-06
O2	0.0	0.0	0.0	0.0	0.0
CH4	8.2081	8.2080	2.8970-05	2.8970-05	0.0
C2H6	2.1401-04	2.1401-04	2.3864-10	2.3864-10	0.0
C3H8	1.3686-08	1.3686-08	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	4.7104-03	4.2717-05	4.2717-05	3.2763-05
H2O	1.6028	1.6879	173.8517	173.8517	5.3079-02
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	280.7105	252.4371	201.1205	201.1205	1.3507
KG/HR	3769.6748	2523.2971	4331.9718	4331.9718	58.0657
CUM/SEC	0.1958	0.1780	1.0377	0.1370	6.4686-03
STATE VARIABLES:					
TEMP C	40.0000	40.0419	110.4955	40.0000	40.0000
PRES BAR	10.4000	10.3000	1.7000	1.5000	1.5000
VFRAC	1.0000	0.9999	1.0000	0.1410	1.0000
LFRAC	0.0	2.3903-05	0.0	0.8589	2.5078-06
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.7934+07	-3.1429+07	-2.5970+08	-3.0177+08	-3.8730+08
J/KG	-5.0588+06	-3.1443+06	-1.2057+07	-1.4010+07	-9.0092+06
WATT	-5.2972+06	-2.2039+06	-1.4508+07	-1.6859+07	-1.4531+05
ENTROPY:					
J/KMOL-K	9177.9829	8948.3616	-3.0380+04	-1.4290+05	894.6224
J/KG-K	683.4425	895.2169	-1410.4489	-6634.5413	20.8105
DENSITY:					
KMOL/CUM	0.3981	0.3939	5.3835-02	0.4077	5.8003-02
KG/CUM	5.3465	3.9378	1.1595	8.7830	2.4934
AVG MW	13.4290	9.9957	21.5391	21.5391	42.9888

Capacity increase of 20% by using a pre-reformer and extra CO₂.

215 215B 216 216B 216C

STREAM ID	215	215B	216	216B	216C
FROM :	V2109	VENTTANK	CO-MIX	K2103-1	H2113
TO :	VENTTANK	CO-MIX	K2103-1	H2113	V2113
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	1.2015-03	1.2015-03	4.0487	4.0487	4.0486
CO	9.0692-04	9.0692-04	6.0642	6.0642	6.0642
CO2	27.2496	25.9520	25.9520	25.9520	25.9520
N2	4.5145-05	4.2964-05	1.0047	1.0047	1.0047
O2	0.0	0.0	0.0	0.0	0.0
CH4	3.0310-05	3.0310-05	8.1415	8.1415	8.1415
C2H6	2.3863-10	2.3863-10	2.3863-10	2.3863-10	2.3864-10
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	3.2763-05	0.0	0.0	0.0	0.0
H2O	1.1146	1.0615	1.0615	1.0615	1.0615
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	28.3664	27.0157	46.2728	46.2728	46.2728
KG/HR	1219.3680	1161.3023	1498.0567	1498.0567	1498.0566
CUM/SEC	0.1358	0.1293	0.2182	4.6313-02	3.4322-02
STATE VARIABLES:					
TEMP C	40.0000	40.0000	34.3187	145.0000	40.0000
PRES BAR	1.5000	1.5000	1.5000	9.6000	9.4000
VFRAC	1.0000	1.0000	1.0000	1.0000	0.9837
LFRAC	0.0	0.0	0.0	0.0	1.6265-02
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-3.8728+08	-3.8727+08	-2.5378+08	-2.4977+08	-2.5450+08
J/KG	-9.0093+06	-9.0093+06	-7.8389+06	-7.7152+06	-7.8612+06
WATT	-3.0516+06	-2.9063+06	-3.2620+06	-3.2105+06	-3.2713+06
ENTROPY:					
J/KMOL-K	918.0661	919.1569	6580.6246	2326.8681	-1.0727+04
J/KG-K	21.3572	21.3826	203.2662	71.8737	-331.3505
DENSITY:					
KMOL/CUM	5.8003-02	5.8003-02	5.8891-02	0.2775	0.3745
KG/CUM	2.4933	2.4933	1.9065	8.9851	12.1242
AVG MW	42.9862	42.9860	32.3744	32.3744	32.3744

Capacity increase of 20% by using a pre-reformer and extra CO₂.

216D 216E 216F 218 218B

STREAM ID	216D	216E	216F	218	218B
FROM :	V2113	K2103-3	H2114	V2116	K2104
TO :	K2103-3	H2114	V2114	K2104	H2115
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	MIXED	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	4.0486	4.0486	4.0486	170.5290	170.5290
CO	6.0642	6.0642	6.0642	63.4024	63.4024
CO2	25.9517	25.9517	25.9517	1.1361-02	1.1361-02
N2	1.0047	1.0047	1.0047	8.5932	8.5932
O2	0.0	0.0	0.0	0.0	0.0
CH4	8.1415	8.1415	8.1415	8.2080	8.2080
C2H6	2.3864-10	2.3864-10	2.3864-10	2.1401-04	2.1401-04
C3H8	0.0	0.0	0.0	1.3686-08	1.3686-08
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	4.7092-03	4.7092-03
H2O	0.3033	0.3033	0.3033	1.4763	1.4763
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	45.5143	45.5143	45.5143	252.2255	252.2255
KG/HR	1484.3853	1484.3853	1484.3853	2519.4850	2519.4850
CUM/SEC	3.3583-02	2.8339-02	1.9573-02	0.1780	8.6718-02
STATE VARIABLES:					
TEMP C	40.0000	145.0000	40.0000	40.0000	145.0000
PRES BAR	9.6000	15.4000	16.2000	10.3000	28.5000
VFRAC	1.0000	1.0000	0.9975	1.0000	1.0000
LFRAC	0.0	0.0	2.4401-03	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.5396+08	-2.5003+08	-2.5422+08	-3.1255+07	-2.8156+07
J/KG	-7.7869+06	-7.6666+06	-7.7948+06	-3.1289+06	-2.8187+06
WATT	-3.2108+06	-3.1612+06	-3.2140+06	-2.1898+06	-1.9727+06
ENTROPY:					
J/KMOL-K	-8336.0993	-1432.3927	-1.3391+04	8958.9903	8944.9214
J/KG-K	-255.6023	-43.9201	-410.6092	896.8840	895.4755
DENSITY:					
KMOL/CUM	0.3764	0.4461	0.6459	0.3935	0.8079
KG/CUM	12.2778	14.5500	21.0659	3.9309	8.0704
AVG MW	32.6135	32.6135	32.6135	9.9890	9.9890

Capacity increase of 20% by using a pre-reformer and extra CO₂.

218C 219 220 224 224A

STREAM ID	218C	219	220	224	224A
FROM :	H2115	V2117	V2118	E21	K2105-1
TO :	V2117	V2118	E21	K2105-1	H2118
SUBSTREAM: MIXED					
PHASE:	MIXED	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	170.5290	170.5290	169.5058	1.0714-03	1.0714-03
CO	63.4024	63.4024	63.0220	64.3243	64.3243
CO2	1.1361-02	1.1361-02	0.0	0.0	0.0
N2	8.5932	8.5932	8.5417	7.6322	7.6322
O2	0.0	0.0	0.0	0.0	0.0
CH4	8.2080	8.2080	8.1588	2.0697-02	2.0697-02
C2H6	2.1401-04	2.1401-04	0.0	0.0	0.0
C3H8	1.3686-08	1.3686-08	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	4.7092-03	4.6927-03	0.0	0.0	0.0
H2O	1.4763	0.4310	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	252.2255	251.1801	249.2284	71.9784	71.9784
KG/HR	2519.4850	2500.6520	2477.1415	2015.8630	2015.8630
CUM/SEC	6.3905-02	6.4121-02	6.3633-02	0.2389	0.1036
STATE VARIABLES:					
TEMP C	35.0000	35.0000	35.0000	21.4699	100.0000
PRES BAR	28.4000	28.3000	28.3000	2.0500	6.0000
VFRAC	0.9958	1.0000	1.0000	1.0000	1.0000
LFRAC	4.1499-03	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-3.1602+07	-3.0534+07	-3.0150+07	-9.8992+07	-9.6701+07
J/KG	-3.1637+06	-3.0671+06	-3.0335+06	-3.5346+06	-3.4528+06
WATT	-2.2141+06	-2.1305+06	-2.0873+06	-1.9792+06	-1.9334+06
ENTROPY:					
J/KMOL-K	-650.1411	77.0194	105.3257	7.6746+04	7.4700+04
J/KG-K	-65.0856	7.7362	10.5969	2740.2918	2667.2242
DENSITY:					
KMOL/CUM	1.0963	1.0881	1.0879	8.3690-02	0.1929
KG/CUM	10.9515	10.8330	10.8134	2.3438	5.4023
AVG MW	9.9890	9.9556	9.9392	28.0064	28.0064

Capacity increase of 20% by using a pre-reformer and extra CO₂.

224B 224C 224D 225 226

STREAM ID	224B	224C	224D	225	226
FROM :	H2118	K2105-2	H2119	CO-SPLIT	CO-SPLIT
TO :	K2105-2	H2119	CO-SPLIT	E21	----
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	1.0714-03	1.0714-03	1.0714-03	1.7515-04	8.9630-04
CO	64.3243	64.3243	64.3243	10.5151	53.8092
CO2	0.0	0.0	0.0	0.0	0.0
N2	7.6322	7.6322	7.6322	1.2476	6.3846
O2	0.0	0.0	0.0	0.0	0.0
CH4	2.0697-02	2.0697-02	2.0697-02	3.3834-03	1.7314-02
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	71.9784	71.9784	71.9784	11.7663	60.2121
KG/HR	2015.8630	2015.8630	2015.8630	329.5331	1686.3299
CUM/SEC	8.6926-02	6.3128-02	5.4560-02	8.9189-03	4.5641-02
STATE VARIABLES:					
TEMP C	30.0000	90.0000	35.0000	35.0000	35.0000
PRES BAR	5.8000	9.6000	9.4000	9.4000	9.4000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-9.8766+07	-9.7008+07	-9.8640+07	-9.8640+07	-9.8640+07
J/KG	-3.5265+06	-3.4638+06	-3.5221+06	-3.5221+06	-3.5221+06
WATT	-1.9747+06	-1.9396+06	-1.9722+06	-3.2240+05	-1.6498+06
ENTROPY:					
J/KMOL-K	6.8853+04	6.9944+04	6.5247+04	6.5247+04	6.5247+04
J/KG-K	2458.4652	2497.4355	2329.6956	2329.6956	2329.6956
DENSITY:					
KMOL/CUM	0.2300	0.3167	0.3664	0.3664	0.3664
KG/CUM	6.4418	8.8702	10.2632	10.2632	10.2632
AVG MW	28.0064	28.0064	28.0064	28.0064	28.0064

Capacity increase of 20% by using a pre-reformer and extra CO₂.

227 228 251 251B 253

STREAM ID	227	228	251	251B	253
FROM :	E21	E21	----	H2106	----
TO :	CO-MIX	B1	H2106	V2103	M2102
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	LIQUID	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	4.0474	165.4575	0.0	0.0	0.0
CO	6.0633	3.1493	0.0	0.0	0.0
CO2	0.0	0.0	0.0	0.0	0.0
N2	1.0047	1.1523	0.0	0.0	0.0
O2	0.0	0.0	0.0	0.0	0.0
CH4	8.1415	0.0	0.0	0.0	0.0
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	299.7502	299.7502	30.5301
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	19.2571	169.7592	299.7502	299.7502	30.5301
KG/HR	336.7543	454.0570	5400.0000	5400.0000	550.0000
CUM/SEC	6.8032-02	0.3216	2.1204-03	2.3230-03	2.1596-04
STATE VARIABLES:					
TEMP C	25.2967	13.3824	104.0000	180.0000	104.0000
PRES BAR	1.9500	3.5000	27.0000	26.8000	27.0000
VFRAC	1.0000	1.0000	0.0	0.0	0.0
LFRAC	0.0	0.0	1.0000	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.6502+07	-2.3797+06	-2.8210+08	-2.7526+08	-2.8210+08
J/KG	-3.8029+06	-8.8970+05	-1.5659+07	-1.5279+07	-1.5659+07
WATT	-3.5573+05	-1.1221+05	-2.3489+07	-2.2919+07	-2.3924+06
ENTROPY:					
J/KMOL-K	-1224.6329	-8675.6728	-1.5042+05	-1.3390+05	-1.5042+05
J/KG-K	-70.0299	-3243.5914	-8349.6409	-7432.5015	-8349.6409
DENSITY:					
KMOL/CUM	7.8628-02	0.1465	39.2685	35.8435	39.2685
KG/CUM	1.3749	0.3920	707.4237	645.7212	707.4237
AVG MW	17.4872	2.6747	18.0150	18.0150	18.0150

Capacity increase of 20% by using a pre-reformer and extra CO₂.

	260	264	265	266	268

STREAM ID	260	264	265	266	268
FROM :	----	V2107	V2108	COND-MIX	V2113
TO :	H2O-MIX2	COND-MIX	COND-MIX	----	CONDMIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	LIQUID	MIXED	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	0.0	1.9010-04	0.0	1.9010-04	8.5973-08
CO	0.0	4.2653-05	0.0	4.2653-05	7.6869-08
CO2	0.0	1.6399-02	0.0	1.6399-02	2.7490-04
N2	0.0	5.3172-06	0.0	5.3172-06	1.1728-08
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	5.8329-05	0.0	5.8329-05	1.0771-06
C2H6	0.0	2.2731-09	0.0	2.2731-09	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	105.4676	225.8582	0.0	225.8582	0.7582
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	105.4676	225.8749	0.0	225.8749	0.7584
KG/HR	1900.0000	4069.5606	0.0	4069.5606	13.6712
CUM/SEC	5.9003-02	1.5134-03	0.0	1.5137-03	5.0828-06
STATE VARIABLES:					
TEMP C	208.0000	40.0000	MISSING	40.1538	40.0000
PRES BAR	18.3000	10.7000	10.4000	3.5000	9.6000
VFRAC	1.0000	0.0	MISSING	0.0	0.0
LFRAC	0.0	1.0000	MISSING	1.0000	1.0000
SFRAC	0.0	0.0	MISSING	0.0	0.0
ENTHALPY:					
J/KMOL	-2.3667+08	-2.8770+08	MISSING	-2.8770+08	-2.8774+08
J/KG	-1.3137+07	-1.5969+07	MISSING	-1.5969+07	-1.5964+07
WATT	-6.9336+06	-1.8051+07	MISSING	-1.8051+07	-6.0623+04
ENTROPY:					
J/KMOL-K	-5.3384+04	-1.6654+05	MISSING	-1.6648+05	-1.6649+05
J/KG-K	-2963.3323	-9243.5005	MISSING	-9240.4213	-9236.6507
DENSITY:					
KMOL/CUM	0.4965	41.4592	MISSING	41.4493	41.4514
KG/CUM	8.9449	746.9662	MISSING	746.7875	747.1381
AVG MW	18.0150	18.0168	MISSING	18.0168	18.0244

Capacity increase of 20% by using a pre-reformer and extra CO₂.

269 270 281 281A 281B

STREAM ID	269	270	281	281A	281B
FROM :	V2114	V2117	----	H2108	H2109A
TO :	CONDMIX2	----	V2105	----	H2108
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	LIQUID	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	2.1641-08	2.0867-06	0.0	0.0	0.0
CO	1.9067-08	4.6854-07	0.0	0.0	0.0
CO2	6.5917-05	8.3636-08	9.5819	10.6936	10.6936
N2	2.9135-09	5.8329-08	0.0	0.0	0.0
O2	0.0	0.0	0.0	0.0	0.0
CH4	2.6321-07	6.6006-07	0.0	0.0	0.0
C2H6	0.0	2.4640-11	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	1.6422-05	87.1169	87.1121	87.1121
H2O	0.1110	1.0453	775.1928	774.8650	774.8650
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	0.1110	1.0453	871.8917	872.6708	872.6708
KG/HR	2.0024	18.8329	1.9708+04	1.9751+04	1.9751+04
CUM/SEC	7.4427-07	6.9764-06	6.0353-03	8.1078-03	8.4903-03
STATE VARIABLES:					
TEMP C	40.0000	35.0000	40.0000	40.0000	88.0000
PRES BAR	16.2000	28.3000	14.7000	14.7000	1.7000
VFRAC	0.0	0.0	0.0	0.0	0.0
LFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.8775+08	-2.8810+08	-2.8449+08	-2.8395+08	-2.7928+08
J/KG	-1.5959+07	-1.5991+07	-1.2586+07	-1.2546+07	-1.2340+07
WATT	-8877.0814	-8.3657+04	-6.8900+07	-6.8832+07	-6.7700+07
ENTROPY:					
J/KMOL-K	-1.6646+05	-1.6799+05	-2.4699+05	-2.2394+05	-2.0993+05
J/KG-K	-9232.0216	-9324.4627	-1.0927+04	-9894.5010	-9275.8331
DENSITY:					
KMOL/CUM	41.4505	41.6232	40.1289	29.8981	28.5513
KG/CUM	747.3714	749.8700	907.0601	676.6661	646.1847
AVG MW	18.0304	18.0156	22.6036	22.6324	22.6324

Capacity increase of 20% by using a pre-reformer and extra CO₂.

	281C	282	282B	284	285
281C 282 282B 284 285					

STREAM ID	281C	282	282B	284	285
FROM :	V2106	V2105	H2109B	CONDMIX2	V2109
TO :	H2109A	H2109B	V2106	V2109	V2106
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	MIXED	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
H2	0.0	1.2014-03	1.2014-03	1.0761-07	1.4412-09
CO	0.0	9.0682-04	9.0682-04	9.5936-08	6.6044-10
CO2	10.6936	37.9429	37.9429	3.4082-04	1.7234-02
N2	0.0	4.5131-05	4.5131-05	1.4642-08	3.0222-11
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	2.8970-05	2.8970-05	1.3403-06	2.3433-10
C2H6	0.0	2.3864-10	2.3864-10	0.0	0.0
C3H8	0.0	2.6699-15	2.6699-15	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	87.1121	87.1121	87.1121	0.0	9.9538-06
H2O	774.8650	775.1077	775.1077	0.8692	173.6063
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	872.6708	900.1651	900.1651	0.8695	173.6236
KG/HR	1.9751+04	2.0954+04	2.0954+04	15.6737	3128.2775
CUM/SEC	6.6500-03	6.5685-03	3.9219-02	5.8288-06	1.1635-03
STATE VARIABLES:					
TEMP C	117.9567	69.3631	104.0000	40.1910	40.0000
PRES BAR	1.7000	10.4000	10.4000	1.5000	1.5000
VFRAC	0.0	0.0	4.2005-02	0.0	0.0
LFRAC	1.0000	1.0000	0.9579	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7826+08	-2.8792+08	-2.8086+08	-2.8774+08	-2.8772+08
J/KG	-1.2295+07	-1.2369+07	-1.2065+07	-1.5963+07	-1.5969+07
WATT	-6.7452+07	-7.1994+07	-7.0228+07	-6.9500+04	-1.3877+07
ENTROPY:					
J/KMOL-K	-2.0057+05	-2.2593+05	-1.9895+05	-1.6641+05	-1.6652+05
J/KG-K	-8861.8678	-9705.5860	-8546.7070	-9232.2332	-9241.9675
DENSITY:					
KMOL/CUM	36.4526	38.0676	6.3756	41.4389	41.4520
KG/CUM	825.0106	886.1483	148.4134	746.9455	746.8649
AVG MW	22.6324	23.2782	23.2782	18.0251	18.0175

Capacity increase of 20% by using a pre-reformer and extra CO₂.3 4 5 6 7

STREAM ID	3	4	5	6	7
FROM :	H2103	H2104	H2105A	H2105B	H2106
TO :	H2104	H2105A	H2105B	H2106	----
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	3.7117-05	3.7117-05	3.7117-05	3.7117-05	3.7117-05
CO	1.2567-06	1.2567-06	1.2567-06	1.2567-06	1.2567-06
CO2	2.3270	2.3270	2.3270	2.3270	2.3270
N2	260.6468	260.6468	260.6468	260.6468	260.6468
O2	9.2645	9.2645	9.2645	9.2645	9.2645
CH4	7.8087-31	7.8087-31	7.8087-31	7.8087-31	7.8087-31
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	121.8739	121.8739	121.8739	121.8739	121.8739
AR	3.1215	3.1215	3.1215	3.1215	3.1215
TOTAL FLOW:					
KMOL/HR	397.2340	397.2340	397.2340	397.2340	397.2340
KG/HR	1.0021+04	1.0021+04	1.0021+04	1.0021+04	1.0021+04
CUM/SEC	6.1686	5.5515	5.0404	4.2528	3.2719
STATE VARIABLES:					
TEMP C	735.0473	634.1897	550.6559	421.9694	261.8248
PRES BAR	1.5000	1.5000	1.5000	1.5000	1.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-5.3349+07	-5.6889+07	-5.9758+07	-6.4061+07	-6.9227+07
J/KG	-2.1148+06	-2.2552+06	-2.3689+06	-2.5395+06	-2.7443+06
WATT	-5.8867+06	-6.2772+06	-6.5938+06	-7.0687+06	-7.6387+06
ENTROPY:					
J/KMOL-K	2.8943+04	2.5244+04	2.1928+04	1.6250+04	7809.6778
J/KG-K	1147.3298	1000.7259	869.2447	644.1855	309.5883
DENSITY:					
KMOL/CUM	1.7888-02	1.9876-02	2.1892-02	2.5946-02	3.3724-02
KG/CUM	0.4512	0.5013	0.5522	0.6545	0.8507
AVG MW	25.2260	25.2260	25.2260	25.2260	25.2260

Capacity increase of 20% by using a pre-reformer and extra CO₂.

A	B	BIN	C	D	

STREAM ID	A	B	BIN	C	D
FROM :	SPLIT	H2102	----	SPLIT	H2105A
TO :	H2102	DRUMMIX	BURNER	H2105A	DRUMMIX
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	VAPOR	LIQUID	MIXED
COMPONENTS: KMOL/HR					
H2	0.0	0.0	121.8254	0.0	0.0
CO	0.0	0.0	2.3027	0.0	0.0
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.0	0.0	260.6468	0.0	0.0
O2	0.0	0.0	71.3772	0.0	0.0
CH4	0.0	0.0	2.4310-02	0.0	0.0
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	681.8751	681.8751	0.0	681.8751	681.8649
AR	0.0	0.0	3.1215	0.0	0.0
TOTAL FLOW:					
KMOL/HR	681.8751	681.8751	459.2980	681.8751	681.8649
KG/HR	1.2284+04	1.2284+04	1.0021+04	1.2284+04	1.2284+04
CUM/SEC	5.6654-03	1.9445-02	1.6619	5.6654-03	1.9445-02
STATE VARIABLES:					
TEMP C	221.0000	221.0000	40.0000	221.0000	221.0000
PRES BAR	24.0094	24.0094	2.0000	24.0094	24.0094
VFRAC	0.0	4.8000-02	1.0000	0.0	4.8000-02
LFRAC	1.0000	0.9520	0.0	1.0000	0.9520
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7130+08	-2.6962+08	-1.3052+05	-2.7130+08	-2.6962+08
J/KG	-1.5059+07	-1.4967+07	-5982.4219	-1.5059+07	-1.4967+07
WATT	-5.1386+07	-5.1070+07	-1.6652+04	-5.1386+07	-5.1069+07
ENTROPY:					
J/KMOL-K	-1.2552+05	-1.2213+05	4700.2516	-1.2552+05	-1.2213+05
J/KG-K	-6967.2642	-6779.5164	215.4359	-6967.2642	-6779.5164
DENSITY:					
KMOL/CUM	33.4325	9.7406	7.6766-02	33.4325	9.7406
KG/CUM	602.2870	175.4783	1.6748	602.2870	175.4783
AVG MW	18.0150	18.0150	21.8174	18.0150	18.0150

Capacity increase of 20% by using a pre-reformer and extra CO₂.

E F G H L10

STREAM ID	E	F	G	H	L10
FROM :	SPLIT	H2105B	SPLIT	H2107	V31
TO :	H2105B	DRUMMIX	H2107	DRUMMIX	MIX2
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	LIQUID	MIXED	VAPOR
COMPONENTS: KMOL/HR					
H2	0.0	0.0	0.0	0.0	1.3802
CO	0.0	0.0	0.0	0.0	1.5707
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.0	0.0	0.0	0.0	0.3822
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	0.0	0.0	0.0	5.5478-13
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	1022.8127	1022.7973	4432.1885	4432.1885	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	1022.8127	1022.7973	4432.1885	4432.1885	3.3332
KG/HR	1.8426+04	1.8426+04	7.9846+04	7.9846+04	57.4859
CUM/SEC	8.4982-03	2.9167-02	3.6825-02	0.1263	2.9257-03
STATE VARIABLES:					
TEMP C	221.0000	221.0000	221.0000	220.9948	-182.9165
PRES BAR	24.0094	24.0094	24.0094	24.0094	2.3000
VFRAC	0.0	4.8000-02	0.0	4.8000-02	1.0000
LFRAC	1.0000	0.9520	1.0000	0.9520	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7130+08	-2.6962+08	-2.7130+08	-2.6963+08	-5.8026+07
J/KG	-1.5059+07	-1.4967+07	-1.5059+07	-1.4967+07	-3.3646+06
WATT	-7.7079+07	-7.6603+07	-3.3401+08	-3.3195+08	-5.3726+04
ENTROPY:					
J/KMOL-K	-1.2552+05	-1.2213+05	-1.2552+05	-1.2213+05	9822.3771
J/KG-K	-6967.2642	-6779.5164	-6967.2642	-6779.5729	569.5337
DENSITY:					
KMOL/CUM	33.4325	9.7406	33.4325	9.7408	0.3164
KG/CUM	602.2870	175.4783	602.2870	175.4806	5.4580
AVG MW	18.0150	18.0150	18.0150	18.0150	17.2463

Capacity increase of 20% by using a pre-reformer and extra CO₂.L11 L13 L2 L22 L28

STREAM ID	L11	L13	L2	L22	L28
FROM :	T24	V47	E21	E23A	V6
TO :	T22MIX	MIX2	E22	E21	T23
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	VAPOR	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	3.3576-04	1.0488-16	169.5058	165.4575	2.6678
CO	20.0260	2.0751-03	63.0220	3.1493	41.4093
CO2	0.0	0.0	0.0	0.0	0.0
N2	3.2569	5.2416-05	8.5417	1.1523	4.1219
O2	0.0	0.0	0.0	0.0	0.0
CH4	1.0083-03	8.0655	8.1588	0.0	8.1588
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	23.2844	8.0676	249.2284	169.7592	56.3579
KG/HR	652.1853	129.4550	2477.1415	454.0570	1411.6139
CUM/SEC	2.7401-04	1.3819-04	2.1770-02	9.7230-02	3.5953-03
STATE VARIABLES:					
TEMP C	-169.2568	-151.3295	-157.6000	-176.1000	-176.5239
PRES BAR	7.6000	2.1500	28.1000	3.9000	5.4700
VFRAC	0.0	4.9670-03	1.0000	1.0000	0.1462
LFRAC	1.0000	0.9950	0.0	0.0	0.8538
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0609+08	-8.9035+07	-3.5809+07	-7.4449+06	-1.0336+08
J/KG	-3.7878+06	-5.5487+06	-3.6028+06	-2.7834+06	-4.1267+06
WATT	-6.8621+05	-1.9953+05	-2.4791+06	-3.5107+05	-1.6182+06
ENTROPY:					
J/KMOL-K	-1.6536+04	-1.8353+05	-2.8668+04	-3.7983+04	-3.6057+04
J/KG-K	-590.3831	-1.1438+04	-2884.3464	-1.4201+04	-1439.5751
DENSITY:					
KMOL/CUM	23.6046	16.2173	3.1800	0.4849	4.3542
KG/CUM	661.1542	260.2266	31.6070	1.2972	109.0621
AVG MW	28.0095	16.0461	9.9392	2.6747	25.0473

Capacity increase of 20% by using a pre-reformer and extra CO₂.

L29 L3 L30 L31 L33

STREAM ID	L29	L3	L30	L31	L33
FROM :	T23	E22	V9	T24	V8
TO :	V29	T21	T24	V31	T22
SUBSTREAM: MIXED					
PHASE:	VAPOR	MIXED	MIXED	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	2.6672	169.5058	1.3805	1.3802	6.1273-04
CO	4.4905	63.0220	18.4633	1.5707	36.9187
CO2	0.0	0.0	0.0	0.0	0.0
N2	0.6224	8.5417	3.2674	0.3822	3.4994
O2	0.0	0.0	0.0	0.0	0.0
CH4	7.5965-02	8.1588	0.0	5.5478-13	8.0828
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	7.8562	249.2284	23.1112	3.3332	48.5017
KG/HR	149.8133	2477.1415	611.4703	57.4859	1261.8005
CUM/SEC	2.9641-03	1.9781-02	1.0070-03	8.7966-04	4.9050-03
STATE VARIABLES:					
TEMP C	-177.5619	-164.8000	-176.6341	-177.5691	-181.3971
PRES BAR	5.4000	28.0000	7.8500	7.6000	2.3830
VFRAC	1.0000	0.9774	0.1329	1.0000	0.1110
LFRAC	0.0	2.2554-02	0.8670	0.0	0.8889
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.9862+07	-3.6146+07	-9.9024+07	-5.8027+07	-1.0853+08
J/KG	-3.6636+06	-3.6366+06	-3.7427+06	-3.3646+06	-4.1719+06
WATT	-1.5246+05	-2.5024+06	-6.3571+05	-5.3727+04	-1.4622+06
ENTROPY:					
J/KMOL-K	1.1175+04	-3.1668+04	-1.8745+04	441.2539	-4.0678+04
J/KG-K	586.0345	-3186.1100	-708.4910	25.5853	-1563.5849
DENSITY:					
KMOL/CUM	0.7362	3.4997	6.3748	1.0525	2.7467
KG/CUM	14.0397	34.7848	168.6645	18.1528	71.4583
AVG MW	19.0694	9.9392	26.4576	17.2463	26.0155

Capacity increase of 20% by using a pre-reformer and extra CO₂.

L35 L36 L37 L38 L39

STREAM ID	L35	L36	L37	L38	L39
FROM :	E21	SPLIT1	E22B	SPLIT1	SPLIT1
TO :	SPLIT1	E22B	V37	E24	V39
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	1.7515-04	2.7323-05	2.7323-05	9.5631-05	5.2194-05
CO	10.5151	1.6403	1.6403	5.7412	3.1335
CO2	0.0	0.0	0.0	0.0	0.0
N2	1.2476	0.1946	0.1946	0.6812	0.3718
O2	0.0	0.0	0.0	0.0	0.0
CH4	3.3834-03	5.2782-04	5.2782-04	1.8474-03	1.0083-03
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	11.7663	1.8355	1.8355	6.4244	3.5063
KG/HR	329.5331	51.4071	51.4071	179.9250	98.2008
CUM/SEC	3.0025-03	4.6839-04	4.3866-04	1.6394-03	8.9475-04
STATE VARIABLES:					
TEMP C	-155.1000	-155.1000	-160.8000	-155.1000	-155.1000
PRES BAR	9.2000	9.2000	9.1000	9.2000	9.2000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0452+08	-1.0452+08	-1.0472+08	-1.0452+08	-1.0452+08
J/KG	-3.7318+06	-3.7318+06	-3.7390+06	-3.7318+06	-3.7318+06
WATT	-3.4160+05	-5.3290+04	-5.3392+04	-1.8651+05	-1.0180+05
ENTROPY:					
J/KMOL-K	3.5433+04	3.5433+04	3.3762+04	3.5433+04	3.5433+04
J/KG-K	1265.1585	1265.1585	1205.5077	1265.1585	1265.1585
DENSITY:					
KMOL/CUM	1.0885	1.0885	1.1623	1.0885	1.0885
KG/CUM	30.4869	30.4869	32.5531	30.4869	30.4869
AVG MW	28.0064	28.0064	28.0064	28.0064	28.0064

Capacity increase of 20% by using a pre-reformer and extra CO₂.

L4 L40 L41 L42 L43

STREAM ID	L4	L40	L41	L42	L43
FROM :	CO2	V39	E24	V41	V37
TO :	E23A	T24	V41	T22MIX	T22MIX
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
H2	165.4575	5.2194-05	9.5631-05	9.5631-05	2.7323-05
CO	3.1493	3.1335	5.7412	5.7412	1.6403
CO2	0.0	0.0	0.0	0.0	0.0
N2	1.1523	0.3718	0.6812	0.6812	0.1946
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	1.0083-03	1.8474-03	1.8474-03	5.2782-04
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	169.7592	3.5063	6.4244	6.4244	1.8355
KG/HR	454.0570	98.2008	179.9250	179.9250	51.4071
CUM/SEC	9.4906-03	1.0383-03	1.6065-03	5.9692-03	1.6354-03
STATE VARIABLES:					
TEMP C	-201.5000	-156.8291	-157.6000	-167.5125	-171.2923
PRES BAR	27.3000	7.9500	9.1000	2.5000	2.5000
VFRAC	1.0000	1.0000	1.0000	1.0000	1.0000
LFRAC	0.0	0.0	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-8.2016+06	-1.0451+08	-1.0460+08	-1.0460+08	-1.0472+08
J/KG	-3.0663+06	-3.7318+06	-3.7348+06	-3.7348+06	-3.7390+06
WATT	-3.8675+05	-1.0180+05	-1.8666+05	-1.8666+05	-5.3392+04
ENTROPY:					
J/KMOL-K	-6.2914+04	3.6491+04	3.4787+04	4.4561+04	4.3436+04
J/KG-K	-2.3522+04	1302.9424	1242.0889	1591.1113	1550.9302
DENSITY:					
KMOL/CUM	4.9686	0.9380	1.1108	0.2989	0.3117
KG/CUM	13.2896	26.2724	31.1114	8.3728	8.7316
AVG MW	2.6747	28.0064	28.0064	28.0064	28.0064

Capacity increase of 20% by using a pre-reformer and extra CO₂.

L44 L46 L47 L48 L49

STREAM ID	L44	L46	L47	L48	L49
FROM :	T22MIX	E25-27	T22	MIX2	T22
TO :	E25-27	T22	V47	E21	E21
SUBSTREAM: MIXED					
PHASE:	MIXED	LIQUID	LIQUID	MIXED	VAPOR
COMPONENTS: KMOL/HR					
H2	4.5872-04	4.5872-04	1.0488-16	4.0474	1.0714-03
CO	27.4076	27.4076	2.0751-03	6.0633	64.3243
CO2	0.0	0.0	0.0	0.0	0.0
N2	4.1328	4.1328	5.2416-05	1.0047	7.6322
O2	0.0	0.0	0.0	0.0	0.0
CH4	3.3834-03	3.3834-03	8.0655	8.1415	2.0697-02
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	31.5443	31.5443	8.0676	19.2571	71.9784
KG/HR	883.5176	883.5176	129.4550	336.7543	2015.8630
CUM/SEC	1.0005-02	3.2511-04	8.9359-05	1.2264-02	6.1283-02
STATE VARIABLES:					
TEMP C	-183.1444	-184.5000	-150.6756	-172.6032	-184.0300
PRES BAR	2.5000	2.3250	2.2500	2.1500	2.2500
VFRAC	0.4043	0.0	0.0	0.6034	1.0000
LFRAC	0.5956	1.0000	1.0000	0.3965	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0571+08	-1.0808+08	-8.9035+07	-7.5845+07	-1.0510+08
J/KG	-3.7742+06	-3.8587+06	-5.5487+06	-4.3372+06	-3.7527+06
WATT	-9.2626+05	-9.4702+05	-1.9953+05	-4.0571+05	-2.1014+06
ENTROPY:					
J/KMOL-K	235.1010	-2.6093+04	-1.8353+05	-6.5744+04	4.0241+04
J/KG-K	8.3938	-931.6042	-1.1438+04	-3759.5362	1436.8414
DENSITY:					
KMOL/CUM	0.8757	26.9521	25.0788	0.4361	0.3262
KG/CUM	24.5288	754.8952	402.4184	7.6271	9.1372
AVG MW	28.0087	28.0087	16.0461	17.4872	28.0064

Capacity increase of 20% by using a pre-reformer and extra CO₂.

L6 L7 L8 L9 LB

STREAM ID	L6	L7	L8	L9	LB
FROM :	T21	V29	T23	SPLIT2	T21
TO :	V6	MIX2	V8	V9	MIX1
SUBSTREAM: MIXED					
PHASE:	LIQUID	VAPOR	LIQUID	LIQUID	VAPOR
COMPONENTS: KMOL/HR					
H2	2.6678	2.6672	6.1273-04	1.3805	170.8418
CO	41.4093	4.4905	36.9187	18.4633	75.1591
CO2	0.0	0.0	0.0	0.0	0.0
N2	4.1219	0.6224	3.4994	3.2674	13.8957
O2	0.0	0.0	0.0	0.0	0.0
CH4	8.1588	7.5965-02	8.0828	0.0	4.4772-13
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	56.3579	7.8562	48.5017	23.1112	259.8967
KG/HR	1411.6139	149.8133	1261.8005	611.4703	2838.8867
CUM/SEC	6.3119-04	7.0041-03	5.3431-04	2.6565-04	1.9508-02
STATE VARIABLES:					
TEMP C	-169.0964	-181.1061	-171.3306	-171.5000	-171.3117
PRES BAR	27.3000	2.3000	5.4000	27.3470	27.3000
VFRAC	0.0	1.0000	0.0	0.0	1.0000
LFRAC	1.0000	0.0	1.0000	1.0000	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.0336+08	-6.9861+07	-1.0853+08	-9.9024+07	-3.7786+07
J/KG	-4.1267+06	-3.6635+06	-4.1719+06	-3.7427+06	-3.4592+06
WATT	-1.6182+06	-1.5246+05	-1.4622+06	-6.3572+05	-2.7279+06
ENTROPY:					
J/KMOL-K	-3.7730+04	1.7899+04	-4.1162+04	-2.0135+04	-2.6796+04
J/KG-K	-1506.3322	938.6122	-1582.1958	-761.0243	-2453.1805
DENSITY:					
KMOL/CUM	24.8023	0.3115	25.2153	24.1661	3.7006
KG/CUM	621.2319	5.9414	655.9905	639.3788	40.4229
AVG MW	25.0473	19.0694	26.0155	26.4576	10.9231

Capacity increase of 20% by using a pre-reformer and extra CO₂.

	LC	LD	LE	LF	LG
LC LD LE LF LG					

STREAM ID	LC	LD	LE	LF	LG
FROM :	CO1	CO1	CO2	MIX1	SPLIT2
TO :	CO2	SPLIT2	MIX1	CO1	T21
SUBSTREAM: MIXED					
PHASE:	VAPOR	LIQUID	LIQUID	MIXED	LIQUID
COMPONENTS: KMOL/HR					
H2	169.5810	5.3843	4.1235	174.9653	4.0038
CO	71.5056	72.0097	68.3562	143.5153	53.5464
CO2	0.0	0.0	0.0	0.0	0.0
N2	16.3949	12.7434	15.2426	29.1384	9.4760
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	0.0	0.0	4.4772-13	0.0
C2H6	0.0	0.0	0.0	0.0	0.0
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	0.0	0.0	0.0	0.0
H2O	0.0	0.0	0.0	0.0	0.0
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	257.4816	90.1375	87.7223	347.6191	67.0263
KG/HR	2804.0201	2384.8299	2349.9632	5188.8500	1773.3595
CUM/SEC	1.9258-02	1.0361-03	8.0929-04	1.8421-02	7.7044-04
STATE VARIABLES:					
TEMP C	-171.5000	-171.5000	-201.5000	-174.9624	-171.5000
PRES BAR	27.3470	27.3470	27.3000	27.3000	27.3470
VFRAC	1.0000	0.0	0.0	0.6661	0.0
LFRAC	0.0	1.0000	1.0000	0.3338	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-3.6520+07	-9.9024+07	-9.8714+07	-5.3161+07	-9.9024+07
J/KG	-3.3535+06	-3.7427+06	-3.6849+06	-3.5614+06	-3.7427+06
WATT	-2.6120+06	-2.4794+06	-2.4054+06	-5.1333+06	-1.8437+06
ENTROPY:					
J/KMOL-K	-2.7760+04	-2.0135+04	-4.2864+04	-3.0114+04	-2.0135+04
J/KG-K	-2549.1155	-761.0243	-1600.0736	-2017.4628	-761.0243
DENSITY:					
KMOL/CUM	3.7139	24.1661	30.1097	5.2419	24.1661
KG/CUM	40.4452	639.3788	806.5977	78.2448	639.3788
AVG MW	10.8901	26.4576	26.7886	14.9268	26.4576

Capacity increase of 20% by using a pre-reformer and extra CO₂.

LIQ MEASUMP OUT STM STMFD

STREAM ID	LIQ	MEASUMP	OUT	STM	STMFD
FROM :	V2103	V2116	V2118	V2103	DRUMMIX
TO :	SPLIT	----	----	----	V2103
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	MIXED	VAPOR	MIXED
COMPONENTS: KMOL/HR					
H2	0.0	1.9042-07	1.0231	0.0	0.0
CO	0.0	4.2828-08	0.3804	0.0	0.0
CO2	0.0	6.6332-09	1.1361-02	0.0	0.0
N2	0.0	5.3378-09	5.1560-02	0.0	0.0
O2	0.0	0.0	0.0	0.0	0.0
CH4	0.0	5.8723-08	4.9248-02	0.0	0.0
C2H6	0.0	0.0	2.1401-04	0.0	0.0
C3H8	0.0	0.0	1.3686-08	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
MEA	0.0	1.2770-06	4.6927-03	0.0	0.0
H2O	6818.7515	0.2116	0.4310	299.6862	6818.6876
AR	0.0	0.0	0.0	0.0	0.0
TOTAL FLOW:					
KMOL/HR	6818.7515	0.2116	1.9516	299.6862	6818.6876
KG/HR	1.2284+05	3.8121	23.5105	5398.8474	1.2284+05
CUM/SEC	5.6654-02	1.4178-06	3.9118-04	0.1286	0.1944
STATE VARIABLES:					
TEMP C	221.0000	40.0000	35.0000	221.0000	221.0000
PRES BAR	24.0094	10.3000	28.3000	24.0094	24.0094
VFRAC	0.0	0.0	0.7799	1.0000	4.7995-02
LFRAC	1.0000	1.0000	0.2200	0.0	0.9520
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.7130+08	-2.8770+08	-8.9605+07	-2.3648+08	-2.6962+08
J/KG	-1.5059+07	-1.5970+07	-7.4384+06	-1.3127+07	-1.4967+07
WATT	-5.1386+08	-1.6911+04	-4.8578+04	-1.9686+07	-5.1069+08
ENTROPY:					
J/KMOL-K	-1.2552+05	-1.6656+05	-3.8272+04	-5.5051+04	-1.2213+05
J/KG-K	-6967.2642	-9245.2464	-3177.0782	-3055.8504	-6779.5357
DENSITY:					
KMOL/CUM	33.4325	41.4597	1.3859	0.6470	9.7413
KG/CUM	602.2870	746.9079	16.6950	11.6559	175.4911
AVG MW	18.0150	18.0152	12.0462	18.0150	18.0150

Appendix 6

Design of new heat exchangers in the MEA-section.

H-2109 A/B	tube side rich MEA (282)	shell side lean MEA (281)
old flows (kg/hr)	21094.7	19707.9
new flows (kg/hr)	24688.47	23058.30
specific heat (kcal/kg °C)	0.930	0.88
temperatures in/out (°C)	120/88	72/103
U, overall heat transfer coefficient (kcal/m ² °C hr)		615
ΔT_{lm} , log mean temperature difference (°C)		16.5
f_t , temperature correction factor		0.82
ΔT_m , true temperature difference (°C)		13.5
A, old surface installed (m ²)		70.4

The following formula has been used to globally estimate the new temperatures using the old unchanged heat exchangers in combination with the new larger flows.

$$U \cdot A \cdot \Delta T_{lm} = (\text{flow in tubes}) \cdot C_p \cdot (T_1 - T_2) = (\text{flow in shell}) \cdot C_p \cdot (t_2 - t_1) \quad \text{form. A6.1}$$

Iteration gives:

	rich MEA	lean MEA
new temperatures (°C)	120-90	72-101

H-2108	tube side cooling water	shell side lean MEA (281)
old flows (kg/hr)	85593	21958.3
new flows (kg/hr)	-	23058.30
specific heat (kcal/kg °C)	0.998	0.905
temperatures in/out (°C)	25/-	88/40

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U, overall heat transfer coefficient (kcal/m ² °C hr)	700
F_t , temperature correction factor	0.89
A, old surface installed (m ²)	50

Formula A6.1 has been used to estimate the new temperatures using the unchanged heat exchanger.

Iteration gives:

	cooling water	lean MEA
new temperatures (°C)	25-30	88-40

With this result comes a new cooling water flow of 200731.98 kg/hr and a cooling water velocity of 0.9 m/s. This velocity exceeds the maximum allowable velocity of 0.5 m/s.

The calculation of the maximum flows in the absorber and stripping column.

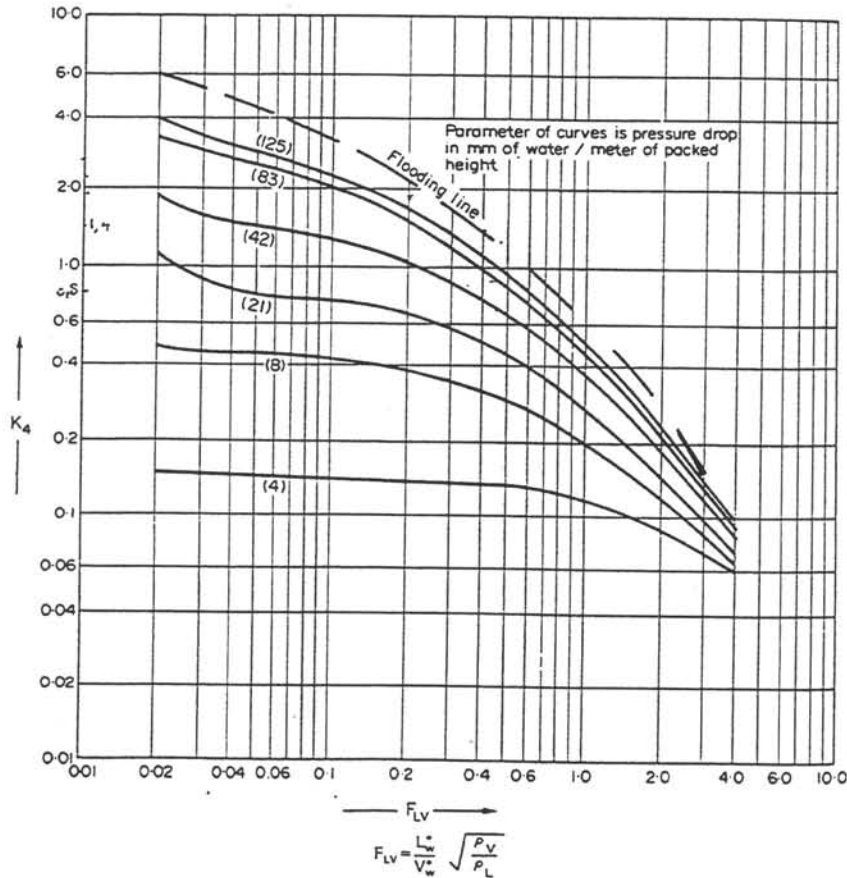


fig. A6.1

The term K_4 is the function:

$$K_4 = \frac{42.9(V_w^*)^2 F_p \left(\frac{\mu_l}{\rho_l}\right)^{0.1}}{\rho_v(\rho_l - \rho_v)}$$

form. 6.2

V_w = gas mass flow-rate per unit column cross sectional area, $\text{kg/m}^2\text{s}$

F_p = packing factor, characteristic of the size and type of packing

μ_l = liquid viscosity, Ns/m^2

ρ_l, ρ_v = liquid and vapour densities, kg/m^3

$$0.8 = \frac{42.9(\text{vapour flow, kg/s})^{2.22} \left(\frac{6.67 \times 10^{-4}}{1076.2} \right)}{\left(\frac{\pi}{4} 0.75^2 \right) 5.66 (1076.2 - 5.66)}$$

For the absorber $F_{iv} = 0.43$ so $K_4 = 0.8$ at 70% flooding. With the current diameter the absorber has a maximum vapour flow of 7365.63 kg/hr.

So the absorber is at half of its maximum capacity in the current process.

For the stripper $F_{iv} = 0.20$ so $K_4 = 1.9$ With the current diameter and at almost 90% flooding the maximum vapour flow is 5184 kg/hr.

$$1.9 = 42.9(\text{vapour flow, kg/s})^{2.29} \left(\frac{3.95 \times 10^{-4}}{824,6118} \right)^{0.1}$$

With an actual flow of 4473.6 kg/hr 17% capacity increase is possible but this is at a very high percentage flooding.

	MEA-absorber V-2105	MEA-stripper V-2106
F_{iv}	0.43	0.196
K_4	0.8	1.9
Max. vapour flow (kg/s)	7365.63	5200
Possible capacity increase (%)	109	16

The new flows in the stripper are the following:

(kg/hr)	KTI 281	New 281
CO ₂	421.7	485
MEA	5321.1	6120
H ₂ O	13965.1	15920
Total mass flow	19707.9	22525

The new flows in the absorber are the following:

	KTI 212	New 212
H ₂	300.5	326.6
CO	1498.1	1811.32
CO ₂	1380.6	1615,68
N ₂	245.4	225,4
CH ₄	69	86,72
H ₂ O	27.9	31,4
Total mass flow	2138.7	4097,12

Report for Block: V2105
 Description:

Stage	LIQUID SIGMAMX	LIQUID RHOMX	VAPOR RHOMX	LIQUID MUMX	VAPOR MUMX	LIQUID MASSFLMX	VAPOR MASSFLMX
	N/M	KG/CUM	KG/CUM	N-SEC/SQM	N-SEC/SQM	KG/SEC	KG/SEC
1	.0881417	907.0558	3.896868	5.65301E-4	1.40398E-5	5.474465	.5948598
2	.0881387	907.0505	3.898998	5.65281E-4	1.40402E-5	5.474528	.5949079
3	.0881346	907.0432	3.901176	5.65254E-4	1.40408E-5	5.474614	.5949711
4	.0881291	907.0331	3.903423	5.65218E-4	1.40415E-5	5.474731	.5950573
5	.0881215	907.0194	3.905762	5.65168E-4	1.40426E-5	5.474891	.5951746
6	.0881113	907.0008	3.908228	5.65100E-4	1.40439E-5	5.475106	.5953338
7	.0880975	906.9756	3.910861	5.65009E-4	1.40458E-5	5.475397	.5955491
8	.0880788	906.9414	3.913721	5.64886E-4	1.40483E-5	5.475790	.5958402
9	.0880536	906.8953	3.916885	5.64719E-4	1.40516E-5	5.476321	.5962333
10	.0880197	906.8329	3.920458	5.64495E-4	1.40561E-5	5.477036	.5967636
11	.0879739	906.7486	3.924580	5.64192E-4	1.40622E-5	5.478000	.5974787
12	.0879122	906.6346	3.929437	5.63784E-4	1.40704E-5	5.479300	.5984432
13	.0878289	906.4805	3.935282	5.63233E-4	1.40814E-5	5.481055	.5997437
14	.0877165	906.2715	3.942443	5.62489E-4	1.40963E-5	5.483424	.6014981
15	.0875639	905.9862	3.951346	5.61479E-4	1.41164E-5	5.486629	.6038671
16	.0873537	905.5866	3.962456	5.60085E-4	1.41441E-5	5.491002	.6070727
17	.0870436	904.9662	3.975760	5.58016E-4	1.41842E-5	5.497170	.6114449
18	.0864608	903.6369	3.987306	5.54064E-4	1.42557E-5	5.507290	.6176137
19	.0847227	899.0018	3.970738	5.42024E-4	1.44553E-5	5.534908	.6277335
20	.0758731	887.4863	3.839470	4.85324E-4	1.51915E-5	5.854966	.6553509

Report for Block: V2106
 Description:

Stage	LIQUID SIGMAMX	LIQUID RHOMX	VAPOR RHOMX	LIQUID MUMX	VAPOR MUMX	LIQUID MASSFLMX	VAPOR MASSFLMX
	N/M	KG/CUM	KG/CUM	N-SEC/SQM	N-SEC/SQM	KG/SEC	KG/SEC
1	.0737715	915.7708	1.510871	4.49126E-4	1.59247E-5	.9140364	.6588332
2	.0705134	905.5043	1.131136	4.29653E-4	1.41703E-5	.9317327	1.572870
3	.0704665	904.8457	1.128907	4.29423E-4	1.41566E-5	1.265030	1.590499
4	.0700196	896.8499	1.122500	4.27446E-4	1.41176E-5	1.279534	1.649576
5	.0639577	839.4263	1.091829	4.00805E-4	1.39529E-5	6.710503	1.217759
6	.0638935	838.2835	1.053035	4.00317E-4	1.36796E-5	6.716703	1.240083
7	.0638905	837.7270	1.034668	4.00252E-4	1.35445E-5	6.719707	1.246283
8	.0639014	837.3967	1.024062	4.00290E-4	1.34647E-5	6.721461	1.249287
9	.0639156	837.1726	1.017038	4.00357E-4	1.34111E-5	6.722644	1.251041
10	.0639306	837.0046	1.011891	4.00433E-4	1.33714E-5	6.723507	1.252224
11	.0639459	836.8678	1.007792	4.00514E-4	1.33395E-5	6.724185	1.253087
12	.0639618	836.7483	1.004283	4.00600E-4	1.33121E-5	6.724760	1.253765
13	.0639790	836.6368	1.001081	4.00695E-4	1.32868E-5	6.725276	1.254340
14	.0639985	836.5258	.9979685	4.00803E-4	1.32621E-5	6.725777	1.254856
15	.0640222	836.4078	.9947418	4.00936E-4	1.32364E-5	6.726311	1.255357
16	.0640528	836.2663	.9911574	4.01110E-4	1.32075E-5	6.727025	1.255890
17	.0640698	835.8208	.9868947	4.01268E-4	1.31728E-5	6.732074	1.256605

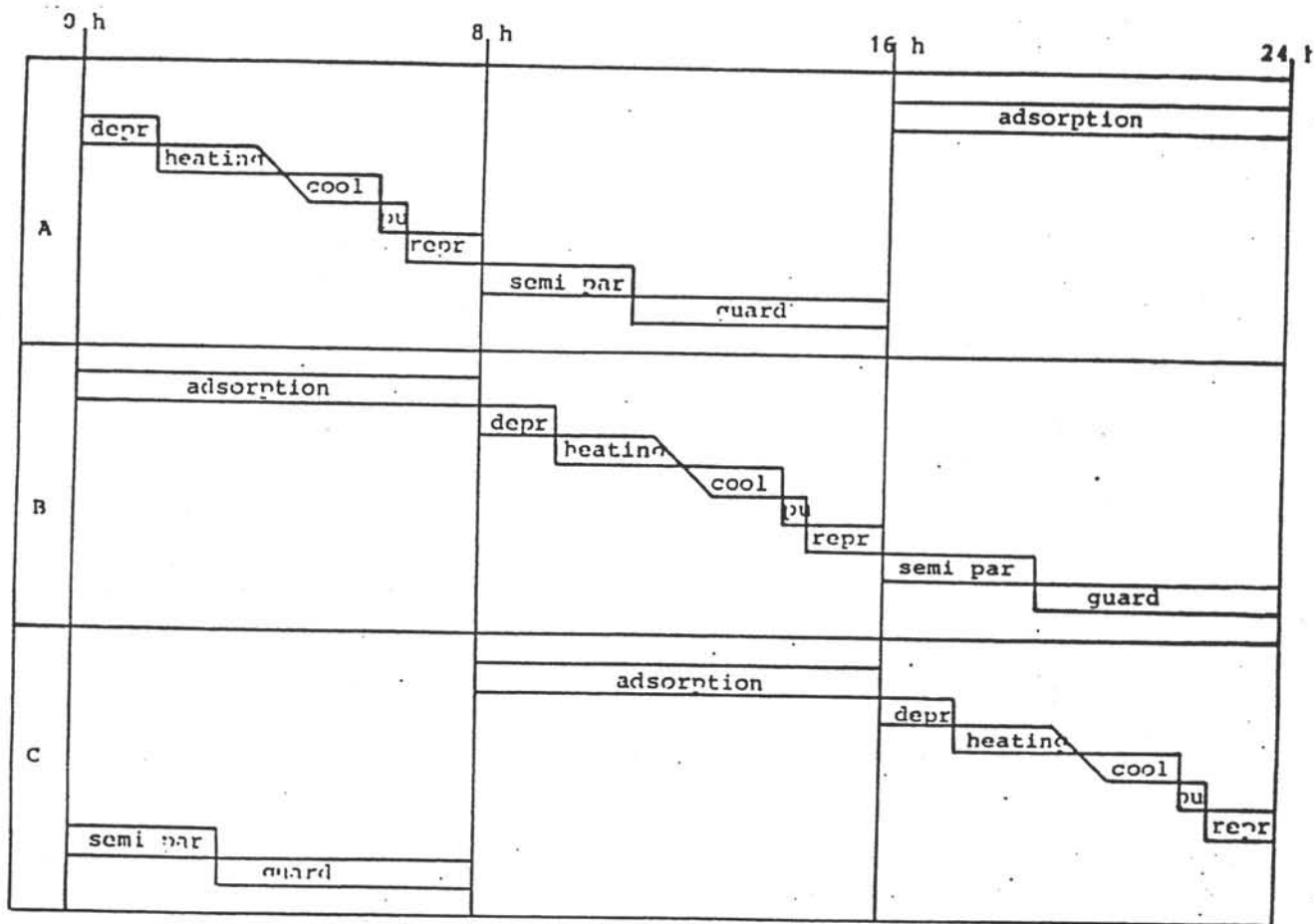


Fig. A6-2: sequence chart process drying

APPENDIX 7

Capacities of the columns

The actual capacities of the columns in the coldbox are estimated with the following equations and diagrams:

$$F_{lg} = \frac{M_l}{M_g} \sqrt{\frac{\rho_g}{\rho_l}} \quad \text{form. A7.1}$$

With this equation the C_{tr} is determined from the Fair diagram.

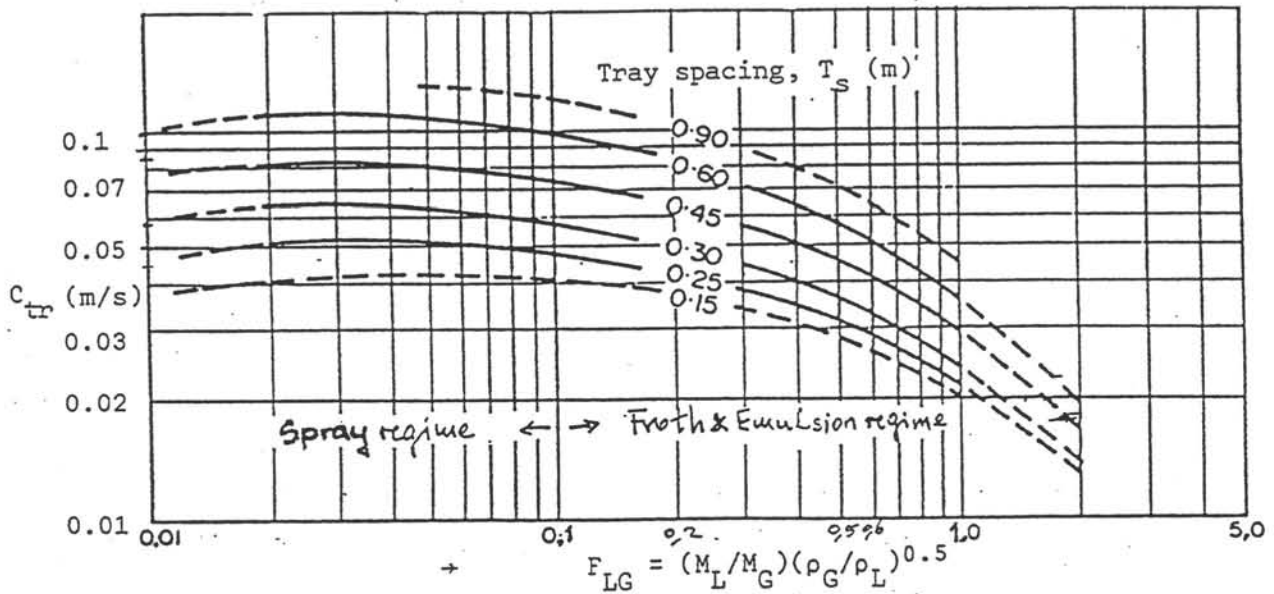


figure A7.1 Flooding gas capacity, sieve (and valve) trays

$$U_{g,max} = C_{tr,v} \sqrt{\frac{\rho_l - \rho_g}{\rho_g}} \quad \text{form. A7.2}$$

This is the Souders equation which gives the maximum gas velocity of the gas in meter per second.

The C_{tr} is multiplied by the surface tension correction factor (STCF), because of the low liquid surface tension, which gives a deviation from the standard case.

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$$C_{tr,new} = C_{tr} \left(\frac{\sigma}{0.02} \right)^{0.2}$$

form. A7.3

$$M_{g,max} = \rho_g U_{g,max} \left(\frac{\pi}{4} d_{col}^2 \right)$$

form. A7.4

This gives the maximum gas flow in kg/s.

	T 21	T 22	T 23	T 24
F_{lg}	0.15	0.044	1.89	0.44
C_{tr}	0.066	0.070	0.018	0.045
STFC	0.75	0.82	0.82	0.74
$U_{g,max}$ (m/s)	0.201	0.518	0.114	0.154
$M_{g,max}$ (kg/s)	0.938	0.595	0.168	0.572
$M_{g,old}$ (kg/s)	0.500	0.347	0.0164	0.0587
maximum capacity increase (%)	87.5	71.4	925	874.6

Report for Block: T21
Description:

Stage	LIQUID SIGMAMX	LIQUID RHOMX	VAPOR RHOMX	LIQUID MASSFLMX	VAPOR MASSFLMX	LIQUID MUMX	VAPOR MUMX
	N/M	KG/CUM	KG/CUM	KG/SEC	KG/SEC	N-SEC/SQM	N-SEC/SQM
1	4.79088E-3	652.6718	36.66274	.3054832	.4863845	7.35368E-5	6.35757E-6
2	4.74532E-3	650.7390	37.08353	.3060007	.4992052	7.30601E-5	6.39620E-6
3	4.74407E-3	650.5099	37.08773	.3057463	.4997225	7.30445E-5	6.39917E-6
4	4.74566E-3	650.4575	37.06915	.3055508	.4994685	7.30584E-5	6.39908E-6
5	4.74663E-3	650.4266	37.05930	.3054394	.4992735	7.30660E-5	6.39896E-6
6	4.74702E-3	650.4003	37.05664	.3053814	.4991625	7.30675E-5	6.39901E-6
7	4.74705E-3	650.3757	37.05833	.3053545	.4991047	7.30654E-5	6.39916E-6
8	4.74688E-3	650.3522	37.06253	.3053454	.4990778	7.30612E-5	6.39938E-6
9	4.74660E-3	650.3292	37.06817	.3053465	.4990686	7.30558E-5	6.39963E-6
10	4.74625E-3	650.3066	37.07463	.3053534	.4990696	7.30497E-5	6.39991E-6
11	4.74586E-3	650.2842	37.08155	.3053637	.4990763	7.30433E-5	6.40020E-6
12	4.74545E-3	650.2618	37.08875	.3053757	.4990864	7.30366E-5	6.40049E-6
13	4.74503E-3	650.2394	37.09608	.3053879	.4990981	7.30298E-5	6.40079E-6
14	4.74463E-3	650.2167	37.10342	.3053967	.4991101	7.30232E-5	6.40109E-6
15	4.74438E-3	650.1921	37.11044	.3053853	.4991186	7.30182E-5	6.40139E-6
16	4.74486E-3	650.1585	37.11560	.3052723	.4991070	7.30207E-5	6.40167E-6
17	4.74895E-3	650.0796	37.11144	.3046550	.4989939	7.30609E-5	6.40186E-6
18	4.77106E-3	649.7726	37.06098	.3015641	.4983765	7.32882E-5	6.40162E-6
19	4.88223E-3	648.2548	36.78384	.2870288	.4952856	7.44282E-5	6.39922E-6
20	5.41872E-3	639.1665	35.46829	.2569722	.4807500	7.96364E-5	6.38667E-6

Report for Block: T22
Description:

Stage	LIQUID SIGMAMX	LIQUID RHOMX	VAPOR RHOMX	LIQUID MASSFLMX	VAPOR MASSFLMX	LIQUID MUMX	VAPOR MUMX
	N/M	KG/CUM	KG/CUM	KG/SEC	KG/SEC	N-SEC/SQM	N-SEC/SQM
1	7.31197E-3	753.2163	9.133204	.1384531	.3470472	1.11930E-4	5.76710E-6
2	7.31579E-3	752.9691	9.126254	.1383641	.3470086	1.11940E-4	5.76724E-6
3	7.31701E-3	752.8975	9.124261	.1383384	.3469197	1.11943E-4	5.76727E-6
4	7.31743E-3	752.8748	9.123677	.1383290	.3468940	1.11944E-4	5.76729E-6
5	7.31777E-3	752.8546	9.123430	.1383139	.3468845	1.11947E-4	5.76732E-6
6	7.31930E-3	752.7674	9.122880	.1382357	.3468695	1.11965E-4	5.76749E-6
7	7.32934E-3	752.2568	9.119863	.1377752	.3467912	1.12072E-4	5.76851E-6
8	7.39643E-3	749.2249	9.102093	.1351342	.3463308	1.12695E-4	5.77455E-6
9	7.80410E-3	732.2524	9.015192	.3522641	.3201784	1.15834E-4	5.80856E-6
10	7.84098E-3	730.5352	8.999117	.3493320	.3202085	1.16066E-4	5.81129E-6
11	8.31009E-3	711.3594	8.886272	.3199710	.3172764	1.19027E-4	5.84672E-6
12	.0110806	566.1318	7.894073	.0320555	.2879155	1.26574E-4	6.12208E-6

Report for Block: T23

Description:

Stage	LIQUID SIGMAMX	LIQUID RHOMX	VAPOR RHOMX	LIQUID MASSFLMX	VAPOR MASSFLMX	LIQUID MUMX	VAPOR MUMX
	N/M	KG/CUM	KG/CUM	KG/SEC	KG/SEC	N-SEC/SQM	N-SEC/SQM
_1	7.62474E-3	714.1853	11.70613	.2417437	.0163610	1.11528E-4	5.77722E-6
_2	7.60289E-3	713.7339	11.78829	.2420287	1.13261E-3	1.11296E-4	5.78930E-6
_3	7.32268E-3	707.7805	12.89407	.2456082	1.41756E-3	1.08324E-4	5.94174E-6
_4	6.33154E-3	682.7448	17.84108	.2406111	4.99709E-3	9.79938E-5	6.46688E-6

Report for Block: T24

Description:

Stage	LIQUID SIGMAMX	LIQUID RHOMX	VAPOR RHOMX	LIQUID MASSFLMX	VAPOR MASSFLMX	LIQUID MUMX	VAPOR MUMX
	N/M	KG/CUM	KG/CUM	KG/SEC	KG/SEC	N-SEC/SQM	N-SEC/SQM
_1	6.53091E-3	674.5717	26.25645	.1185112	.0377395	8.21173E-5	6.85642E-6
_2	4.68666E-3	662.1656	29.47594	.1217341	.0553528	7.86710E-5	7.06115E-6
_3	4.33531E-3	661.2773	29.63363	.1218169	.0585758	7.85070E-5	7.07235E-6
_4	4.29080E-3	660.8510	29.60289	.1207022	.0586585	7.85221E-5	7.07358E-6
_5	4.28863E-3	660.8447	29.60244	.1190000	1.70217E-3	7.85228E-5	7.07359E-6

APPENDIX 8. SIMULATION AND RESULTS OF CRYOGENIC METHODS I & II

A8.1. Simulation and results of cryogenic method II on Aspen Plus

Before using Aspen Plus, the patent was first simulated on ChemCad, in order to gain more sufficient specifications to make simulation on Aspen Plus successful.

On Aspen Plus, only the separating part of the patent was simulated. The reason why the simulation of the heat balance is omitted, is that Aspen Plus does not have a suitable enthalpy routine for simulating multifold heat exchangers, because of the inaccuracy of the enthalpy calculations (see chapter 12).

Figure A8.1 on the next page shows the adapted flowsheet of the second method. The stream and equipment numbers are not the same as the numbers shown on the flowsheet.

Results

The main streams of this method are listed in Table A8-1 on page A8-35.

According to the print-outs the following conclusions can be made:

- About 80% of the N_2 can be removed from the natural gas feed.
- Only 3.3% nitrogen remains in the new feed stream.
- The CH_4 -purity of the natural gas feed stream raised from 81.3% to 91.8%.
- Only 0.03 kmol/hr CH_4 leaves the plant with the nitrogen stream (ID 16), which is only 0.068% of the CH_4 -feed stream

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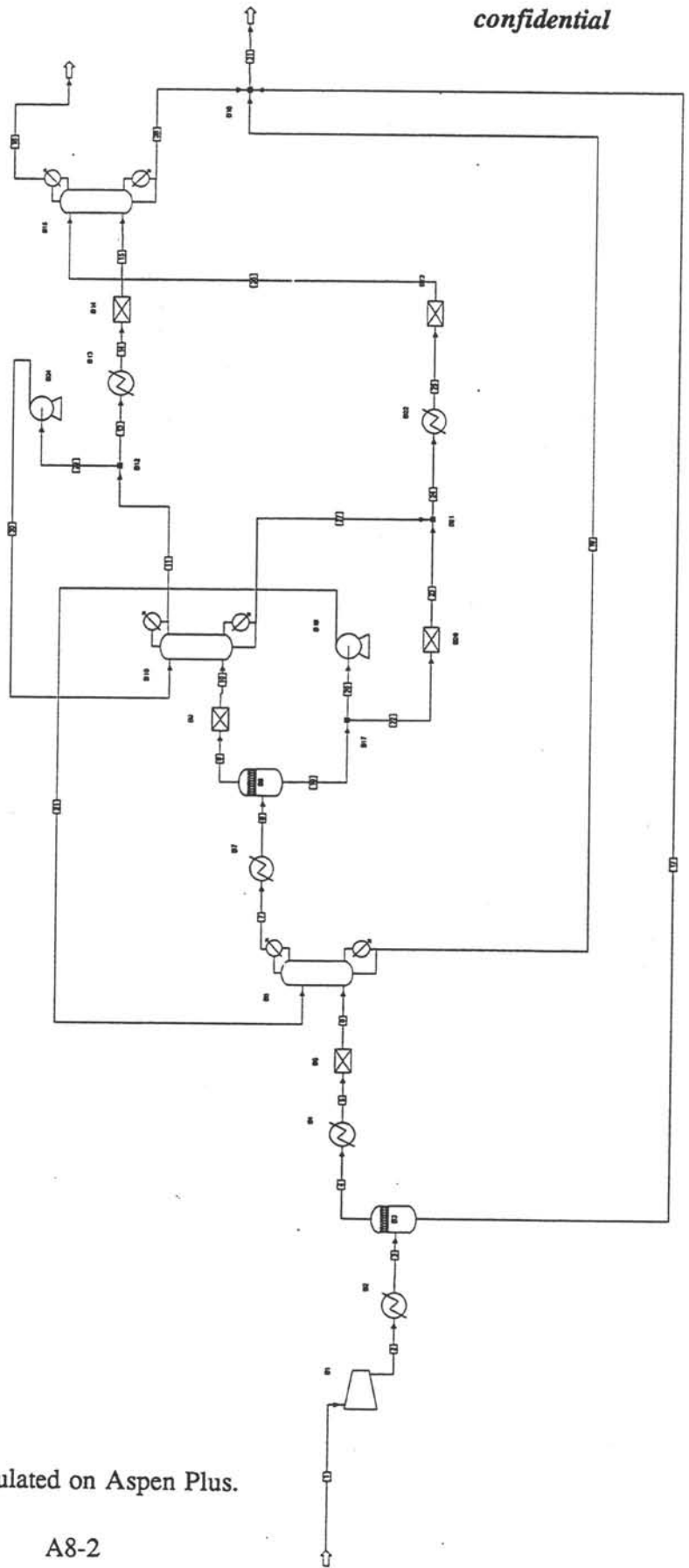


Fig. A8.1. Flowsheet of method II simulated on Aspen Plus.

U-O-S BLOCK SECTION

BLOCK: B1 MODEL: COMPR

 INLET STREAM: 1
 OUTLET STREAM: 2
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

	***	MASS AND ENERGY BALANCE	***	
		IN	OUT	RELATIVE DIFF.
TOTAL BALANCE				
MOLE (KMOL/HR)		53.1084	53.1084	0.
MASS (KG/SEC)		0.274843	0.274843	0.
ENTHALPY (WATT)		-0.100652E+07	-959938.	-0.462820E-01

*** INPUT DATA ***

GAS PHASE CALCULATION

NO FLASH PERFORMED

TYPE : POLYTROPIC CENTRIFUGAL COMPRESSOR

PRESSURE RATIO

2.40000

OUTLET TEMPERATURE C

103.116

MECHANICAL EFFICIENCY

1.00000

*** RESULTS ***

INDICATED HORSEPOWER REQUIREMENT	WATT	46,583.9
BRAKE HORSEPOWER REQUIREMENT	WATT	46,583.9
NET WORK,	WATT	-46,583.9
CALCULATED OUTLET PRES	BAR	48.0000
CALC EFFICIENCY (POLYTR/ISENTR)		0.73510
OUTLET VAPOR FRACTION		1.00000

BLOCK: B10 MODEL: RADFRAC

 INLETS - 10 STAGE 5
 30 STAGE 2
 OUTLETS - 11 STAGE 1
 27 STAGE 6
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

	***	MASS AND ENERGY BALANCE	***	
		IN	OUT	RELATIVE DIFF.
TOTAL BALANCE				
MOLE (KMOL/HR)		5.46843	5.46844	-0.592896E-06
MASS (KG/SEC)		0.373975E-01	0.373975E-01	-0.587825E-06
ENTHALPY (WATT)		-42758.6	-46792.9	0.862164E-01

ASPEN PLUS VER: SUN-4 REL: 8.5-4 INST: DELEFTUOT 05/02/93 PAGE

U-O-S BLOCK SECTION

BLOCK: B10 MODEL: RADFRAC (CONTINUED)

**** INPUT DATA ****

**** INPUT PARAMETERS ****

NUMBER OF STAGES	6
ALGORITHM OPTION	STANDARD
ABSORBER OPTION	NO
INITIALIZATION OPTION	STANDARD
HYDRAULIC PARAMETER CALCULATIONS	NO
INSIDE LOOP CONVERGENCE METHOD	BROYDEN
DESIGN SPECIFICATION METHOD	NESTED
MAXIMUM NO. OF OUTSIDE LOOP ITERATIONS	25
MAXIMUM NO. OF INSIDE LOOP ITERATIONS	10
MAXIMUM NUMBER OF FLASH ITERATIONS	50
FLASH TOLERANCE	0.000100000
OUTSIDE LOOP CONVERGENCE TOLERANCE	0.000100000

**** COL-SPECS ****

MOLAR VAPOR DIST / TOTAL DIST	0.0	
MOLAR REFLUX RATIO	0.82151	
REBOILER DUTY	WATT	404.750

**** PROFILES ****

P-SPEC	STAGE 1	PRES, BAR	28.0000
--------	---------	-----------	---------

U-O-S BLOCK SECTION

BLOCK: B10 MODEL: RADFRAC (CONTINUED)

 **** RESULTS ****

TOP STAGE TEMPERATURE	C	-149.434
BOTTOM STAGE TEMPERATURE	C	-122.473
TOP STAGE LIQUID FLOW	KMOL/HR	5.30635
BOTTOM STAGE LIQUID FLOW	KMOL/HR	1.96602
TOP STAGE VAPOR FLOW	KMOL/HR	0.0
BOTTOM STAGE VAPOR FLOW	KMOL/HR	1.56339
MOLAR REFLUX RATIO		1.51505
MOLAR BOILUP RATIO		0.79521
CONDENSER DUTY (W/O SUBCOOL)	WATT	-6,012.93
REBOILER DUTY	WATT	1,978.72

**** MANIPULATED VARIABLES ****

REBOILER DUTY	WATT	1,978.72
MOLAR REFLUX RATIO		1.51505

**** DESIGN SPECIFICATIONS ****

NO	SPEC-TYPE	UNIT	SPECIFIED VALUE	CALCULATED VALUE	ABS. ERROR
1	MOLE-FRAC		0.30000	0.30000	0.10055E-08
2	MOLE-FRAC		0.50000E-01	0.50000E-01	-0.58788E-09

**** MAXIMUM FINAL RELATIVE ERRORS ****

DEW POINT	0.44439E-05	STAGE= 2
BUBBLE POINT	0.72003E-05	STAGE= 3
COMPONENT MASS BALANCE	0.36807E-06	STAGE= 1 COMP=N2
ENERGY BALANCE	0.34618E-04	STAGE= 2

**** PROFILES ****

STAGE	TEMPERATURE C	PRESSURE BAR	ENTHALPY J/KMOL		HEAT DUTY WATT
			LIQUID	VAPOR	
1	-149.43	28.000	-0.12691E+08	-0.81658E+07	-6012.9298
2	-146.92	28.200	-0.18742E+08	-0.10234E+08	
3	-141.95	28.400	-0.30973E+08	-0.14435E+08	
4	-133.81	28.600	-0.47611E+08	-0.22233E+08	
5	-130.25	28.800	-0.53083E+08	-0.26087E+08	
6	-122.47	29.000	-0.63074E+08	-0.35963E+08	1978.7217

U-O-S BLOCK SECTION

BLOCK: B10 MODEL: RADFRAC (CONTINUED)

STAGE	FLOW RATE KMOL/HR		FEED RATE KMOL/HR			PRODUCT RATE KMOL/HR	
	LIQUID	VAPOR	LIQUID	VAPOR	MIXED	LIQUID	VAPOR
1	5.306	0.					
2	5.730	8.809	1.7512			3.5024	
3	4.386	7.482					
4	3.600	6.138			3.6521		
5	3.529	1.699	.65043-01				
6	1.966	1.563					1.9660

**** X-PROFILE ****

STAGE	CH4	C2H6	N2	O2	CO2
1	.50000E-01	.61921E-13	.94975	.25016E-03	.13923E-11
2	.12652	.15372E-11	.87316	.31890E-03	.23382E-10
3	.28207	.41403E-10	.71756	.36405E-03	.42158E-09
4	.49628	.10701E-08	.50339	.33852E-03	.72854E-08
5	.56775	.15966E-08	.43191	.34559E-03	.98664E-08
6	.69969	.27739E-08	.30000	.31188E-03	.16888E-07

**** Y-PROFILE ****

STAGE	CH4	C2H6	N2	O2	CO2
1	.20907E-01	.33152E-14	.97890	.19466E-03	.10775E-12
2	.50000E-01	.61921E-13	.94975	.25016E-03	.13923E-11
3	.10861	.11932E-11	.89109	.30281E-03	.18288E-10
4	.21586	.29609E-10	.78381	.33155E-03	.30176E-09
5	.26828	.49054E-10	.73135	.36522E-03	.45067E-09
6	.40182	.11609E-09	.59779	.38799E-03	.10365E-08

**** K-VALUES ****

STAGE	CH4	C2H6	N2	O2	CO2
1	.41813	.53536E-01	1.0307	.77812	.77383E-01
2	.39518	.40281E-01	1.0877	.78444	.59543E-01
3	.38503	.28815E-01	1.2418	.83177	.43374E-01
4	.43495	.27669E-01	1.5571	.97942	.41418E-01
5	.47254	.30724E-01	1.6933	1.0568	.45678E-01
6	.57429	.41852E-01	1.9926	1.2440	.61374E-01

BLOCK: B12 MODEL: FSPLIT

 INLET STREAM: 11
 OUTLET STREAMS: 13 29
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	3.50242	3.50242	0.
MASS (KG/SEC)	0.266724E-01	0.266724E-01	0.
ENTHALPY (WATT)	-12347.0	-12347.0	0.

*** INPUT DATA ***

U-O-S BLOCK SECTION

BLOCK: B12 MODEL: FSPLIT (CONTINUED)

FRACTION OF FLOW STRM=13 FRAC= 0.50000

*** RESULTS ***

STREAM= 13 SPLIT= 0.50000 KEY= 0
 29 0.50000 0

BLOCK: B13 MODEL: HEATER

 INLET STREAM: 13
 OUTLET STREAM: 14
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	1.75121	1.75121	0.
MASS (KG/SEC)	0.133362E-01	0.133362E-01	0.130076E-15
ENTHALPY (WATT)	-6173.52	-6221.28	0.767771E-02

*** INPUT DATA ***

TWO PHASE PV FLASH
 SPECIFIED PRESSURE BAR 27.0000
 VAPOR FRACTION 0.0
 MAXIMUM NO. ITERATIONS 30
 CONVERGENCE TOLERANCE 0.000100000

*** RESULTS ***

OUTLET TEMPERATURE C -150.24
 OUTLET PRESSURE BAR 27.000
 HEAT DUTY WATT -47.765
 VAPOR FRACTION 0.

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.50000E-01	0.50000E-01	0.19645E-01	0.39291
C2H6	0.56346E-13	0.56346E-13	0.24485E-14	0.43453E-01
N2	0.94975	0.94975	0.98016	1.0321
O2	0.25016E-03	0.25016E-03	0.19104E-03	0.76370
CO2	0.11648E-11	0.11648E-11	0.74908E-13	0.64310E-01

U-O-S BLOCK SECTION

BLOCK: B14 MODEL: HEATER

 INLET STREAM: 14
 OUTLET STREAM: 15
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	1.75121	1.75121	0.
MASS (KG/SEC)	0.133362E-01	0.133362E-01	0.
ENTHALPY (WATT)	-6221.28	-6221.28	-0.337016E-06

*** INPUT DATA ***

TWO PHASE PQ FLASH
 SPECIFIED PRESSURE BAR 2.33000
 SPECIFIED HEAT DUTY WATT 0.0
 MAXIMUM NO. ITERATIONS 30
 CONVERGENCE TOLERANCE 0.000100000

*** RESULTS ***

OUTLET TEMPERATURE C -186.68
 OUTLET PRESSURE BAR 2.3300
 VAPOR FRACTION 0.52882

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.50000E-01	0.10102	0.45441E-02	0.44984E-0
C2H6	0.56346E-13	0.11958E-12	0.28990E-17	0.24242E-0
N2	0.94975	0.89861	0.99532	1.1077
O2	0.25016E-03	0.37405E-03	0.13978E-03	0.37370
CO2	0.11648E-11	0.24719E-11	0.13122E-15	0.53083E-0

BLOCK: B15 MODEL: RADFRAC

 INLETS - 15 STAGE 4
 26 STAGE 2
 OUTLETS - 16 STAGE 1
 28 STAGE 5
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

U-O-S BLOCK SECTION

BLOCK: B15 MODEL: RADFRAC (CONTINUED)

*** MASS AND ENERGY BALANCE ***			
TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	16.3147	16.3147	0.217761E-15
MASS (KG/SEC)	0.939035E-01	0.939035E-01	0.591151E-15
ENTHALPY (WATT)	-264610.	-257924.	-0.252700E-01

 **** INPUT DATA ****

**** INPUT PARAMETERS ****

NUMBER OF STAGES	5
ALGORITHM OPTION	STANDARD
ABSORBER OPTION	NO
INITIALIZATION OPTION	STANDARD
HYDRAULIC PARAMETER CALCULATIONS	NO
INSIDE LOOP CONVERGENCE METHOD	BROYDEN
DESIGN SPECIFICATION METHOD	NESTED
MAXIMUM NO. OF OUTSIDE LOOP ITERATIONS	25
MAXIMUM NO. OF INSIDE LOOP ITERATIONS	10
MAXIMUM NUMBER OF FLASH ITERATIONS	50
FLASH TOLERANCE	0.000100000
OUTSIDE LOOP CONVERGENCE TOLERANCE	0.000100000

**** COL-SPECS ****

MOLAR VAPOR DIST / TOTAL DIST	1.00000
MOLAR REFLUX RATIO	0.42481
REBOILER DUTY	WATT 13,097.5

**** PROFILES ****

P-SPEC	STAGE 1 PRES, BAR	1.50000
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U-O-S BLOCK SECTION

BLOCK: B15 MODEL: RADFRAC (CONTINUED)

 **** RESULTS ****

TOP STAGE TEMPERATURE	C	-190.717
BOTTOM STAGE TEMPERATURE	C	-154.181
TOP STAGE LIQUID FLOW	KMOL/HR	4.75544
BOTTOM STAGE LIQUID FLOW	KMOL/HR	10.2204
TOP STAGE VAPOR FLOW	KMOL/HR	6.09426
BOTTOM STAGE VAPOR FLOW	KMOL/HR	6.12364
MOLAR REFLUX RATIO		0.78031
MOLAR BOILUP RATIO		0.59916
CONDENSER DUTY (W/O SUBCOOL)	WATT	-8,854.12
REBOILER DUTY	WATT	15,541.1

**** MANIPULATED VARIABLES ****

REBOILER DUTY	WATT	15,541.1
MOLAR REFLUX RATIO		0.78031

**** DESIGN SPECIFICATIONS ****

NO	SPEC-TYPE	UNIT	SPECIFIED VALUE	CALCULATED VALUE	ABS. ERROR
1	MOLE-FRAC		0.30000E-01	0.29982E-01	0.18348E-04
2	MOLE-FRAC		0.99500	0.99500	0.97063E-07

**** MAXIMUM FINAL RELATIVE ERRORS ****

DEW POINT	0.86568E-04	STAGE=	5
BUBBLE POINT	0.26321E-03	STAGE=	2
COMPONENT MASS BALANCE	0.92157E-08	STAGE=	2 COMP=CH4
ENERGY BALANCE	0.41431E-04	STAGE=	3

**** PROFILES ****

STAGE	TEMPERATURE C	PRESSURE BAR	ENTHALPY J/KMOL		HEAT DUTY WATT
			LIQUID	VAPOR	
1	-190.72	1.5000	-0.23072E+08	-0.67438E+07	-8854.1177
2	-181.79	1.7500	-0.62149E+08	-0.98671E+07	
3	-177.95	2.0000	-0.67550E+08	-0.11770E+08	
4	-167.79	2.2500	-0.79425E+08	-0.23268E+08	
5	-154.18	2.5000	-0.86829E+08	-0.57933E+08	.15541+05

U-O-S BLOCK SECTION

BLOCK: B15 MODEL: RADFRAC (CONTINUED)

STAGE	FLOW RATE KMOL/HR		FEED RATE KMOL/HR			PRODUCT RATE KMOL/HR	
	LIQUID	VAPOR	LIQUID	VAPOR	MIXED	LIQUID	VAPOR
1	4.755	6.094		2.4953			6.0942
2	16.30	8.354	12.0681				
3	16.00	7.834		0.9305			
4	16.34	6.599	0.8206				
5	10.22	6.124				10.2204	

**** X-PROFILE ****

STAGE	CH4	C2H6	N2	O2	CO2
1	.14065	.22718E-11	.85894	.41507E-03	.14428E-10
2	.63900	.22316E-07	.36060	.39702E-03	.95132E-07
3	.70973	.22744E-07	.28969	.58096E-03	.96958E-07
4	.86648	.22291E-07	.13292	.59102E-03	.95095E-07
5	.96971	.35597E-07	.29982E-01	.30834E-03	.15174E-06

**** Y-PROFILE ****

STAGE	CH4	C2H6	N2	O2	CO2
1	.48577E-02	0.	.99500	.14242E-03	0.
2	.49421E-01	.59849E-12	.95032	.26155E-03	.39305E-11
3	.75870E-01	.12167E-11	.92365	.47989E-03	.78221E-11
4	.23142	.70233E-11	.76760	.97771E-03	.44141E-10
5	.69420	.84400E-10	.30474	.10628E-02	.54755E-09

**** K-VALUES ****

STAGE	CH4	C2H6	N2	O2	CO2
1	.34534E-01	.88838E-05	1.1584	.34310	.18871E-04
2	.77344E-01	.26850E-04	2.6346	.65855	.41351E-04
3	.10691	.53552E-04	3.1875	.82593	.80764E-04
4	.26707	.31501E-03	5.7748	1.6543	.46411E-03
5	.71582	.23700E-02	10.165	3.4469	.36067E-02

BLOCK: B16 MODEL: MIXER

 INLET STREAMS: 17 18 28
 OUTLET STREAM: 31
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	47.0146	47.0146	0.
MASS (KG/SEC)	0.227521	0.227521	0.
ENTHALPY (WATT)	-0.114290E+07	-0.114290E+07	-0.353128E-05

U-O-S BLOCK SECTION

BLOCK: B16 MODEL: MIXER (CONTINUED)

*** INPUT DATA ***

TWO PHASE FLASH	
MAXIMUM NO. ITERATIONS	30
CONVERGENCE TOLERANCE	0.000100000
OUTLET PRESSURE BAR	15.0000

BLOCK: B17 MODEL: FSPLIT

 INLET STREAM: 19
 OUTLET STREAMS: 20 22
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF
TOTAL BALANCE			
MOLE (KMOL/HR)	16.2548	16.2548	0.
MASS (KG/SEC)	0.901189E-01	0.901189E-01	0.
ENTHALPY (WATT)	-274641.	-274641.	0.211941E-15

*** INPUT DATA ***

FRACTION OF FLOW STRM=20 FRAC= 0.22500

*** RESULTS ***

STREAM= 20 SPLIT= 0.22500 KEY= 0
 22 0.77500 0

BLOCK: B18 MODEL: PUMP

 INLET STREAM: 20
 OUTLET STREAM: 21
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF
TOTAL BALANCE			
MOLE (KMOL/HR)	3.65733	3.65733	0.
MASS (KG/SEC)	0.202767E-01	0.202767E-01	0.171105E-15
ENTHALPY (WATT)	-61794.3	-61775.6	-0.302297E-03

U-O-S BLOCK SECTION

BLOCK: B18 MODEL: PUMP (CONTINUED)

*** INPUT DATA ***

OUTLET PRESSURE (BAR) 32.5000
 DRIVER EFFICIENCY 1.00000

FLASH SPECIFICATIONS:

LIQUID PHASE CALCULATION

NO FLASH PERFORMED

MAXIMUM NUMBER OF ITERATIONS 30

TOLERANCE 0.000100000

*** RESULTS ***

VOLUMETRIC FLOW RATE (CUM/SEC) 0.552301-04
 PRESSURE CHANGE (BAR) 1.00000
 FLUID POWER (WATT) 5.52301
 BRAKE POWER (WATT) 18.6804
 ELECTRICITY (WATT) 18.6804
 PUMP EFFICIENCY USED 0.29566
 NET WORK (WATT) -18.6804

BLOCK: B2 MODEL: HEATER

 INLET STREAM: 2
 OUTLET STREAM: 3
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	53.1084	53.1084	0.
MASS (KG/SEC)	0.274843	0.274843	0.
ENTHALPY (WATT)	-959938.	-0.109029E+07	0.119558

*** INPUT DATA ***

TWO PHASE TP FLASH
 SPECIFIED TEMPERATURE C -80.0000
 SPECIFIED PRESSURE BAR 47.5000
 MAXIMUM NO. ITERATIONS 30
 CONVERGENCE TOLERANCE 0.000100000

U-O-S BLOCK SECTION

BLOCK: B2 MODEL: HEATER (CONTINUED)

*** RESULTS ***

OUTLET TEMPERATURE	C	-80.000
OUTLET PRESSURE	BAR	47.500
HEAT DUTY	WATT	-0.13035E+06
VAPOR FRACTION		0.90064

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.81305	0.76520	0.81833	1.0694
C2H6	0.28492E-01	0.10314	0.20257E-01	0.19639
C3H8	0.37152E-02	0.24988E-01	0.13683E-02	0.54754E-01
C4H10	0.13931E-02	0.12269E-01	0.19315E-03	0.15741E-01
C5H12	0.39148E-03	0.37888E-02	0.16673E-04	0.44000E-01
C6H14	0.50257E-03	0.50001E-02	0.63826E-05	0.12763E-01
N2	0.14344	0.57286E-01	0.15294	2.6700
O2	0.11769E-03	0.69774E-04	0.12297E-03	1.7625
CO2	0.88991E-02	0.28251E-01	0.67641E-02	0.23942

BLOCK: B20 MODEL: HEATER

INLET STREAM:	22
OUTLET STREAM:	23
PROPERTY OPTION SET:	SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	12.5975	12.5975	0.
MASS (KG/SEC)	0.698421E-01	0.698421E-01	0.198702E-15
ENTHALPY (WATT)	-212847.	-212847.	0.133706E-05

*** INPUT DATA ***

TWO PHASE PQ FLASH	
SPECIFIED PRESSURE	BAR 28.5000
SPECIFIED HEAT DUTY	WATT 0.0
MAXIMUM NO. ITERATIONS	30
CONVERGENCE TOLERANCE	0.000100000

U-O-S BLOCK SECTION

BLOCK: B20 MODEL: HEATER (CONTINUED)

*** RESULTS ***

OUTLET TEMPERATURE	C	-123.60
OUTLET PRESSURE	BAR	28.500
VAPOR FRACTION		0.59809E-01

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.67294	0.69115	0.38658	0.55931
C2H6	0.28447E-07	0.30183E-07	0.11650E-08	0.38600E-01
N2	0.32683	0.30861	0.61313	1.9867
O2	0.23561E-03	0.23241E-03	0.28591E-03	1.2302
CO2	0.12047E-06	0.12768E-06	0.72647E-08	0.56900E-01

BLOCK: B21 MODEL: MIXER

INLET STREAMS:	23	27
OUTLET STREAM:	24	
PROPERTY OPTION SET:	SYSOP3	REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	14.5635	14.5635	0.
MASS (KG/SEC)	0.805673E-01	0.805673E-01	-0.689004E-15
ENTHALPY (WATT)	-247293.	-247293.	0.147643E-05

*** INPUT DATA ***

TWO PHASE FLASH	
MAXIMUM NO. ITERATIONS	30
CONVERGENCE TOLERANCE	0.000100000
OUTLET PRESSURE BAR	28.5000

BLOCK: B22 MODEL: HEATER

INLET STREAM:	24
OUTLET STREAM:	25
PROPERTY OPTION SET:	SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

U-O-S BLOCK SECTION

BLOCK: B22 MODEL: HEATER (CONTINUED)

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF
MOLE (KMOL/HR)	14.5635	14.5635	0.
MASS (KG/SEC)	0.805673E-01	0.805673E-01	0.
ENTHALPY (WATT)	-247293.	-258390.	0.429436E-01

*** INPUT DATA ***

TWO PHASE TP FLASH	C	
SPECIFIED TEMPERATURE	BAR	-161.150
SPECIFIED PRESSURE		28.0000
MAXIMUM NO. ITERATIONS		30
CONVERGENCE TOLERANCE		0.000100000

*** RESULTS ***

OUTLET TEMPERATURE	C	-161.15
OUTLET PRESSURE	BAR	28.000
HEAT DUTY	WATT	-11096.
VAPOR FRACTION		0.

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.67655	0.67655	0.19999	0.90044E-
C2H6	0.24981E-07	0.24981E-07	0.81017E-10	0.95570E-
N2	0.32321	0.32321	0.79978	0.76429
O2	0.24590E-03	0.24590E-03	0.23810E-03	0.29802
CO2	0.10649E-06	0.10649E-06	0.38900E-09	0.10843E-

BLOCK: B23 MODEL: HEATER

 INLET STREAM: 25
 OUTLET STREAM: 26
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	14.5635	14.5635	0.
MASS (KG/SEC)	0.805673E-01	0.805673E-01	0.
ENTHALPY (WATT)	-258390.	-258389.	-0.246774E-05

U-O-S BLOCK SECTION

BLOCK: B23 MODEL: HEATER (CONTINUED)

*** INPUT DATA ***

TWO PHASE PQ FLASH		
SPECIFIED PRESSURE	BAR	2.00000
SPECIFIED HEAT DUTY	WATT	0.0
MAXIMUM NO. ITERATIONS		30
CONVERGENCE TOLERANCE		0.000100000

*** RESULTS ***

OUTLET TEMPERATURE	C	-174.92
OUTLET PRESSURE	BAR	2.0000
VAPOR FRACTION		0.15871

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.67655	0.78236	0.11564	0.14780
C2H6	0.24981E-07	0.29693E-07	0.27838E-11	0.93738E-04
N2	0.32321	0.21739	0.88410	4.0671
O2	0.24590E-03	0.24322E-03	0.26014E-03	1.0696
CO2	0.10649E-06	0.12658E-06	0.17458E-10	0.13791E-03

BLOCK: B24 MODEL: PUMP

INLET STREAM:	29	
OUTLET STREAM:	30	
PROPERTY OPTION SET:	SYSOP3	REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	1.75121	1.75121	0.
MASS (KG/SEC)	0.133362E-01	0.133362E-01	0.130076E-15
ENTHALPY (WATT)	-6173.52	-6168.38	-0.832438E-03

*** INPUT DATA ***

OUTLET PRESSURE (BAR)	28.5000
DRIVER EFFICIENCY	1.00000

FLASH SPECIFICATIONS:

LIQUID PHASE CALCULATION

NO FLASH PERFORMED

MAXIMUM NUMBER OF ITERATIONS

30

TOLERANCE

0.000100000

ASPEN PLUS VER: SUN-4 REL: 8.5-4 INST: DELFTUOT 05/02/93 PAGE

U-O-S BLOCK SECTION

BLOCK: B24 MODEL: PUMP (CONTINUED)

*** RESULTS ***

VOLUMETRIC FLOW RATE (CUM/SEC)	0.303778-04
PRESSURE CHANGE (BAR)	0.50000
FLUID POWER (WATT)	1.51889
BRAKE POWER (WATT)	5.13732
ELECTRICITY (WATT)	5.13732
PUMP EFFICIENCY USED	0.29566
NET WORK (WATT)	-5.13732

BLOCK: B3 MODEL: FLASH2

 INLET STREAM: 3
 OUTLET VAPOR STREAM: 4
 OUTLET LIQUID STREAM: 17
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF
MOLE (KMOL/HR)	53.1084	53.1084	0.
MASS (KG/SEC)	0.274843	0.274843	0.163440E-08
ENTHALPY (WATT)	-0.109029E+07	-0.109029E+07	0.

*** INPUT DATA ***

TWO PHASE TP FLASH	
SPECIFIED TEMPERATURE C	-80.0000
SPECIFIED PRESSURE BAR	47.5000
MAXIMUM NO. ITERATIONS	30
CONVERGENCE TOLERANCE	0.000100000

ASPEN PLUS VER: SUN-4 REL: 8.5-4 INST: DELFTUOT 05/02/93 PAGE 23

U-O-S BLOCK SECTION

BLOCK: B3 MODEL: FLASH2 (CONTINUED)

*** RESULTS ***

OUTLET TEMPERATURE	C	-80.000
OUTLET PRESSURE	BAR	47.500
HEAT DUTY	WATT	0.
VAPOR FRACTION		0.90064

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.81305	0.76520	0.81833	1.0694
C2H6	0.28492E-01	0.10314	0.20257E-01	0.19639
C3H8	0.37152E-02	0.24988E-01	0.13683E-02	0.54754E-01
C4H10	0.13931E-02	0.12269E-01	0.19315E-03	0.15741E-01
C5H12	0.39148E-03	0.37888E-02	0.16673E-04	0.44000E-02
C6H14	0.50257E-03	0.50001E-02	0.63826E-05	0.12763E-02
N2	0.14344	0.57286E-01	0.15294	2.6700
O2	0.11769E-03	0.69774E-04	0.12297E-03	1.7625
CO2	0.88991E-02	0.28251E-01	0.67641E-02	0.23942

BLOCK: B4 MODEL: HEATER

 INLET STREAM: 4
 OUTLET STREAM: 5
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	47.8314	47.8314	0.
MASS (KG/SEC)	0.244431	0.244431	-0.113552E-15
ENTHALPY (WATT)	-952223.	-995184.	0.431686E-01

*** INPUT DATA ***

TWO PHASE PV FLASH		
SPECIFIED PRESSURE	BAR	47.0000
VAPOR FRACTION		0.0
MAXIMUM NO. ITERATIONS		500
CONVERGENCE TOLERANCE		0.000100000

ASPEN PLUS VER: SUN-4 REL: 8.5-4 INST: DELFTUOT 05/02/93 PAGE 24

U-O-S BLOCK SECTION

BLOCK: B4 MODEL: HEATER (CONTINUED)

*** RESULTS ***

OUTLET TEMPERATURE	C	-92.387
OUTLET PRESSURE	BAR	47.000
HEAT DUTY	WATT	-42961.
VAPOR FRACTION		0.

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.81833	0.81833	0.74902	0.91540
C2H6	0.20257E-01	0.20257E-01	0.62748E-02	0.30979
C3H8	0.13683E-02	0.13683E-02	0.18186E-03	0.13292
C4H10	0.19315E-03	0.19315E-03	0.11222E-04	0.58104E-
C5H12	0.16673E-04	0.16673E-04	0.41165E-06	0.24692E-
C6H14	0.63826E-05	0.63826E-05	0.67510E-07	0.10579E-
N2	0.15294	0.15294	0.24184	1.5814
O2	0.12297E-03	0.12297E-03	0.15644E-03	1.2722
CO2	0.67641E-02	0.67641E-02	0.25090E-02	0.37095

BLOCK: B5 MODEL: HEATER

INLET STREAM: 5
 OUTLET STREAM: 6
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	47.8314	47.8314	0.
MASS (KG/SEC)	0.244431	0.244431	0.
ENTHALPY (WATT)	-995184.	-995184.	0.105697E-06

*** INPUT DATA ***

TWO PHASE PQ FLASH
 SPECIFIED PRESSURE BAR 32.9000
 SPECIFIED HEAT DUTY WATT 0.0
 MAXIMUM NO. ITERATIONS 30
 CONVERGENCE TOLERANCE 0.000100000

ASPEN PLUS VER: SUN-4 REL: 8.5-4 INST: DELFTUOT 05/02/93 PAGE 25

U-O-S BLOCK SECTION

BLOCK: B5 MODEL: HEATER (CONTINUED)

*** RESULTS ***

OUTLET TEMPERATURE	C	-102.07
OUTLET PRESSURE	BAR	32.900
VAPOR FRACTION		0.31399

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.81833	0.85540	0.73732	0.86198
C2H6	0.20257E-01	0.28182E-01	0.29414E-02	0.10437
C3H8	0.13683E-02	0.19763E-02	0.39947E-04	0.20212E-01
C4H10	0.19315E-03	0.28104E-03	0.11432E-05	0.40676E-02
C5H12	0.16673E-04	0.24295E-04	0.18769E-07	0.77248E-03
C6H14	0.63826E-05	0.93033E-05	0.13947E-08	0.14991E-03
N2	0.15294	0.10477	0.25820	2.4646
O2	0.12297E-03	0.10299E-03	0.16663E-03	1.6179
CO2	0.67641E-02	0.92520E-02	0.13283E-02	0.14357

BLOCK: B6 MODEL: RADFRAC

INLETS	- 6	STAGE	8
	21	STAGE	2
OUTLETS	- 7	STAGE	1
	18	STAGE	10

PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	51.4887	51.4892	-0.873958E-05
MASS (KG/SEC)	0.264707	0.264710	-0.977278E-05
ENTHALPY (WATT)	-0.105696E+07	-0.104888E+07	-0.764883E-02

ASPEN PLUS VER: SUN-4 REL: 8.5-4 INST: DELFTUOT 05/02/93 PAGE

U-O-S BLOCK SECTION

BLOCK: B6 MODEL: RADFRAC (CONTINUED)

**** INPUT DATA ****

**** INPUT PARAMETERS ****

NUMBER OF STAGES	10
ALGORITHM OPTION	STANDARD
ABSORBER OPTION	NO
INITIALIZATION OPTION	STANDARD
HYDRAULIC PARAMETER CALCULATIONS	NO
INSIDE LOOP CONVERGENCE METHOD	BROYDEN
DESIGN SPECIFICATION METHOD	NESTED
MAXIMUM NO. OF OUTSIDE LOOP ITERATIONS	25
MAXIMUM NO. OF INSIDE LOOP ITERATIONS	10
MAXIMUM NUMBER OF FLASH ITERATIONS	50
FLASH TOLERANCE	0.000100000
OUTSIDE LOOP CONVERGENCE TOLERANCE	0.000100000

**** COL-SPECS ****

MOLAR VAPOR DIST / TOTAL DIST	1.00000
MOLAR REFLUX RATIO	1.02800
REBOILER DUTY	WATT
	35,463.9

**** PROFILES ****

P-SPEC	STAGE	1	PRES, BAR	32.0000
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U-O-S BLOCK SECTION

BLOCK: B6 MODEL: RADFRAC (CONTINUED)

 **** RESULTS ****

TOP STAGE TEMPERATURE	C	-109.926
BOTTOM STAGE TEMPERATURE	C	-94.2057
TOP STAGE LIQUID FLOW	KMOL/HR	20.5308
BOTTOM STAGE LIQUID FLOW	KMOL/HR	31.5172
TOP STAGE VAPOR FLOW	KMOL/HR	19.9716
BOTTOM STAGE VAPOR FLOW	KMOL/HR	27.8142
MOLAR REFLUX RATIO		1.02800
MOLAR BOILUP RATIO		0.88251
CONDENSER DUTY (W/O SUBCOOL)	WATT	-25,351.0
REBOILER DUTY	WATT	33,441.8

**** MANIPULATED VARIABLES ****

REBOILER DUTY	WATT	33,441.8
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**** DESIGN SPECIFICATIONS ****

NO	SPEC-TYPE	UNIT	SPECIFIED VALUE	CALCULATED VALUE	ABS. ERROR
1	MOLE-FRAC		0.30000E-01	0.30000E-01	-0.31455E-08

**** MAXIMUM FINAL RELATIVE ERRORS ****

DEW POINT	0.20609E-06	STAGE=	1
BUBBLE POINT	0.12126E-05	STAGE=	1
COMPONENT MASS BALANCE	0.31919E-07	STAGE=	2 COMP=N2
ENERGY BALANCE	0.10821E-05	STAGE=	2

**** PROFILES ****

STAGE	TEMPERATURE C	PRESSURE BAR	ENTHALPY J/KMOL		HEAT DUTY WATT
			LIQUID	VAPOR	
1	-109.93	32.000	-0.72023E+08	-0.52372E+08	-.25351+05
2	-105.35	32.111	-0.75863E+08	-0.60080E+08	
3	-103.28	32.222	-0.77461E+08	-0.63619E+08	
5	-102.30	32.444	-0.78090E+08	-0.65069E+08	
6	-102.10	32.556	-0.78263E+08	-0.65195E+08	
7	-101.76	32.667	-0.78848E+08	-0.65268E+08	
8	-99.623	32.778	-0.81729E+08	-0.67412E+08	
9	-97.224	32.889	-0.83475E+08	-0.71552E+08	
10	-94.206	33.000	-0.86619E+08	-0.75584E+08	.33442+05

U-O-S BLOCK SECTION

BLOCK: B6 MODEL: RADFRAC (CONTINUED)

STAGE	FLOW RATE KMOL/HR		FEED RATE KMOL/HR			PRODUCT RATE KMOL/HR	
	LIQUID	VAPOR	LIQUID	VAPOR	MIXED	LIQUID	VAPOR
1	20.53	19.97					
2	24.96	40.50	3.6573				19.9720
3	25.40	41.28					
5	25.69	41.92					
6	25.59	42.01					
7	24.67	41.90		15.0897			
8	58.49	25.90	32.7416				
9	59.33	26.98					
10	31.52	27.81				31.5171	

STAGE	CH4	**** X-PROFILE ****			C5H12	C6H14
		C2H6	C3H8	C4H10		
1	.82420	.27081E-06	.38504E-12	.13900E-17	.20492E-23	.14844E-2
2	.87698	.13878E-05	.91424E-11	.15543E-15	.11576E-20	.41825E-2
3	.89916	.71541E-05	.21996E-09	.17006E-13	.61049E-18	.10486E-2
5	.90720	.17364E-03	.11280E-06	.17233E-09	.13683E-12	.50618E-1
6	.90674	.84370E-03	.24984E-05	.16814E-07	.62182E-10	.10573E-1
7	.90299	.41196E-02	.55979E-04	.16700E-05	.28943E-07	.22742E-0
8	.89971	.17754E-01	.11354E-02	.15848E-03	.13644E-04	.52201E-0
9	.92159	.18472E-01	.11362E-02	.15682E-03	.13463E-04	.51474E-0
10	.92653	.30742E-01	.20766E-02	.29314E-03	.25303E-04	.96864E-0

STAGE	N2	**** X-PROFILE ****		
		O2	CO2	
1	.17562	.16993E-03	.82063E-06	
2	.12288	.13562E-03	.31218E-05	
3	.10071	.11527E-03	.11760E-04	
5	.92374E-01	.10288E-03	.15126E-03	
6	.91777E-01	.10134E-03	.53509E-03	
7	.90832E-01	.10023E-03	.19031E-02	
8	.75040E-01	.93591E-04	.60932E-02	
9	.52132E-01	.79906E-04	.64194E-02	
10	.30000E-01	.59104E-04	.10265E-01	

STAGE	CH4	**** Y-PROFILE ****			C5H12	C6H14
		C2H6	C3H8	C4H10		
1	.62095	.23424E-07	.61143E-14	.42169E-20	.11118E-26	.14646E-3
2	.72398	.14883E-06	.19820E-12	.70666E-18	.10393E-23	.75250E-2
3	.77118	.84809E-06	.55319E-11	.93996E-16	.70008E-21	.25294E-2
5	.79027	.21695E-04	.30772E-08	.10625E-11	.18027E-15	.14439E-1
6	.79150	.10621E-03	.68990E-07	.10540E-09	.83689E-13	.30959E-1
7	.79093	.51523E-03	.15257E-05	.10268E-07	.37972E-10	.64566E-1
8	.81452	.22217E-02	.30350E-04	.93735E-06	.16912E-07	.13792E-0
9	.86837	.25800E-02	.35784E-04	.11518E-05	.21901E-07	.18842E-0
10	.91598	.45688E-02	.70627E-04	.23586E-05	.45867E-07	.40250E-0

U-O-S BLOCK SECTION

BLOCK: B6 MODEL: RADFRAC (CONTINUED)

STAGE	N2	**** Y-PROFILE ****	
		O2	CO2
1	.37881	.24439E-03	.99703E-07
2	.27581	.20665E-03	.46514E-06
3	.22864	.17939E-03	.19255E-05
5	.20952	.16094E-03	.25996E-04
6	.20814	.15860E-03	.92555E-04
7	.20807	.15780E-03	.32680E-03
8	.18203	.15356E-03	.10434E-02
9	.12766	.13388E-03	.12189E-02
10	.77210E-01	.10348E-03	.20614E-02

STAGE	CH4	**** K-VALUES ****				
		C2H6	C3H8	C4H10	C5H12	C6H14
1	.75339	.86494E-01	.15879E-01	.30337E-02	.54254E-03	.98667E-04
2	.82554	.10724	.21679E-01	.45466E-02	.89779E-03	.17992E-03
3	.85767	.11855	.25149E-01	.55271E-02	.11468E-02	.24122E-03
5	.87111	.12494	.27280E-01	.61656E-02	.13175E-02	.28526E-03
6	.87291	.12589	.27614E-01	.62687E-02	.13459E-02	.29281E-03
7	.87591	.12507	.27255E-01	.61484E-02	.13120E-02	.28391E-03
8	.90531	.12514	.26731E-01	.59147E-02	.12395E-02	.26422E-03
9	.94226	.13967	.31494E-01	.73443E-02	.16268E-02	.36606E-03
10	.98862	.14862	.34011E-01	.80462E-02	.18127E-02	.41553E-03

STAGE	N2	**** K-VALUES ****	
		O2	CO2
1	2.1569	1.4382	.12149
2	2.2445	1.5237	.14900
3	2.2703	1.5563	.16373
5	2.2681	1.5644	.17186
6	2.2679	1.5650	.17297
7	2.2907	1.5743	.17171
8	2.4258	1.6408	.17124
9	2.4488	1.6755	.18988
10	2.5737	1.7508	.20081

BLOCK: B7 MODEL: HEATER

 INLET STREAM: 7
 OUTLET STREAM: 8
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	19.9720	19.9720	0.
MASS (KG/SEC)	0.114180	0.114180	0.
ENTHALPY (WATT)	-290546.	-311231.	0.664621E-01

U-O-S BLOCK SECTION

BLOCK: B7 MODEL: HEATER (CONTINUED)

*** INPUT DATA ***

TWO PHASE TP FLASH		
SPECIFIED TEMPERATURE	C	-121.407
SPECIFIED PRESSURE	BAR	31.5000
MAXIMUM NO. ITERATIONS		30
CONVERGENCE TOLERANCE		0.000100000

*** RESULTS ***

OUTLET TEMPERATURE	C	-121.41
OUTLET PRESSURE	BAR	31.500
HEAT DUTY	WATT	-20685.
VAPOR FRACTION		0.18612

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.62095	0.67294	0.39362	0.58492
C2H6	0.23426E-07	0.28447E-07	0.14672E-08	0.51575E-
C3H8	0.13582E-13	0.16659E-13	0.12802E-15	0.76848E-
N2	0.37881	0.32683	0.60610	1.8545
O2	0.24439E-03	0.23561E-03	0.28280E-03	1.2003
CO2	0.99715E-07	0.12047E-06	0.89327E-08	0.74146E-

BLOCK: B8 MODEL: FLASH2

INLET STREAM: 8
 OUTLET VAPOR STREAM: 9
 OUTLET LIQUID STREAM: 19
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	19.9720	19.9720	0.135192E-13
MASS (KG/SEC)	0.114180	0.114180	0.291703E-13
ENTHALPY (WATT)	-311231.	-311231.	-0.147749E-13

*** INPUT DATA ***

TWO PHASE TP FLASH		
SPECIFIED TEMPERATURE	C	-121.407
SPECIFIED PRESSURE	BAR	31.5000
MAXIMUM NO. ITERATIONS		30
CONVERGENCE TOLERANCE		0.000100000

ASPEN PLUS VER: SUN-4 REL: 8.5-4 INST: DELFTUOT 05/02/93 PAGE 31

U-O-S BLOCK SECTION

BLOCK: B8 MODEL: FLASH2 (CONTINUED)

*** RESULTS ***

OUTLET TEMPERATURE	C	-121.41
OUTLET PRESSURE	BAR	31.500
HEAT DUTY	WATT	0.
VAPOR FRACTION		0.18612

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.62095	0.67294	0.39362	0.58492
C2H6	0.23426E-07	0.28447E-07	0.14672E-08	0.51575E-01
C3H8	0.13582E-13	0.16659E-13	0.12802E-15	0.76848E-02
N2	0.37881	0.32683	0.60610	1.8545
O2	0.24439E-03	0.23561E-03	0.28280E-03	1.2003
CO2	0.99715E-07	0.12047E-06	0.89327E-08	0.74146E-01

BLOCK: B9 MODEL: HEATER

INLET STREAM: 9
 OUTLET STREAM: 10
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

TOTAL BALANCE	IN	OUT	RELATIVE DIFF.
MOLE (KMOL/HR)	3.71722	3.71722	0.
MASS (KG/SEC)	0.240613E-01	0.240613E-01	0.
ENTHALPY (WATT)	-36590.2	-36590.2	-0.135820E-06

*** INPUT DATA ***

TWO PHASE PQ FLASH

SPECIFIED PRESSURE	BAR	28.9000
SPECIFIED HEAT DUTY	WATT	0.0
MAXIMUM NO. ITERATIONS		30
CONVERGENCE TOLERANCE		0.000100000

ASPEN PLUS VER: SUN-4 REL: 8.5-4 INST: DELFTUOT 05/02/93 PAGE 32

U-O-S BLOCK SECTION

BLOCK: B9 MODEL: HEATER (CONTINUED)

*** RESULTS ***
OUTLET TEMPERATURE C -123.24
OUTLET PRESSURE BAR 28.900
VAPOR FRACTION 0.98304

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CH4	0.39362	0.68957	0.38851	0.56340
C2H6	0.14672E-08	0.25969E-07	0.10444E-08	0.40219E-01
N2	0.60610	0.31020	0.61121	1.9704
O2	0.28280E-03	0.23112E-03	0.28369E-03	1.2274
CO2	0.89327E-08	0.11902E-06	0.70333E-08	0.59093E-01

STREAM SECTION

1 10 11 13 14

STREAM ID	1	10	11	13	14
FROM :	----	B9	B10	B12	B13
TO :	B1	B10	B12	B13	B14
SUBSTREAM: MIXED					
PHASE:	VAPOR	MIXED	LIQUID	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
CH4	43.1797	1.4631	0.1751	8.7561-02	8.7561-02
C2H6	1.5131	5.4537-09	1.9735-13	9.8674-14	9.8674-14
C3H8	0.1973	0.0	0.0	0.0	0.0
C4H10	7.3985-02	0.0	0.0	0.0	0.0
C5H12	2.0791-02	0.0	0.0	0.0	0.0
C6H14	2.6691-02	0.0	0.0	0.0	0.0
N2	7.6178	2.2530	3.3264	1.6632	1.6632
O2	6.2502-03	1.0512-03	8.7617-04	4.3809-04	4.3809-04
CO2	0.4726	3.3205-08	4.0795-12	2.0398-12	2.0398-12
TOTAL FLOW:					
KMOL/HR	53.1083	3.7172	3.5024	1.7512	1.7512
KG/SEC	0.2748	2.4061-02	2.6672-02	1.3336-02	1.3336-02
CUM/SEC	1.7079-02	2.9622-04	6.0756-05	3.0378-05	2.9597-05
STATE VARIABLES:					
TEMP C	15.0000	-123.2436	-149.4342	-149.4342	-150.2351
PRES BAR	20.0000	28.9000	28.0000	28.0000	27.0000
VFRAC	1.0000	0.9830	0.0	0.0	0.0
LFRAC	0.0	1.6961-02	1.0000	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.8228+07	-3.5436+07	-1.2691+07	-1.2691+07	-1.2789+07
J/KG	-3.6622+06	-1.5207+06	-4.6291+05	-4.6291+05	-4.6650+05
WATT	-1.0065+06	-3.6590+04	-1.2347+04	-6173.5184	-6221.2836
ENTROPY:					
J/KMOL-K	-9.4159+04	-8.1977+04	-8.3488+04	-8.3488+04	-8.4234+04
J/KG-K	-5054.0421	-3517.9462	-3045.2735	-3045.2745	-3072.4847
DENSITY:					
KMOL/CUM	0.8637	3.4857	16.0132	16.0132	16.4359
KG/CUM	16.0920	81.2270	439.0113	439.0116	450.5983
AVG MW	18.6304	23.3025	27.4154	27.4154	27.4154

STREAM SECTION

15 16 17 18 19

STREAM ID	15	16	17	18	19
FROM :	B14	B15	B3	B6	B8
TO :	B15	----	B16	B16	B17
SUBSTREAM: MIXED					
PHASE:	MIXED	VAPOR	LIQUID	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
CH4	8.7561-02	2.9604-02	4.0379	29.2015	10.938
C2H6	9.8674-14	1.2303-16	0.5442	0.9689	4.6240-0
C3H8	0.0	0.0	0.1318	6.5447-02	0.0
C4H10	0.0	0.0	6.4746-02	9.2388-03	0.0
C5H12	0.0	0.0	1.9993-02	7.9749-04	0.0
C6H14	0.0	0.0	2.6386-02	3.0529-04	0.0
N2	1.6632	6.0637	0.3023	0.9455	5.312
O2	4.3809-04	8.6794-04	3.6819-04	1.8628-03	3.8297-0
CO2	2.0398-12	1.6600-15	0.1490	0.3235	1.9583-0
TOTAL FLOW:					
KMOL/HR	1.7512	6.0942	5.2769	31.5171	16.2548
KG/SEC	1.3336-02	4.7324-02	3.0412-02	0.1505	9.0119-02
CUM/SEC	7.4602-04	7.3438-03	8.7516-05	5.3491-04	2.4547-04
STATE VARIABLES:					
TEMP C	-186.6798	-190.7165	-80.0000	-94.2056	-121.4070
PRES BAR	2.3300	1.5000	47.5000	33.0000	31.5000
VFRAC	0.5288	1.0000	0.0	0.0	0.0
LFRAC	0.4711	0.0	1.0000	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.2789+07	-6.7438+06	-9.4191+07	-8.6619+07	-6.0826+07
J/KG	-4.6650+05	-2.4123+05	-4.5399+06	-5.0377+06	-3.0475+06
WATT	-6221.2815	-1.1416+04	-1.3807+05	-7.5833+05	-2.7464+05
ENTROPY:					
J/KMOL-K	-7.7101+04	-4.1539+04	-1.6793+05	-1.5548+05	-1.3217+05
J/KG-K	-2812.3125	-1485.9117	-8093.9812	-9042.4387	-6622.1546
DENSITY:					
KMOL/CUM	0.6520	0.2305	16.7492	16.3667	18.3944
KG/CUM	17.8765	6.4441	347.5005	281.4105	367.1320
AVG MW	27.4154	27.9554	20.7472	17.1940	19.9588

STREAM SECTION

2 20 21 22 23

STREAM ID	2	20	21	22	23
FROM :	B1	B17	B18	B17	B20
TO :	B2	B18	B6	B20	B21
SUBSTREAM: MIXED					
PHASE:	VAPOR	LIQUID	LIQUID	LIQUID	MIXED
COMPONENTS: KMOL/HR					
CH4	43.1797	2.4611	2.4611	8.4773	8.4773
C2H6	1.5131	1.0404-07	1.0404-07	3.5836-07	3.5836-07
C3H8	0.1973	0.0	0.0	0.0	0.0
C4H10	7.3985-02	0.0	0.0	0.0	0.0
C5H12	2.0791-02	0.0	0.0	0.0	0.0
C6H14	2.6691-02	0.0	0.0	0.0	0.0
N2	7.6178	1.1953	1.1953	4.1172	4.1172
O2	6.2502-03	8.6169-04	8.6169-04	2.9680-03	2.9680-03
CO2	0.4726	4.4062-07	4.4062-07	1.5177-06	1.5177-06
TOTAL FLOW:					
KMOL/HR	53.1083	3.6573	3.6573	12.5974	12.5974
KG/SEC	0.2748	2.0277-02	2.0277-02	6.9842-02	6.9842-02
CUM/SEC	9.4941-03	5.5230-05	5.5295-05	1.9024-04	2.3405-04
STATE VARIABLES:					
TEMP C	103.1160	-121.4070	-121.1247	-121.4070	-123.6012
PRES BAR	48.0000	31.5000	32.5000	31.5000	28.5000
VFRAC	1.0000	0.0	0.0	0.0	5.9809-02
LFRAC	0.0	1.0000	1.0000	1.0000	0.9401
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.5070+07	-6.0826+07	-6.0807+07	-6.0826+07	-6.0826+07
J/KG	-3.4927+06	-3.0475+06	-3.0466+06	-3.0475+06	-3.0475+06
WATT	-9.5994+05	-6.1794+04	-6.1776+04	-2.1285+05	-2.1285+05
ENTROPY:					
J/KMOL-K	-9.1728+04	-1.3217+05	-1.3209+05	-1.3217+05	-1.3205+05
J/KG-K	-4923.5423	-6622.1546	-6617.8835	-6622.1546	-6616.1550
DENSITY:					
KMOL/CUM	1.5538	18.3944	18.3727	18.3944	14.9508
KG/CUM	28.9486	367.1320	366.6996	367.1320	298.4019
AVG MW	18.6304	19.9588	19.9588	19.9588	19.9588

STREAM SECTION

24 25 26 27 28

STREAM ID	24	25	26	27	28
FROM :	B21	B22	B23	B10	B15
TO :	B22	B23	B15	B21	B16
SUBSTREAM: MIXED					
PHASE:	MIXED	LIQUID	MIXED	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
CH4	9.8529	9.8529	9.8529	1.3756	9.9108
C2H6	3.6382-07	3.6382-07	3.6382-07	5.4536-09	3.6382-07
C3H8	0.0	0.0	0.0	0.0	0.0
C4H10	0.0	0.0	0.0	0.0	0.0
C5H12	0.0	0.0	0.0	0.0	0.0
C6H14	0.0	0.0	0.0	0.0	0.0
N2	4.7070	4.7070	4.7070	0.5898	0.3064
O2	3.5812-03	3.5812-03	3.5812-03	6.1317-04	3.1514-03
CO2	1.5509-06	1.5509-06	1.5509-06	3.3202-08	1.5509-06
TOTAL FLOW:					
KMOL/HR	14.5635	14.5635	14.5635	1.9660	10.2204
KG/SEC	8.0567-02	8.0567-02	8.0567-02	1.0725-02	4.6579-02
CUM/SEC	2.6405-04	1.5676-04	2.6239-03	2.8824-05	1.1170-04
STATE VARIABLES:					
TEMP C	-123.5024	-161.1500	-174.9204	-122.4728	-154.1806
PRES BAR	28.5000	28.0000	2.0000	29.0000	2.5000
VFRAC	5.3068-02	0.0	0.1587	0.0	0.0
LFRAC	0.9469	1.0000	0.8412	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.1129+07	-6.3872+07	-6.3872+07	-6.3074+07	-8.6829+07
J/KG	-3.0694+06	-3.2071+06	-3.2071+06	-3.2117+06	-5.2922+06
WATT	-2.4729+05	-2.5839+05	-2.5839+05	-3.4446+04	-2.4651+05
ENTROPY:					
J/KMOL-K	-1.3253+05	-1.5330+05	-1.5159+05	-1.3565+05	-1.8135+05
J/KG-K	-6654.7664	-7697.5162	-7611.7420	-6907.0263	-1.1053+04
DENSITY:					
KMOL/CUM	15.3204	25.8057	1.5417	18.9467	25.4167
KG/CUM	305.1171	513.9386	30.7056	372.0937	417.0070
AVG MW	19.9157	19.9157	19.9157	19.6389	16.4068

STREAM SECTION

29 3 30 31 4

STREAM ID	29	3	30	31	4
FROM :	B12	B2	B24	B16	B3
TO :	B24	B3	B10	----	B4
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	LIQUID	MIXED	VAPOR
COMPONENTS: KMOL/HR					
CH4	8.7561-02	43.1797	8.7561-02	43.1503	39.1417
C2H6	9.8674-14	1.5131	9.8674-14	1.5131	0.9689
C3H8	0.0	0.1973	0.0	0.1973	6.5447-02
C4H10	0.0	7.3985-02	0.0	7.3984-02	9.2388-03
C5H12	0.0	2.0791-02	0.0	2.0791-02	7.9749-04
C6H14	0.0	2.6691-02	0.0	2.6691-02	3.0529-04
N2	1.6632	7.6178	1.6632	1.5542	7.3155
O2	4.3809-04	6.2502-03	4.3809-04	5.3823-03	5.8820-03
CO2	2.0398-12	0.4726	2.0398-12	0.4726	0.3235
TOTAL FLOW:					
KMOL/HR	1.7512	53.1083	1.7512	47.0145	47.8314
KG/SEC	1.3336-02	0.2748	1.3336-02	0.2275	0.2444
CUM/SEC	3.0378-05	2.6732-03	3.0411-05	2.0317-03	2.5857-03
STATE VARIABLES:					
TEMP C	-149.4342	-80.0000	-149.2685	-115.8763	-80.0000
PRES BAR	28.0000	47.5000	28.5000	15.0000	47.5000
VFRAC	0.0	0.9006	0.0	0.1677	1.0000
LFRAC	1.0000	9.9362-02	1.0000	0.8323	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-1.2691+07	-7.3906+07	-1.2680+07	-8.7514+07	-7.1668+07
J/KG	-4.6291+05	-3.9670+06	-4.6253+05	-5.0233+06	-3.8957+06
WATT	-6173.5184	-1.0903+06	-6168.3793	-1.1429+06	-9.5222+05
ENTROPY:					
J/KMOL-K	-8.3488+04	-1.2531+05	-8.3427+04	-1.6065+05	-1.2060+05
J/KG-K	-3045.2745	-6725.8631	-3043.0718	-9221.0652	-6555.6428
DENSITY:					
KMOL/CUM	16.0132	5.5185	15.9955	6.4280	5.1384
KG/CUM	439.0116	102.8126	438.5270	111.9870	94.5309
AVG MW	27.4154	18.6304	27.4154	17.4217	18.3969

STREAM SECTION

5 6 7 8 9

STREAM ID	5	6	7	8	9
FROM :	B4	B5	B6	B7	B8
TO :	B5	B6	B7	B8	B9
SUBSTREAM: MIXED					
PHASE:	LIQUID	MIXED	VAPOR	MIXED	VAPOR
COMPONENTS: KMOL/HR					
CH4	39.1417	39.1417	12.4016	12.4016	1.4631
C2H6	0.9689	0.9689	4.6786-07	4.6786-07	5.4537-09
C3H8	6.5447-02	6.5447-02	2.7126-13	2.7126-13	0.0
C4H10	9.2388-03	9.2388-03	1.9514-19	1.9514-19	0.0
C5H12	7.9749-04	7.9749-04	5.3779-26	5.3779-26	0.0
C6H14	3.0529-04	3.0529-04	7.4010-32	7.4010-32	0.0
N2	7.3155	7.3155	7.5655	7.5655	2.2530
O2	5.8820-03	5.8820-03	4.8810-03	4.8810-03	1.0512-03
CO2	0.3235	0.3235	1.9915-06	1.9915-06	3.3205-08
TOTAL FLOW:					
KMOL/HR	47.8314	47.8314	19.9720	19.9720	3.7172
KG/SEC	0.2444	0.2444	0.1141	0.1141	2.4061-02
CUM/SEC	9.6303-04	1.6785-03	1.5088-03	5.1430-04	2.6883-04
STATE VARIABLES:					
TEMP C	-92.3867	-102.0710	-109.9262	-121.4070	-121.4070
PRES BAR	47.0000	32.9000	32.0000	31.5000	31.5000
VFRAC	0.0	0.3139	1.0000	0.1861	1.0000
LFRAC	1.0000	0.6860	0.0	0.8138	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-7.4902+07	-7.4902+07	-5.2372+07	-5.6100+07	-3.5436+07
J/KG	-4.0714+06	-4.0714+06	-2.5446+06	-2.7258+06	-1.5207+06
WATT	-9.9518+05	-9.9518+05	-2.9055+05	-3.1123+05	-3.6590+04
ENTROPY:					
J/KMOL-K	-1.3796+05	-1.3719+05	-9.9377+04	-1.2292+05	-8.2448+04
J/KG-K	-7499.2351	-7457.3461	-4828.5361	-5972.2615	-3538.1620
DENSITY:					
KMOL/CUM	13.7965	7.9155	3.6769	10.7870	3.8408
KG/CUM	253.8137	145.6224	75.6752	222.0102	89.5024
AVG MW	18.3969	18.3969	20.5812	20.5812	23.3025

Table A8-1. Main streams of cryogenic method II.

COMPONENTS [kmol/hr]	STREAM ID 1	STREAM ID 31
CH ₄	43.1797	43.1503
C ₂ H ₆	1.5131	1.5131
C ₃ H ₈	0.1973	0.1973
C ₄ H ₁₀	7.3985-02	7.3984-02
C ₅ H ₁₂	2.0791-02	2.0791-02
C ₆ H ₁₄	2.6691-02	2.6691-02
N ₂	7.6178	1.5542
O ₂	6.2502-03	5.3823-03
CO ₂	0.4726	0.4726
Total flow [kmol/hr]	53.1083	47.0145
Temperature [°C]	15.00	-115.8763
Pressure [bar]	20.00	15.00
Phase	VAPOR	MIXED

A8.2. Simulation and results of cryogenic method I on Aspen Plus

Cryogenic method I was simulated on Aspen Plus. The stream and equipment numbers are not the same as the numbers on the flowsheet shown in figure 8.1. The numbers used in this appendix are the numbers used in the Aspen Plus simulation.

The most important streams are printed in table A8.2 on the next page. Stream 1 in this table is the feed stream of the plant. Streams 8 and 23 are the nitrogen-poor and the nitrogen-rich flow respectively. The flowsheet of Aspen Plus is shown in figure A8.2 on page A8-38.

Further all results of the Aspen Plus simulation are shown on the following pages.

Block B9 and B10 are respectively the reboiler and the condensor of the column. Block B19 is the condensor in the middle of the column. Block B21 is defined in the simulating, but it works as a compressor. Defining this block as compressor gave some simulation problems.

Results

The simulation of method I gave the following results:

- 81.6 mol % N_2 is removed from the feed
- the new feed stream contains 3.2 mol % N_2
- the purity of CH_4 in the feed stream goes from 81.3 % to 91.7 %
- 2.5 kmol CH_4 leaves the process in stream 23 (this is 5.8 % of the total CH_4 flow)

Table A8.2. Main streams of cryogenic method I.

COMPONENTS [kmol/hr]	STREAM ID 1	STREAM ID 8	STREAM ID 23
CH ₄	43.29	40.7839	2.5059
C ₂ H ₆	1.517	1.5170	7.3088-18
C ₃ H ₈	0.198	0.1980	6.2150-33
C ₄ H ₁₀	7.4-02	7.4-02	0.0
C ₅ H ₁₂	2.1-02	2.1-02	0.0
C ₆ H ₁₄	.. 2.67-02	2.67-02	0.0
N ₂	7.62	1.4030	6.2169
O ₂	6.0-03	3.3911-03	2.6089-03
CO ₂	0.47	0.47	1.0341-15
Total flow [kmol/hr]	53.223	44.4971	8.7255
Temperature [°C]	15.00	18.0845	15.00
Pressure [bar]	54.00	5.7688	2.00
Phase	VAPOR	VAPOR	VAPOR

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BLOCK: B2 MODEL: MHEATX

HOT SIDE: INLET STREAM OUTLET STREAM

 1 2

COLD SIDE: INLET STREAM OUTLET STREAM

 7 8
 22A 23

PROPERTIES FOR STREAM 1
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

PROPERTIES FOR STREAM 22A
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

PROPERTIES FOR STREAM 7
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

	*** MASS AND ENERGY BALANCE ***		
	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	106.445	106.445	0.
MASS (KG/HR)	1982.60	1982.60	0.
ENTHALPY (WATT)	-0.216216E+07	-0.216216E+07	0.529260E-07

*** INPUT DATA ***

SPECIFICATIONS FOR STREAM 1 :
TWO PHASE TP FLASH
SPECIFIED TEMPERATURE CHANGE C -120.000
PRESSURE DROP BAR 0.0
MAXIMUM NO. ITERATIONS 500
CONVERGENCE TOLERANCE 0.000100000

BLOCK: B2 MODEL: MHEATX (CONTINUED)

SPECIFICATIONS FOR STREAM 22A :
 ONE PHASE TP FLASH SPECIFIED PHASE IS VAPOR
 SPECIFIED TEMPERATURE CHANGE C 125.000
 PRESSURE DROP BAR 0.0
 MAXIMUM NO. ITERATIONS 500
 CONVERGENCE TOLERANCE 0.000100000

SPECIFICATIONS FOR STREAM 7 :
 ONE PHASE FLASH SPECIFIED PHASE IS VAPOR
 MAXIMUM NO. ITERATIONS 500
 CONVERGENCE TOLERANCE 0.000100000

*** RESULTS ***

INLET STREAM	OUTLET TEMPERATURE C	OUTLET PRESSURE BAR	OUTLET VAPOR FRAC	DUTY WATT
1	-105.00	54.000	0.	-.14099E+06
22A	15.000	2.0000	1.0000	9318.0
7	18.085	5.7688	1.0000	0.13168E+06



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BLOCK: B3 MODEL: MHEATX

HOT SIDE: INLET STREAM OUTLET STREAM

 10 11

COLD SIDE: INLET STREAM OUTLET STREAM

 13 14
 19 200

PROPERTIES FOR STREAM 10
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

PROPERTIES FOR STREAM 19
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

PROPERTIES FOR STREAM 13
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

 *** MASS AND ENERGY BALANCE ***
 IN OUT RELATIVE DIFF.
TOTAL BALANCE
MOLE (KMOL/HR) 18.6137 18.6137 0.
MASS (KG/HR) 387.077 387.077 0.
ENTHALPY (WATT) -296630. -296630. 0.463637E-09

 *** INPUT DATA ***

SPECIFICATIONS FOR STREAM 10 :
TWO PHASE TP FLASH
SPECIFIED TEMPERATURE CHANGE C -7.00000
PRESSURE DROP BAR 0.0
MAXIMUM NO. ITERATIONS 500
CONVERGENCE TOLERANCE 0.000100000

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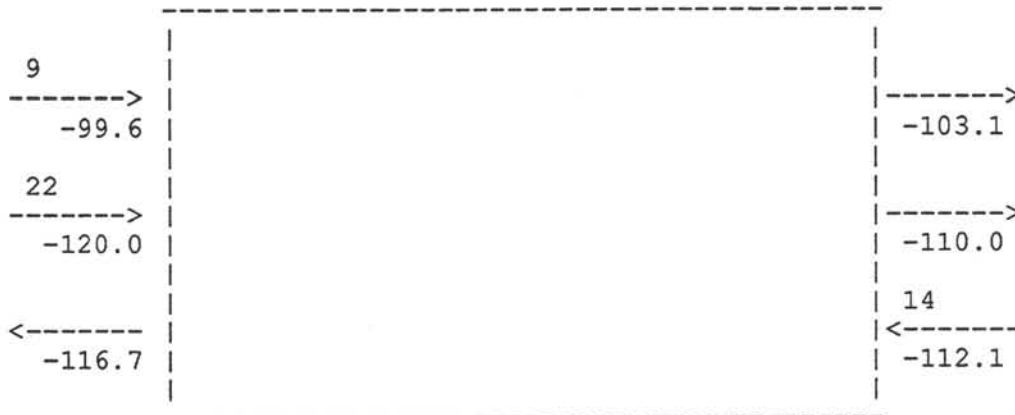
BLOCK: B4 MODEL: MHEATX (CONTINUED)

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SPECIFICATIONS FOR STREAM 22      :
TWO PHASE TP FLASH
SPECIFIED TEMPERATURE CHANGE     C          10.0000
PRESSURE DROP                     BAR         0.0
MAXIMUM NO. ITERATIONS            500
CONVERGENCE TOLERANCE              0.000100000

SPECIFICATIONS FOR STREAM 14      :
TWO PHASE FLASH
MAXIMUM NO. ITERATIONS            500
CONVERGENCE TOLERANCE              0.000100000
```

*** RESULTS ***

INLET STREAM	OUTLET TEMPERATURE C	OUTLET PRESSURE BAR	OUTLET VAPOR FRAC	DUTY WATT
9	-103.07	28.000	0.	-429.62
22	-110.00	2.0000	1.0000	748.15
14	-116.72	2.0000	0.98513	-318.53



BLOCK: B1 MODEL: RADFRAC

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-----
INLETS   - 4          STAGE 16
          18          STAGE 19
          MID         STAGE 10
OUTLETS  - 19         STAGE 1
          5           STAGE 20
```

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BLOCK: B1 MODEL: RADFRAC (CONTINUED)
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

	***	MASS AND ENERGY BALANCE	***	
		IN	OUT	RELATIVE DIFF.
TOTAL BALANCE				
MOLE (KMOL/HR)		58.1668	58.1668	0.131838E-05
MASS (KG/HR)		1077.62	1077.62	0.128652E-05
ENTHALPY (WATT)		-0.130696E+07	-0.126896E+07	-0.290763E-01

**** INPUT DATA ****

**** INPUT PARAMETERS ****

NUMBER OF STAGES		20
ALGORITHM OPTION		STANDARD
ABSORBER OPTION		NO
INITIALIZATION OPTION		STANDARD
HYDRAULIC PARAMETER CALCULATIONS		NO
INSIDE LOOP CONVERGENCE METHOD		BROYDEN
DESIGN SPECIFICATION METHOD		NESTED
MAXIMUM NO. OF OUTSIDE LOOP ITERATIONS		25
MAXIMUM NO. OF INSIDE LOOP ITERATIONS		10
MAXIMUM NUMBER OF FLASH ITERATIONS		50
FLASH TOLERANCE		0.000100000
OUTSIDE LOOP CONVERGENCE TOLERANCE		0.000100000

**** COL-SPECS ****

MOLAR VAPOR DIST / TOTAL DIST		1.00000
CONDENSER DUTY (W/O SUBCOOL)	WATT	-10,000.0
REBOILER DUTY	WATT	48,000.0

**** PROFILES ****

P-SPEC	STAGE	1	PRES, BAR	27.5000
TEMP-EST	STAGE	1	TEMP, C	-140.000
		20		-100.000

BLOCK: B1 MODEL: RADFRAC (CONTINUED)

 **** RESULTS ****

TOP STAGE TEMPERATURE	C	-129.855
BOTTOM STAGE TEMPERATURE	C	-99.5663
TOP STAGE LIQUID FLOW	KMOL/HR	7.09941
BOTTOM STAGE LIQUID FLOW	KMOL/HR	49.4413
TOP STAGE VAPOR FLOW	KMOL/HR	8.72550
BOTTOM STAGE VAPOR FLOW	KMOL/HR	34.1620
MOLAR REFLUX RATIO		0.81364
MOLAR BOILUP RATIO		0.69096
CONDENSER DUTY (W/O SUBCOOL)	WATT	-10,000.0
REBOILER DUTY	WATT	48,000.0

**** MAXIMUM FINAL RELATIVE ERRORS ****

DEW POINT	0.39666E-06	STAGE= 17
BUBBLE POINT	0.15760E-05	STAGE= 1
COMPONENT MASS BALANCE	0.31802E-05	STAGE= 18 COMP=C3H8
ENERGY BALANCE	0.16001E-05	STAGE= 18

**** PROFILES ****

STAGE	TEMPERATURE C	PRESSURE BAR	ENTHALPY J/KMOL		HEAT DUTY WATT
			LIQUID	VAPOR	
1	-129.85	27.500	-0.55925E+08	-0.27438E+08	-.10000+05
2	-122.05	27.526	-0.65604E+08	-0.37943E+08	
3	-119.39	27.553	-0.68338E+08	-0.41975E+08	
4	-118.61	27.579	-0.69077E+08	-0.43180E+08	
9	-118.21	27.711	-0.69297E+08	-0.43661E+08	
10	-118.19	27.737	-0.69284E+08	-0.43665E+08	-4114.8801
11	-116.20	27.763	-0.71136E+08	-0.46879E+08	
15	-115.14	27.868	-0.72296E+08	-0.48348E+08	
16	-114.19	27.895	-0.75567E+08	-0.48209E+08	
17	-110.24	27.921	-0.78641E+08	-0.55041E+08	
18	-106.08	27.947	-0.81824E+08	-0.62530E+08	
19	-103.10	27.974	-0.83960E+08	-0.68167E+08	
20	-99.566	28.000	-0.87555E+08	-0.73699E+08	.48000+05

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BLOCK: B1 MODEL: RADFRAC (CONTINUED)

STAGE	FLOW RATE KMOL/HR		FEED RATE KMOL/HR			PRODUCT RATE KMOL/HR	
	LIQUID	VAPOR	LIQUID	VAPOR	MIXED	LIQUID	VAPOR
1	7.099	8.725					8.7255
2	6.892	15.82					
3	6.891	15.62					
4	6.899	15.62					
9	6.929	15.65					
10	9.839	15.65					
11	9.876	18.56					
15	9.634	18.61					
16	80.33	18.36	53.2227				
17	81.66	35.83					
18	82.28	37.16		4.7977			
19	83.60	32.98	0.1463				
20	49.44	34.16				49.4412	

STAGE	**** X-PROFILE ****		****			
	CO2	N2	O2	CH4	C2H6	C3H8
1	.28392E-14	.39630	.27270E-03	.60343	.30099E-16	.23713E-30
2	.23121E-13	.26825	.22047E-03	.73153	.35671E-15	.22652E-28
3	.15616E-12	.23158	.19241E-03	.76823	.35348E-14	.17699E-26
4	.10090E-11	.22161	.18076E-03	.77821	.33554E-13	.13073E-24
9	.10362E-07	.21848	.17421E-03	.78135	.23529E-08	.24924E-15
10	.65340E-07	.21861	.17424E-03	.78122	.21773E-07	.17685E-13
11	.44759E-06	.19356	.16051E-03	.80628	.21807E-06	.13050E-11
15	.84628E-03	.18208	.14869E-03	.81502	.18682E-02	.30687E-04
16	.60938E-02	.17052	.14377E-03	.79984	.19422E-01	.24745E-02
17	.60840E-02	.12811	.13904E-03	.84243	.19307E-01	.24383E-02
18	.65877E-02	.86742E-01	.12267E-03	.88169	.20853E-01	.25208E-02
19	.68450E-02	.58520E-01	.10541E-03	.90871	.21539E-01	.26619E-02
20	.10562E-01	.31532E-01	.76209E-04	.91655	.34092E-01	.44497E-02

STAGE	**** X-PROFILE ****		
	C4H10	C5H12	C6H14
1	.11184E-43	.11568E-57	.77387E-71
2	.83900E-41	.74753E-54	.42004E-66
3	.49260E-38	.35933E-50	.16126E-61
4	.26924E-35	.15816E-46	.55776E-57
9	.10862E-21	.20546E-28	.20664E-34
10	.56217E-19	.84326E-25	.65656E-30
11	.29110E-16	.33224E-21	.19228E-25
15	.15505E-05	.55107E-07	.90558E-08
16	.92175E-03	.26145E-03	.33239E-03
17	.90702E-03	.25721E-03	.32698E-03
18	.90847E-03	.25563E-03	.32459E-03
19	.98548E-03	.27919E-03	.35487E-03
20	.16630E-02	.47194E-03	.60004E-03

BLOCK: B1 MODEL: RADFRAC (CONTINUED)

STAGE	**** Y-PROFILE ****					
	CO2	N2	O2	CH4	C2H6	C3H8
1	.11852E-15	.71250	.29900E-03	.28720	.83764E-18	.71229E-33
2	.13391E-14	.57064	.28720E-03	.42907	.13965E-16	.10677E-30
3	.10269E-13	.51645	.26435E-03	.48328	.15788E-15	.99969E-29
4	.68973E-13	.50029	.25197E-03	.49946	.15602E-14	.78099E-27
9	.72585E-09	.49389	.24379E-03	.50587	.11228E-09	.15488E-17
10	.45865E-08	.49384	.24377E-03	.50592	.10414E-08	.11032E-15
11	.34629E-07	.45075	.23288E-03	.54902	.11539E-07	.93729E-14
15	.68212E-04	.43132	.21961E-03	.56829	.10331E-03	.23425E-06
16	.44407E-03	.43417	.22013E-03	.56417	.98031E-03	.16103E-04
17	.54418E-03	.34312	.22768E-03	.65488	.12032E-02	.21382E-04
18	.72125E-03	.24375	.21428E-03	.75368	.16025E-02	.29789E-04
19	.86438E-03	.16913	.19186E-03	.82785	.19204E-02	.38967E-04
20	.14649E-02	.97580E-01	.14768E-03	.89736	.33712E-02	.74429E-04

STAGE	**** Y-PROFILE ****		
	C4H10	C5H12	C6H14
1	.38374E-47	.41010E-62	.29080E-76
2	.50193E-44	.51897E-58	.34718E-71
3	.37025E-41	.32988E-54	.18536E-66
4	.21736E-38	.15856E-50	.71157E-62
9	.92400E-25	.22002E-32	.28541E-39
10	.48079E-22	.90938E-29	.91462E-35
11	.29794E-19	.44691E-25	.34797E-30
15	.17144E-08	.81523E-11	.18350E-12
16	.81362E-06	.28917E-07	.47519E-08
17	.11739E-05	.45639E-07	.81839E-08
18	.17262E-05	.72961E-07	.14282E-07
19	.24715E-05	.11249E-06	.23536E-07
20	.49054E-05	.23113E-06	.50088E-07

STAGE	**** K-VALUES ****					
	CO2	N2	O2	CH4	C2H6	C3H8
1	.41743E-01	1.7979	1.0965	.47595	.27829E-01	.30037E-02
2	.57918E-01	2.1273	1.3027	.58654	.39149E-01	.47136E-02
3	.65762E-01	2.2301	1.3738	.62909	.44666E-01	.56482E-02
4	.68356E-01	2.2575	1.3939	.64180	.46499E-01	.59744E-02
9	.70047E-01	2.2606	1.3994	.64743	.47721E-01	.62138E-02
10	.70195E-01	2.2590	1.3990	.64760	.47830E-01	.62379E-02
11	.77369E-01	2.3288	1.4508	.68093	.52916E-01	.71826E-02
15	.80601E-01	2.3688	1.4770	.69726	.55297E-01	.76335E-02
16	.72873E-01	2.5462	1.5311	.70535	.50474E-01	.65075E-02
17	.89443E-01	2.6784	1.6376	.77737	.62319E-01	.87691E-02
18	.10949	2.8101	1.7468	.85482	.76849E-01	.11817E-01
19	.12628	2.8901	1.8201	.91102	.89159E-01	.14639E-01
20	.13869	3.0947	1.9378	.97906	.98885E-01	.16727E-01

BLOCK: B1 MODEL: RADFRAC (CONTINUED)

STAGE	**** K-VALUES ****		
	C4H10	C5H12	C6H14
1	.34312E-03	.35452E-04	.37576E-05
2	.59825E-03	.69425E-04	.82654E-05
3	.75163E-03	.91806E-04	.11495E-04
4	.80734E-03	.10025E-03	.12758E-04
9	.85064E-03	.10709E-03	.13812E-04
10	.85524E-03	.10784E-03	.13931E-04
11	.10235E-02	.13452E-03	.18096E-04
15	.11057E-02	.14793E-03	.20263E-04
16	.88268E-03	.11060E-03	.14296E-04
17	.12942E-02	.17744E-03	.25028E-04
18	.19001E-02	.28541E-03	.44000E-04
19	.25079E-02	.40292E-03	.66322E-04
20	.29497E-02	.48975E-03	.83475E-04

 ***** HYDRAULIC PARAMETERS *****

*** DEFINITIONS ***

FLOW PARAM = (ML/MV)*SQRT(RHOV/RHOL)

QR = QV*SQRT(RHOV/(RHOL-RHOV))

WHERE:

ML IS THE MASS FLOW OF LIQUID FROM THE STAGE

MV IS THE MASS FLOW OF VAPOR TO THE STAGE

RHOL IS THE MASS DENSITY OF LIQUID FROM THE STAGE

RHOV IS THE MASS DENSITY OF VAPOR TO THE STAGE

QV IS THE VOLUMETRIC FLOW RATE OF VAPOR TO THE STAGE

STAGE	MASS FLOW KG/HR		VOLUME FLOW CUM/SEC		DENSITY KG/CUM	
	LIQUID FROM	VAPOR TO	LIQUID FROM	VAPOR TO	LIQUID FROM	VAPOR TO
1	147.60	362.05	0.10053E-03	0.13814E-02	407.87	72.799
2	132.72	347.16	0.99822E-04	0.13854E-02	369.33	69.606
3	129.67	344.11	0.10099E-03	0.13899E-02	356.68	68.771
4	129.00	343.45	0.10150E-03	0.13907E-02	353.03	68.599
9	129.30	343.74	0.10225E-03	0.13859E-02	351.27	68.897
10	183.62	398.06	0.14523E-03	0.16588E-02	351.19	66.656
11	181.35	395.79	0.14729E-03	0.16652E-02	342.01	66.024
15	176.06	390.50	0.14412E-03	0.16722E-02	339.34	64.868
16	1500.1	723.28	0.11702E-02	0.33123E-02	356.09	60.657
17	1483.2	706.32	0.12173E-02	0.34756E-02	338.44	56.451
18	1456.8	679.98	0.12568E-02	0.36895E-02	322.00	51.195
19	1454.3	591.14	0.13014E-02	0.32576E-02	310.41	50.407
20	863.18	591.14	0.77289E-03	0.32576E-02	310.23	50.407

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BLOCK: B1 MODEL: RADFRAC (CONTINUED)

STAGE	VISCOSITY N-SEC/SQM		SURFACE TENSION N/M	FLOW PARAM	QR CUM/SEC
	LIQUID FROM	VAPOR TO	LIQUID FROM		
1	0.53785E-04	0.97746E-05	0.39674E-02	0.17224	0.64392E-03
2	0.51695E-04	0.96473E-05	0.38635E-02	0.16597	0.66765E-03
3	0.50622E-04	0.96104E-05	0.37277E-02	0.16547	0.67931E-03
4	0.50274E-04	0.96025E-05	0.36792E-02	0.16558	0.68298E-03
9	0.50004E-04	0.96156E-05	0.36450E-02	0.16659	0.68457E-03
10	0.49975E-04	0.95074E-05	0.36418E-02	0.20096	0.80289E-03
11	0.49088E-04	0.94743E-05	0.35058E-02	0.20132	0.81446E-03
15	0.48911E-04	0.94903E-05	0.34757E-02	0.19712	0.81293E-03
16	0.52459E-04	0.92400E-05	0.39383E-02	0.85601	0.15008E-02
17	0.50270E-04	0.89465E-05	0.35748E-02	0.85761	0.15551E-02
18	0.48240E-04	0.86754E-05	0.31911E-02	0.85428	0.16042E-02
19	0.46760E-04	0.84854E-05	0.28887E-02	0.99139	0.14343E-02
20	0.47201E-04	0.84854E-05	0.28468E-02	0.58860	0.14348E-02

 ***** TRAY SIZING CALCULATIONS *****

 *** SECTION 1 ***

STARTING STAGE NUMBER	2
ENDING STAGE NUMBER	19
FLOODING CALCULATION METHOD	GLITSCH

DESIGN PARAMETERS

PEAK CAPACITY FACTOR	1.00000	
SYSTEM FOAMING FACTOR	1.00000	
FLOODING FACTOR	0.80000	
MINIMUM COLUMN DIAMETER	METER	0.30480
MINIMUM DC AREA/COLUMN AREA		0.100000
HOLE AREA/ACTIVE AREA		0.12000

TRAY SPECIFICATIONS

TRAY TYPE	SIEVE	
NUMBER OF PASSES	1	
TRAY SPACING	METER	0.60960

BLOCK: B1 MODEL: RADFRAC (CONTINUED)

***** SIZING RESULTS @ STAGE WITH MAXIMUM DIAMETER *****

STAGE WITH MAXIMUM DIAMETER		19
COLUMN DIAMETER	METER	0.26078
DC AREA/COLUMN AREA		0.32316
DOWNCOMER VELOCITY	M/SEC	0.075395
WEIR LENGTH	METER	0.25023

**** SIZING PROFILES ****

STAGE	DIAMETER METER	TOTAL AREA SQM	ACTIVE AREA SQM	SIDE DC AREA SQM
2	0.11276	0.99863E-02	0.75198E-02	0.12332E-02
3	0.11387	0.10183E-01	0.76372E-02	0.12729E-02
4	0.11424	0.10250E-01	0.76758E-02	0.12872E-02
5	0.11437	0.10274E-01	0.76882E-02	0.12926E-02
6	0.11443	0.10284E-01	0.76930E-02	0.12954E-02
7	0.11447	0.10291E-01	0.76957E-02	0.12975E-02
8	0.11450	0.10297E-01	0.76979E-02	0.12995E-02
9	0.11453	0.10303E-01	0.76999E-02	0.13013E-02
10	0.12727	0.12721E-01	0.90384E-02	0.18414E-02
11	0.12841	0.12950E-01	0.91574E-02	0.18962E-02
12	0.12881	0.13032E-01	0.91990E-02	0.19164E-02
13	0.12895	0.13060E-01	0.92126E-02	0.19238E-02
14	0.12891	0.13051E-01	0.92082E-02	0.19214E-02
15	0.12782	0.12833E-01	0.91114E-02	0.18605E-02
16	0.24884	0.48632E-01	0.19509E-01	0.14561E-01
17	0.25516	0.51134E-01	0.20124E-01	0.15504E-01
18	0.26042	0.53265E-01	0.20597E-01	0.16334E-01
19	0.26078	0.53414E-01	0.18890E-01	0.17261E-01

BLOCK: B10 MODEL: HEATER

 INLET STREAM: 12
 OUTLET STREAM: 13
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

	*** MASS AND ENERGY BALANCE ***		
	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	4.94413	4.94413	0.
MASS (KG/HR)	86.3177	86.3177	0.
ENTHALPY (WATT)	-121453.	-109453.	-0.988036E-01

BLOCK: B10 MODEL: HEATER (CONTINUED)

*** INPUT DATA ***

TWO PHASE PQ FLASH
 SPECIFIED PRESSURE BAR 2.00000
 SPECIFIED HEAT DUTY WATT 12,000.0
 MAXIMUM NO. ITERATIONS 30
 CONVERGENCE TOLERANCE 0.000100000

*** RESULTS ***

OUTLET TEMPERATURE C -63.409
 OUTLET PRESSURE BAR 2.0000
 VAPOR FRACTION 0.99946

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CO2	0.10562E-01	0.53645E-02	0.10565E-01	1.9695
N2	0.31532E-01	0.15768E-03	0.31549E-01	200.09
O2	0.76209E-04	0.94360E-06	0.76249E-04	80.807
CH4	0.91655	0.27068E-01	0.91703	33.879
C2H6	0.34092E-01	0.20986E-01	0.34099E-01	1.6248
C3H8	0.44497E-02	0.24429E-01	0.44390E-02	0.18171
C4H10	0.16630E-02	0.75446E-01	0.16233E-02	0.21516E-01
C5H12	0.47194E-03	0.15478	0.38883E-03	0.25121E-02
C6H14	0.60004E-03	0.69177	0.22777E-03	0.32925E-03

BLOCK: B13 MODEL: COMPR

 INLET STREAM: 200
 OUTLET STREAM: 21
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	8.72550	8.72550	0.
MASS (KG/HR)	214.441	214.441	0.
ENTHALPY (WATT)	-63167.8	-67115.9	0.588259E-01

*** INPUT DATA ***

GAS PHASE CALCULATION
 NO FLASH PERFORMED
 TYPE : ISENTROPIC TURBINE
 PRESSURE RATIO 0.073000
 ISENTROPIC EFFICIENCY 0.72000
 MECHANICAL EFFICIENCY 1.00000

BLOCK: B13 MODEL: COMPR (CONTINUED)

*** RESULTS ***

INDICATED HORSEPOWER REQUIREMENT	WATT	-3,948.15
BRAKE HORSEPOWER REQUIREMENT	WATT	-3,948.15
NET WORK,	WATT	3,948.15
ISENTROPIC HORSEPOWER REQUIREMENT	WATT	-5,483.55
CALCULATED OUTLET PRES	BAR	2.00750
CALCULATED OUTLET TEMP	C	-174.215
ISENTROPIC TEMPERATURE	C	-193.632
OUTLET VAPOR FRACTION		1.00000

BLOCK: B15 MODEL: HEATER

 INLET STREAM: 6
 OUTLET STREAM: 7
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

	*** MASS AND ENERGY BALANCE ***		
	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	44.4971	44.4971	0.
MASS (KG/HR)	776.859	776.859	0.
ENTHALPY (WATT)	-0.108221E+07	-0.108221E+07	-0.360261E-07

*** INPUT DATA ***

TWO PHASE TO FLASH		
SPECIFIED TEMPERATURE	C	-135.000
SPECIFIED HEAT DUTY	WATT	0.0
MAXIMUM NO. ITERATIONS		30
CONVERGENCE TOLERANCE		0.000100000

BLOCK: B15 MODEL: HEATER (CONTINUED)

*** RESULTS ***
 OUTLET TEMPERATURE C -135.00
 OUTLET PRESSURE BAR 5.7688
 VAPOR FRACTION 0.36771

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CO2	0.10562E-01	0.16554E-01	0.26045E-03	0.15734E-01
N2	0.31532E-01	0.79240E-02	0.72126E-01	9.1021
O2	0.76209E-04	0.38588E-04	0.14090E-03	3.6514
CH4	0.91655	0.91053	0.92690	1.0180
C2H6	0.34092E-01	0.53589E-01	0.56654E-03	0.10572E-01
C3H8	0.44497E-02	0.70362E-02	0.22948E-05	0.32614E-03
C4H10	0.16630E-02	0.26302E-02	0.29294E-07	0.11138E-04
C5H12	0.47194E-03	0.74640E-03	0.24551E-09	0.32892E-06
C6H14	0.60004E-03	0.94900E-03	0.10216E-10	0.10766E-07

BLOCK: B16 MODEL: HEATER

 INLET STREAM: 11
 OUTLET STREAM: 12
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	4.94413	4.94413	0.
MASS (KG/HR)	86.3177	86.3177	0.
ENTHALPY (WATT)	-121453.	-121453.	-0.466853E-07

*** INPUT DATA ***

TWO PHASE PQ FLASH
 SPECIFIED PRESSURE BAR 2.00000
 SPECIFIED HEAT DUTY WATT 0.0
 MAXIMUM NO. ITERATIONS 30
 CONVERGENCE TOLERANCE 0.000100000

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BLOCK: B16 MODEL: HEATER (CONTINUED)

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*** RESULTS ***
OUTLET TEMPERATURE C -152.31
OUTLET PRESSURE BAR 2.0000
VAPOR FRACTION 0.36162
```

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CO2	0.10562E-01	0.16506E-01	0.71132E-04	0.43096E-02
N2	0.31532E-01	0.50818E-02	0.78224E-01	15.393
O2	0.76209E-04	0.31878E-04	0.15447E-03	4.8456
CH4	0.91655	0.91382	0.92139	1.0083
C2H6	0.34092E-01	0.53311E-01	0.16485E-03	0.30922E-02
C3H8	0.44497E-02	0.69702E-02	0.27192E-06	0.39012E-04
C4H10	0.16630E-02	0.26051E-02	0.14760E-08	0.56660E-06
C5H12	0.47194E-03	0.73928E-03	0.49679E-11	0.67199E-08
C6H14	0.60004E-03	0.93994E-03	0.87963E-13	0.93583E-10

BLOCK: B17 MODEL: HEATER

INLET STREAM: 3
OUTLET STREAM: 4
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

```
*** MASS AND ENERGY BALANCE ***
IN OUT RELATIVE DIFF.
TOTAL BALANCE
MOLE (KMOL/HR ) 53.2227 53.2227 0.
MASS (KG/HR ) 991.302 991.302 0.
ENTHALPY (WATT ) -0.119070E+07 -0.119070E+07 0.122882E-06
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*** INPUT DATA ***
TWO PHASE PQ FLASH
SPECIFIED PRESSURE BAR 28.0000
SPECIFIED HEAT DUTY WATT 0.0
MAXIMUM NO. ITERATIONS 150
CONVERGENCE TOLERANCE 0.000100000
```

BLOCK: B17 MODEL: HEATER (CONTINUED)

*** RESULTS ***

OUTLET TEMPERATURE	C	-135.80
OUTLET PRESSURE	BAR	28.000
VAPOR FRACTION		0.

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CO2	0.88308E-02	0.88308E-02	0.23944E-03	0.14917E-01
N2	0.14317	0.14317	0.47321	1.8157
O2	0.11273E-03	0.11273E-03	0.17807E-03	0.86786
CH4	0.81337	0.81337	0.52570	0.35524
C2H6	0.28503E-01	0.28503E-01	0.66524E-03	0.12844E-01
C3H8	0.37202E-02	0.37202E-02	0.73466E-05	0.10878E-02
C4H10	0.13904E-02	0.13904E-02	0.25142E-06	0.99702E-04
C5H12	0.39457E-03	0.39457E-03	0.59492E-08	0.83217E-05
C6H14	0.50167E-03	0.50167E-03	0.70384E-09	0.77512E-06

BLOCK: B18 MODEL: FSPLIT

INLET STREAM:	5	
OUTLET STREAMS:	6	9
PROPERTY OPTION SET:	SYSOP3	REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	49.4413	49.4413	0.
MASS (KG/HR)	863.177	863.177	0.
ENTHALPY (WATT)	-0.120245E+07	-0.120245E+07	0.

*** INPUT DATA ***

FRACTION OF FLOW	STRM=6	FRAC=	0.90000
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*** RESULTS ***

STREAM= 6	SPLIT=	0.90000	KEY= 0
9		0.100000	0

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BLOCK: B19 MODEL: HEATER

INLET STREAM: 21
OUTLET STREAM: 22
OUTLET HEAT STREAM: MID
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***
IN OUT RELATIVE DIFF.
TOTAL BALANCE
MOLE (KMOL/HR) 8.72550 8.72550 0.
MASS (KG/HR) 214.441 214.441 0.
ENTHALPY (WATT) -67115.9 -67115.9 0.216818E-15

*** INPUT DATA ***
TWO PHASE TP FLASH
SPECIFIED TEMPERATURE C -120.000
SPECIFIED PRESSURE BAR 2.00000
MAXIMUM NO. ITERATIONS 30
CONVERGENCE TOLERANCE 0.000100000

*** RESULTS ***
OUTLET TEMPERATURE C -120.00
OUTLET PRESSURE BAR 2.0000
HEAT DUTY WATT 4114.8
VAPOR FRACTION 1.0000

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
N2	0.71250	0.51936	0.71250	17.232
O2	0.29900E-03	0.26697E-03	0.29900E-03	14.660
CH4	0.28720	0.48038	0.28720	9.0474

BLOCK: B20 MODEL: HEATER

INLET STREAM: 17
OUTLET STREAM: 17A
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***
IN OUT RELATIVE DIFF.
TOTAL BALANCE
MOLE (KMOL/HR) 4.94413 4.94413 0.
MASS (KG/HR) 86.3177 86.3177 0.
ENTHALPY (WATT) -102771. -106120. 0.315597E-01

BLOCK: B20 MODEL: HEATER (CONTINUED)

*** INPUT DATA ***

TWO PHASE TP FLASH		
SPECIFIED TEMPERATURE	C	20.0000
PRESSURE DROP	BAR	0.0
MAXIMUM NO. ITERATIONS		30
CONVERGENCE TOLERANCE		0.000100000

*** RESULTS ***

OUTLET TEMPERATURE	C	20.000
OUTLET PRESSURE	BAR	28.500
HEAT DUTY	WATT	-3349.1
VAPOR FRACTION		1.0000

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CO2	0.10562E-01	0.15453E-01	0.10562E-01	2.6225
N2	0.31532E-01	0.22066E-01	0.31532E-01	4.9435
O2	0.76209E-04	0.59481E-04	0.76209E-04	4.4068
CH4	0.91655	0.87061	0.91655	3.7314
C2H6	0.34092E-01	0.64950E-01	0.34092E-01	2.3021
C3H8	0.44497E-02	0.11412E-01	0.44497E-02	1.5833
C4H10	0.16630E-02	0.66148E-02	0.16630E-02	1.0931
C5H12	0.47194E-03	0.29329E-02	0.47194E-03	0.75275
C6H14	0.60004E-03	0.59004E-02	0.60004E-03	0.51630

BLOCK: B21 MODEL: HEATER

 INLET STREAM: 16
 OUTLET STREAM: 17
 PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	4.94413	4.94413	0.
MASS (KG/HR)	86.3177	86.3177	0.
ENTHALPY (WATT)	-106307.	-102771.	-0.332650E-01

*** INPUT DATA ***

TWO PHASE TP FLASH		
SPECIFIED TEMPERATURE	C	80.0000
SPECIFIED PRESSURE	BAR	28.5000
MAXIMUM NO. ITERATIONS		30
CONVERGENCE TOLERANCE		0.000100000

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BLOCK: B21 MODEL: HEATER (CONTINUED)

*** RESULTS ***

OUTLET TEMPERATURE	C	80.000
OUTLET PRESSURE	BAR	28.500
HEAT DUTY	WATT	3536.3
VAPOR FRACTION		1.0000

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CO2	0.10562E-01	0.13622E-01	0.10562E-01	4.4383
N2	0.31532E-01	0.23812E-01	0.31532E-01	7.2088
O2	0.76209E-04	0.63865E-04	0.76209E-04	6.4840
CH4	0.91655	0.88024	0.91655	5.7685
C2H6	0.34092E-01	0.51552E-01	0.34092E-01	4.0815
C3H8	0.44497E-02	0.12974E-01	0.44497E-02	3.1739
C4H10	0.16630E-02	0.13403E-01	0.16630E-02	2.4767
C5H12	0.47194E-03	0.16043E-02	0.47194E-03	1.9447
C6H14	0.60004E-03	0.27289E-02	0.60004E-03	1.5297

BLOCK: B5 MODEL: HEATX

HOT SIDE:

INLET STREAM: 17A
OUTLET STREAM: 18
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

COLD SIDE:

INLET STREAM: 15
OUTLET STREAM: 16
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

*** MASS AND ENERGY BALANCE ***

	IN	OUT	RELATIVE DIFF.
TOTAL BALANCE			
MOLE (KMOL/HR)	9.88825	9.88826	-0.558420E-06
MASS (KG/HR)	172.635	172.635	-0.380306E-06
ENTHALPY (WATT)	-218448.	-218448.	0.691928E-06

*** INPUT DATA ***

FLASH SPECS FOR HOT SIDE:

TWO PHASE FLASH
MAXIMUM NO. ITERATIONS 30
CONVERGENCE TOLERANCE 0.000100000

BLOCK: B5 MODEL: HEATX (CONTINUED)

FLASH SPECS FOR COLD SIDE:
TWO PHASE FLASH
MAXIMUM NO. ITERATIONS
CONVERGENCE TOLERANCE

30
0.000100000

COUNTERCURRENT HEAT EXCHANGER WITH SPECIFIED COLD OUTLET TEMPERATURE
COLD STREAM OUTLET TEMPERATURE (C) 2.00000
HOT STREAM PRESSURE DROP (BAR) 1.50000
COLD STREAM PRESSURE DROP (BAR) 0.0

HEAT TRANSFER COEFFICIENTS (WATT/SQM-K) :

HOT STREAM PHASE	COLD STREAM PHASE	
LIQUID	LIQUID	850.000
BOILING LIQUID	LIQUID	850.000
VAPOR	LIQUID	850.000
LIQUID	BOILING LIQUID	850.000
BOILING LIQUID	BOILING LIQUID	850.000
VAPOR	BOILING LIQUID	850.000
LIQUID	VAPOR	850.000
BOILING LIQUID	VAPOR	850.000
VAPOR	VAPOR	850.000

*** RESULTS ***

HOT STREAM INLET TEMPERATURE (C)	20.0000
HOT STREAM OUTLET TEMPERATURE (C)	-79.8053
COLD STREAM INLET TEMPERATURE (C)	-116.718
COLD STREAM OUTLET TEMPERATURE (C)	2.00000
EXCHANGER HEAT DUTY (WATT)	6,021.03
HEAT TRANSFER AREA (SQM)	0.27696

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BLOCK: B5 MODEL: HEATX (CONTINUED)

AREA CALCULATION RESULTS:

SECTION	CONDITIONS (HOT-COLD)	AREA (SQM)	HEAT DUTY (WATT)
1	V-B	0.14972	2,719.20
2	B-V	0.0	-11.6519
3	V-B	0.12725	3,313.48

SECTION	TEMPERATURE LEAVING SECTION HOT STREAM (C)	TEMPERATURE LEAVING SECTION COLD STREAM (C)
1	-29.7179	2.00000
2	-29.5048	-54.8480
3	-79.8053	-54.6177

17A	VAP	BOIL	VAP	18
20.0	-29.7	-29.5	-79.8	
16	BOIL	VAP	BOIL	15
2.0	-54.8	-54.6	-116.7	

BLOCK: B9 MODEL: HEATER

INLET STREAM: 2
OUTLET STREAM: 3
PROPERTY OPTION SET: SYSOP3 REDLICH-KWONG-SOAVE EQUATION OF STATE

	*** MASS AND ENERGY BALANCE ***	***	RELATIVE DIFF.
TOTAL BALANCE	IN	OUT	
MOLE (KMOL/HR)	53.2227	53.2227	0.
MASS (KG/HR)	991.302	991.302	0.
ENTHALPY (WATT)	-0.115869E+07	-0.119070E+07	0.268826E-01

*** INPUT DATA ***

TWO PHASE PQ FLASH
SPECIFIED PRESSURE BAR 54.0000
SPECIFIED HEAT DUTY WATT -32,000.0
MAXIMUM NO. ITERATIONS 500
CONVERGENCE TOLERANCE 0.000100000

BLOCK: B9 MODEL: HEATER (CONTINUED)

*** RESULTS ***

OUTLET TEMPERATURE	C	-136.23
OUTLET PRESSURE	BAR	54.000
VAPOR FRACTION		0.

V-L PHASE EQUILIBRIUM :

COMP	F(I)	X(I)	Y(I)	K(I)
CO2	0.88308E-02	0.88308E-02	0.25583E-03	0.86622E-02
N2	0.14317	0.14317	0.46231	0.98334
O2	0.11273E-03	0.11273E-03	0.17133E-03	0.46192
CH4	0.81337	0.81337	0.53647	0.19954
C2H6	0.28503E-01	0.28503E-01	0.77945E-03	0.81502E-02
C3H8	0.37202E-02	0.37202E-02	0.98983E-05	0.78333E-03
C4H10	0.13904E-02	0.13904E-02	0.38949E-06	0.81491E-04
C5H12	0.39457E-03	0.39457E-03	0.10718E-07	0.78025E-05
C6H14	0.50167E-03	0.50167E-03	0.14999E-08	0.84790E-06

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1 10 11 12 13

STREAM ID	1	10	11	12	13
FROM :	----	B4	B3	B16	B10
TO :	B2	B3	B16	B10	B3
SUBSTREAM: MIXED					
PHASE:	VAPOR	LIQUID	LIQUID	MIXED	MIXED
COMPONENTS: KMOL/HR					
CO2	0.4700	5.2222-02	5.2222-02	5.2222-02	5.2222-02
N2	7.6200	0.1559	0.1559	0.1559	0.1559
O2	6.0000-03	3.7678-04	3.7678-04	3.7678-04	3.7678-04
CH4	43.2900	4.5315	4.5315	4.5315	4.5315
C2H6	1.5170	0.1685	0.1685	0.1685	0.1685
C3H8	0.1980	2.2000-02	2.2000-02	2.2000-02	2.2000-02
C4H10	7.4000-02	8.2222-03	8.2222-03	8.2222-03	8.2222-03
C5H12	2.1000-02	2.3333-03	2.3333-03	2.3333-03	2.3333-03
C6H14	2.6700-02	2.9667-03	2.9667-03	2.9667-03	2.9667-03
TOTAL FLOW:					
KMOL/HR	53.2227	4.9441	4.9441	4.9441	4.9441
KG/HR	991.3019	86.3177	86.3177	86.3177	86.3177
CUM/SEC	6.0091-03	7.4045-05	6.9072-05	2.4058-03	1.1823-02
STATE VARIABLES:					
TEMP C	15.0000	-103.0662	-110.0662	-152.3090	-63.4090
PRES BAR	54.0000	28.0000	28.0000	2.0000	2.0000
VFRAC	1.0000	0.0	0.0	0.3616	0.9994
LFRAC	0.0	1.0000	1.0000	0.6383	5.3832-04
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-6.8837+07	-8.7868+07	-8.8434+07	-8.8434+07	-7.9697+07
J/KG	-3.6959+06	-5.0329+06	-5.0654+06	-5.0654+06	-4.5649+06
WATT	-1.0177+06	-1.2068+05	-1.2145+05	-1.2145+05	-1.0945+05
ENTROPY:					
J/KMOL-K	-1.0411+05	-1.6176+05	-1.6516+05	-1.6063+05	-9.6987+04
J/KG-K	-5589.8717	-9265.1222	-9459.8357	-9200.4041	-5555.2636
DENSITY:					
KMOL/CUM	2.4603	18.5476	19.8832	0.5708	0.1161
KG/CUM	45.8244	323.8169	347.1337	9.9665	2.0279
AVG MW	18.6255	17.4586	17.4586	17.4586	17.4586

14 15 16 17 17A

STREAM ID	14	15	16	17	17A
FROM :	B3	B4	B5	B21	B20
TO :	B4	B5	B21	B20	B5
SUBSTREAM: MIXED					
PHASE:	MIXED	MIXED	VAPOR	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
CO2	5.2222-02	5.2222-02	5.2222-02	5.2222-02	5.2222-02
N2	0.1559	0.1559	0.1559	0.1559	0.1559
O2	3.7678-04	3.7678-04	3.7678-04	3.7678-04	3.7678-04
CH4	4.5315	4.5315	4.5315	4.5315	4.5315
C2H6	0.1685	0.1685	0.1685	0.1685	0.1685
C3H8	2.2000-02	2.2000-02	2.2000-02	2.2000-02	2.2000-02
C4H10	8.2222-03	8.2222-03	8.2222-03	8.2222-03	8.2222-03
C5H12	2.3333-03	2.3333-03	2.3333-03	2.3333-03	2.3333-03
C6H14	2.9667-03	2.9667-03	2.9667-03	2.9667-03	2.9667-03
TOTAL FLOW:					
KMOL/HR	4.9441	4.9441	4.9441	4.9441	4.9441
KG/HR	86.3177	86.3177	86.3177	86.3177	86.3177
CUM/SEC	8.8695-03	8.5600-03	1.5633-02	1.3862-03	1.1130-03
STATE VARIABLES:					
TEMP C	-112.1358	-116.7183	2.0000	80.0000	20.0000
PRES BAR	2.0000	2.0000	2.0000	28.5000	28.5000
VFRAC	0.9897	0.9851	1.0000	1.0000	1.0000
LFRAC	1.0210-02	1.4872-02	0.0	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-8.1558+07	-8.1790+07	-7.7406+07	-7.4831+07	-7.7270+07
J/KG	-4.6715+06	-4.6848+06	-4.4337+06	-4.2862+06	-4.4259+06
WATT	-1.1201+05	-1.1233+05	-1.0631+05	-1.0277+05	-1.0612+05
ENTROPY:					
J/KMOL-K	-1.0714+05	-1.0860+05	-8.7480+04	-1.0106+05	-1.0862+05
J/KG-K	-6136.9183	-6220.6520	-5010.6757	-5788.4810	-6221.6141
DENSITY:					
KMOL/CUM	0.1548	0.1604	8.7853-02	0.9907	1.2339
KG/CUM	2.7033	2.8010	1.5338	17.2970	21.5432
AVG MW	17.4586	17.4586	17.4586	17.4586	17.4586

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18 19 2 200 21

STREAM ID	18	19	2	200	21
FROM :	B5	B1	B2	B3	B13
TO :	B1	B3	B9	B13	B19
SUBSTREAM: MIXED					
PHASE:	MIXED	VAPOR	LIQUID	VAPOR	VAPOR
COMPONENTS: KMOL/HR					
CO2	5.2222-02	1.0341-15	0.4700	1.0341-15	1.0341-15
N2	0.1558	6.2169	7.6200	6.2169	6.2169
O2	3.7678-04	2.6089-03	6.0000-03	2.6089-03	2.6089-03
CH4	4.5315	2.5059	43.2900	2.5059	2.5059
C2H6	0.1685	7.3088-18	1.5170	7.3088-18	7.3088-18
C3H8	2.2000-02	6.2150-33	0.1980	6.2150-33	6.2150-33
C4H10	8.2222-03	0.0	7.4000-02	0.0	0.0
C5H12	2.3333-03	0.0	2.1000-02	0.0	0.0
C6H14	2.9667-03	0.0	2.6700-02	0.0	0.0
TOTAL FLOW:					
KMOL/HR	4.9441	8.7255	53.2227	8.7255	8.7255
KG/HR	86.3177	214.4412	991.3019	214.4412	214.4412
CUM/SEC	6.0795-04	7.1321-04	7.9051-04	1.0851-03	9.4043-03
STATE VARIABLES:					
TEMP C	-79.8052	-129.8547	-105.0000	-99.8547	-174.2148
PRES BAR	27.0000	27.5000	54.0000	27.5000	2.0075
VFRAC	0.9697	1.0000	0.0	1.0000	1.0000
LFRAC	3.0212-02	0.0	1.0000	0.0	0.0
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-8.1654+07	-2.7438+07	-7.8374+07	-2.6062+07	-2.7691+07
J/KG	-4.6770+06	-1.1164+06	-4.2079+06	-1.0604+06	-1.1267+06
WATT	-1.1214+05	-6.6502+04	-1.1587+06	-6.3168+04	-6.7116+04
ENTROPY:					
J/KMOL-K	-1.2682+05	-7.3989+04	-1.4997+05	-6.5218+04	-5.8085+04
J/KG-K	-7263.8423	-3010.5654	-8051.6661	-2653.6850	-2363.4266
DENSITY:					
KMOL/CUM	2.2590	3.3983	18.7020	2.2336	0.2577
KG/CUM	39.4392	83.5195	348.3353	54.8945	6.3340
AVG MW	17.4586	24.5763	18.6255	24.5763	24.5763

22 22A 23 3 4

STREAM ID	22	22A	23	3	4
FROM :	B19	B4	B2	B9	B17
TO :	B4	B2	----	B17	B1
SUBSTREAM: MIXED					
PHASE:	VAPOR	VAPOR	VAPOR	LIQUID	LIQUID
COMPONENTS: KMOL/HR					
CO2	1.0341-15	1.0341-15	1.0341-15	0.4700	0.4700
N2	6.2169	6.2169	6.2169	7.6200	7.6200
O2	2.6089-03	2.6089-03	2.6089-03	6.0000-03	6.0000-03
CH4	2.5059	2.5059	2.5059	43.2900	43.2900
C2H6	7.3088-18	7.3088-18	7.3088-18	1.5170	1.5170
C3H8	6.2150-33	6.2150-33	6.2150-33	0.1980	0.1980
C4H10	0.0	0.0	0.0	7.4000-02	7.4000-02
C5H12	0.0	0.0	0.0	2.1000-02	2.1000-02
C6H14	0.0	0.0	0.0	2.6700-02	2.6700-02
TOTAL FLOW:					
KMOL/HR	8.7255	8.7255	8.7255	53.2227	53.2227
KG/HR	214.4412	214.4412	214.4412	991.3019	991.3019
CUM/SEC	1.5198-02	1.6236-02	2.9011-02	6.2850-04	6.4006-04
STATE VARIABLES:					
TEMP C	-120.0000	-110.0000	15.0000	-136.2263	-135.7970
PRES BAR	2.0000	2.0000	2.0000	54.0000	28.0000
VFRAC	1.0000	1.0000	1.0000	0.0	0.0
LFRAC	0.0	0.0	0.0	1.0000	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-2.5993+07	-2.5685+07	-2.1840+07	-8.0539+07	-8.0539+07
J/KG	-1.0577+06	-1.0451+06	-8.8866+05	-4.3241+06	-4.3241+06
WATT	-6.3001+04	-6.2253+04	-5.2935+04	-1.1907+06	-1.1907+06
ENTROPY:					
J/KMOL-K	-4.4354+04	-4.2401+04	-2.4910+04	-1.6415+05	-1.6333+05
J/KG-K	-1804.7275	-1725.2827	-1013.5797	-8812.9866	-8769.3321
DENSITY:					
KMOL/CUM	0.1594	0.1492	8.3547-02	23.5228	23.0978
KG/CUM	3.9193	3.6689	2.0532	438.1255	430.2106
AVG MW	24.5763	24.5763	24.5763	18.6255	18.6255

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5 6 7 8 9

STREAM ID	5	6	7	8	9
FROM :	B1	B18	B15	B2	B18
TO :	B18	B15	B2	----	B4
SUBSTREAM: MIXED					
PHASE:	LIQUID	LIQUID	MIXED	VAPOR	LIQUID
COMPONENTS: KMOL/HR					
CO2	0.5222	0.4700	0.4700	0.4700	5.2222-02
N2	1.5589	1.4030	1.4030	1.4030	0.1559
O2	3.7678-03	3.3911-03	3.3911-03	3.3911-03	3.7678-04
CH4	45.3155	40.7839	40.7839	40.7839	4.5315
C2H6	1.6855	1.5170	1.5170	1.5170	0.1685
C3H8	0.2200	0.1980	0.1980	0.1980	2.2000-02
C4H10	8.2222-02	7.4000-02	7.4000-02	7.4000-02	8.2222-03
C5H12	2.3333-02	2.1000-02	2.1000-02	2.1000-02	2.3333-03
C6H14	2.9667-02	2.6700-02	2.6700-02	2.6700-02	2.9667-03
TOTAL FLOW:					
KMOL/HR	49.4412	44.4971	44.4971	44.4971	4.9441
KG/HR	863.1770	776.8593	776.8593	776.8593	86.3177
CUM/SEC	7.7289-04	6.9560-04	8.4152-03	5.1299-02	7.7289-05
STATE VARIABLES:					
TEMP C	-99.5662	-99.5662	-135.0000	18.0845	-99.5662
PRES BAR	28.0000	28.0000	5.7688	5.7688	28.0000
VFRAC	0.0	0.0	0.3677	1.0000	0.0
LFRAC	1.0000	1.0000	0.6322	0.0	1.0000
SFRAC	0.0	0.0	0.0	0.0	0.0
ENTHALPY:					
J/KMOL	-8.7555+07	-8.7555+07	-8.7555+07	-7.6902+07	-8.7555+07
J/KG	-5.0150+06	-5.0150+06	-5.0150+06	-4.4048+06	-5.0150+06
WATT	-1.2025+06	-1.0822+06	-1.0822+06	-9.5053+05	-1.2025+05
ENTROPY:					
J/KMOL-K	-1.5994+05	-1.5994+05	-1.5687+05	-9.4438+04	-1.5994+05
J/KG-K	-9160.8578	-9160.8578	-8985.3131	-5409.2686	-9160.8578
DENSITY:					
KMOL/CUM	17.7692	17.7692	1.4688	0.2409	17.7692
KG/CUM	310.2261	310.2261	25.6433	4.2066	310.2261
AVG MW	17.4586	17.4586	17.4586	17.4586	17.4586

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MID

STREAM ID	MID
FROM :	B19
TO :	B1
CLASS:	HEAT

STREAM ATTRIBUTES:

HEAT

Q	WATT	-4114.8363
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APPENDIX 9

The factors and coefficients for the investment cost calculation of the design process

The dimension of C_i is $10^3 \times$ US \$. The last column of the tables is the PCE.

Table A9.1. The furnace.

Code	F_m	F_p	F_t	Q (kW)	C_{fu0}	C_{fu}
H-2101	1.6	1	1.5	4170	335.8	857.5

Table A9.2. The pressure vessels.

Code	F_m	F_p	l (m)	d (m)	a	b	C_{pvo}	C_{pv}
V-2101	1	1.4	4.10	0.66	1500	1100	3.81	5.67
V-2102	1	1.4	3.60	1.00	1500	1100	5.46	8.14
V-2103	1	1.4	3.40	1.05	890	705	3.47	5.17
V-2104	3.7	1	5.24	0.70	1294	1141	4.91	19.33
V-2107	3.7	1.2	2.55	0.65	1500	1100	2.68	12.67
V-2108	3.7	1.2	2.30	0.61	1500	1100	2.34	11.05
V-2109	3.7	1	2.60	0.51	1500	1100	2.07	8.15
V-2110	1	1	4.80	2.00	1500	1100	14.53	15.47
V-2111	1	1	1.20	0.70	1500	1100	1.91	2.03
V-2112	1	1.4	3.60	1.00	1500	1100	5.46	8.14
V-2113	3.7	1.2	1.85	0.45	1500	1100	1.47	6.94
V-2114	3.7	1.2	1.85	0.35	1500	1100	1.11	5.26
V-2115	1	1.2	4.50	0.66	1500	1100	4.09	5.22
V-2116	1	1.2	2.35	0.61	1500	1100	2.37	3.03
V-2117	1	1.6	2.35	0.51	1500	1100	1.94	3.30
V-2119	1	1.2	2.00	0.41	1500	1100	1.37	1.76

Table A9.3. The columns without internals.

Code	F_m	F_p	l (m)	d (m)	a	b	C_{pvo}	C_{pv}
V-2105	1	1.2	21.56	0.76	1294	1141	18.86	24.09
V-2106	2.3	1.1	19.90	0.90	1294	1141	21.37	57.55
V-2118	1	1.6	4.48	1.10	1500	1100	7.13	12.14
T-21	3.7	1.6	14.70	0.80	1294	1141	14.13	89.03
T-22	3.7	1.2	10.54	0.40	1294	1141	4.86	22.97
T-23	3.7	1.2	6.02	0.40	1294	1141	2.98	14.07
T-24	3.7	1.2	5.55	0.40	1294	1141	2.78	13.14

Table A9.4. The trays.

Code	N_{tr}	F_{nt}	F_m	F_{ut}	d (m)	C_{tr}	C_{int}
V-2105	1	3	1	1	0.76	0.16	0.50
V-2106	4	2.5	1.7	1	0.90	0.18	3.28
V-2118	1	3	1	1	1.10	0.22	0.70
T-21	20	1	1.7	1	0.80	0.16	5.91
T-22	25	1	1.7	2.9	0.40	0.10	13.44
T-23	11	1.5	1.7	2.9	0.40	0.10	8.87
T-24	11	1.5	1.7	2.9	0.40	0.10	8.87

Table A9.5. The packings.

Code	$V_{packing}$ (m ³)	C_{po}	N_{db}	C_{do}	C_{pd}
V-2105	6.0	15.00	3	4.10	19.10
V-2106	5.7	14.18	3	5.70	19.88

$V_{packing}$ is the volume of the packing, C_{po} is the price of the packing (Pall rings), N_{db} the number of distributors, C_{do} the price of the distributors and C_{pd} the total price.

Table A9.6. The compressors.

Code	F_m	P (kW)	a	b	C_{como}	C_{com}
K-2103	1	200	6.628	0.968	76.57	102.60
K-2104	1	300	6.628	0.968	113.40	151.90
K-2105	1	140	6.628	0.968	54.21	72.62

Table A9.7. The drivers.

Code	F_t	F_{red}	C_{emo}	C_{em}
K-2103	1.4	1	11.91	19.23
K-2104	1.4	1	16.54	26.70
K-2105	1.4	1	8.77	14.16

Table A9.8. The heat exchangers.

Code	F_t	F_p	A (m ²)	a	b	C_{heo}	C_{he}
H-2102	0.8	1.25	4.6	7.085	0.656	3.23	3.43
H-2103	0.8	1.1	22.7	6.587	0.9955	16.24	15.21
H-2104	0.8	1.25	9.1	6.587	0.9955	6.54	6.96
H-2105	0.8	1.25	9.1	7.085	0.656	5.08	5.41
H-2106	0.8	1.3	18.2	7.085	0.656	8.01	8.87
H-2107	1	1.3	23.9	7.085	0.656	9.58	13.25
H-2108	0.85	1	50.0	7.085	0.656	15.54	14.06
H-2109A	0.85	1.1	35.2	7.085	0.656	12.35	12.29
H-2109B	0.85	1.1	35.2	6.587	0.9955	25.14	25.01
H-2110	0.85	1.1	83.3	6.587	0.9955	59.25	58.96
H-2111	0.85	1.1	51.0	6.587	0.9955	36.36	36.18
H-2112	0.85	1	28.4	6.587	0.9955	20.30	18.36
H-2113	0.8	1.1	14.9	7.085	0.656	7.02	6.58
H-2114	0.85	1.1	9.3	7.085	0.656	5.16	5.13
H-2115	0.85	1.3	15.5	7.085	0.656	7.21	8.48
H-2116	0.85	1.25	6.5	7.085	0.656	4.08	4.61
H-2117	0.85	1.1	7.5	7.085	0.656	4.48	4.46
H-2118	0.8	1.1	16.8	7.085	0.656	7.60	7.12
H-2119	0.8	1.1	7.2	7.085	0.656	4.36	4.08
H-2120	0.85	1	6.6	7.085	0.656	4.12	3.72
E-21	0.8	1.3	10.0*	7.439	0.8861	13.09	14.48
E-22	1.4	1.3	18.0	7.439	0.8861	22.03	42.66
E-23L	1.4	1.3	62.2	8.6063	0.601	65.42	126.69
E-23U	1.4	1.3	116.7	8.6063	0.601	95.49	184.92
E-24	1.4	1.1	4.3	7.439	0.8861	6.19	10.15
E-25	1.4	1.1	4.3	7.439	0.8861	6.19	10.15
E-27	1.4	1.1	4.3	7.439	0.8861	6.19	10.15

* The transfer area of the heat exchanger E-21 is an estimation.

APPENDIX 10

H₂-PSA design

The basic parameters for the design of the H₂-PSA system are in following table:

Table A10.1 : Basic data for H₂-PSA system

Feedgas (ChemCad)	adsorbent (zeolite 5A)	PSA system
$\omega_o = 0.68 \text{ kg/s}$ $\rho_o = 17.1 \text{ kg/m}^3$ $\mu = 1.416 * 10^{-5}$	$D_p = 0.016 \text{ m}$ $\epsilon = 0.42$ $\rho_a = 786 \text{ kg/m}^3$	$P_{ads} = 27 \text{ bar}$ $P_{des} = 1 \text{ bar}$ $t_c = 600 \text{ sec}$

With the equations 10.4...10.7 and the basic data above, the minimal column diameter is calculated:

$$v_s(\text{max}) = 1.02 \text{ m/s}$$

$$A(\text{min}) = 0.082 \text{ m}_2$$

$$d(\text{min}) = 0.32 \text{ m}$$

Table A10.2 : PSA-feedgas

component	feedgas (mol/s)
H ₂	27.17
CO	0.00
CO ₂	11.83
N ₂	4.76
CH ₄	0.513

According to the adsorption- and desorption pressure, cycle time (table A10.1) and the amount of adsorbed component in equilibrium, the column height is determined with equation 10.8. Taking a column diameter greater than the minimum diameter thinking about the pressure drop ($A = 0.65 \text{ m}_2$, $d = 0.90 \text{ m}$) the length of the H₂-PSA beds is:

$$L = 1.6 \text{ m}$$

With equation 10.9 the pressure drop over these beds is:

$$\delta P = 0.12 \text{ bar}$$

An estimation of the practical work of compression is made by calculating the energy necessary for compressing (efficiency = 0.75) the feed from desorption pressure to adsorption pressure:

$$P_{\text{total}} = 1.0 \text{ MW}$$

CO-VSA design

The basic parameters for the design of the CO-VSA system are in following table:

Table A10.3 : basic data for CO-VSA system

Feedgas (ChemCad)	adsorbent (CO-selective)	VSA system
$\omega_o = 1.14 \text{ kg/s}$ $\rho_o = 11.78 \text{ kg/m}^3$ $\mu = 1.597 * 10^{-5}$	$D_p = 0.016 \text{ m}$ $\epsilon = 0.42$ $\rho_a = 800 \text{ kg/m}^3$	$P_{ads} = 1 \text{ bar}$ $P_{des} = 0.04 \text{ bar}$ $t_c = 600 \text{ sec}$

With the equations 10.4 .. 10.7 and the basic data above, the minimal column diameter is calculated:

$$v_s(\text{max}) = 0.88 \text{ m/s}$$

$$A(\text{min}) = 0.12 \text{ m}^2$$

$$d(\text{min}) = 0.37 \text{ m}$$

Table A10.4 : CO-VSA feedgas

component	feedgas (mol/s)
H ₂	27.17
CO	16.00
CO ₂	11.83
N ₂	4.76
CH ₄	0.513

According to the adsorption- and desorption pressure, cycle time (table A10.3) and the amount of adsorbed component in equilibrium, the column height is determined with equation 10.8. Taking a column diameter greater than the minimum diameter thinking about the pressure drop ($A = 0.5 \text{ m}^2$, $d = 0.80 \text{ m}$) the length of the CO-VSA beds is:

$$L = 1.37 \text{ m}$$

With equation 10.9 the pressure drop over these beds and the total pressure drop are:

$$\delta P = 0.11 \text{ bar} \qquad \delta P_{\text{total}} = \delta P_{\text{psa}} + \delta P_{\text{vsa}} = 0.23 \text{ bar}$$

An estimation of the practical work of vacuum treatment is made by calculating the energy necessary (efficiency = 0.75) to change the feed from adsorption pressure to desorption pressure:

$$P_{\text{total}} = 0.3 \text{ MW}$$

CO-VSA and H₂-PSA investment costTable A10.5: H₂ PSA equipment cost(PCE)

investment cost for H ₂ PSA	prices in fl. (Webci)
5 columns (d=0.90, L=1.6)	175000.-
18 valves (pneumatic)	54000.-
compressor (1 MW ,acc.Olujić)	600000.-
controlsystem (64*in/output)	168000.-
Zeolite 5A (22 m ³)	120000.-
Purchased Cost Equipment(PCE):	1117000.-

Table A10.6: H₂-PSA system investment cost

H ₂ -PSA system	f _a	f _b	PCE	PPC	FC
	3.25	1.45	1.12*10 ⁶	3.63*10 ⁶	5.26*10 ⁶

Table A10.7 : CO-VSA equipment cost(PCE)

investment cost for CO VSA	prices in fl. (Webci)
5 columns (d=0.8, L=1.40)	185000.-
18 valves (pneumatic)	54000.-
vacuum-pump (0.3 MW)	314000.-
controlsystem (64*in/output)	168000.-
CO-selective adsorbent (20 m ³)	156000.-
Purchased Cost Equipment(PCE):	877000.-

Table A10.8: CO-VSA system investment cost

CO-VSA system	f _a	f _b	PCE	PPC	FC
	3.25	1.45	877000.-	2.85*10 ⁶	4.13*10 ⁶

Item	Process type		
	Fluids	Fluids- solids	Solids
1. MAJOR EQUIPMENT, TOTAL PURCHASE COST	PCE	PCE	PCE
f_1 Equipment erection	0.4	0.45	0.50
f_2 Piping	0.70	0.45	0.20
f_3 Instrumentation	0.20	0.15	0.10
f_4 Electrical	0.10	0.10	0.10
f_5 Buildings, process	0.15	0.10	0.05
f_6 Utilities	0.50	0.45	0.25
f_7 Storages	0.15	0.20	0.25
f_8 Site development	0.05	0.05	0.05
f_9 Ancillary buildings	0.15	0.20	0.30
2. TOTAL PHYSICAL PLANT COST (PPC)			
PPC = PCE (1 + f_1 ... + f_9)			
= PCE x	3.40	3.15	2.80
f_{10} Design and Engineering	0.30	0.25	0.20
f_{11} Contractor's fee	0.05	0.05	0.05
f_{12} Contingency	0.10	0.10	0.10
FIXED CAPITAL = PPC(1 + f_{10} + f_{11} + f_{12})			
= PPC x	1.45	1.40	1.35

* Omitted for minor extensions or additions to existing sites.

Fig A10.1 : factor for factoring method

Table A10.9 : H₂ PSA + CO VSA equipment cost(PCE)

investment cost for H ₂ PSA + CO VSA	prices in fl. (Webci)
5 columns (d=0.8, L=1.40)	175000.-
5 columns (d=0.90, L=1.6)	185000.-
36 valves (pneumatic)	108000.-
compressor (1 MW ,acc.Olujić)	600000.-
vacuumpump (0.3 MW)	314000.-
controlsystem (64*in/output)	168000.-
CO selective adsorbent (20 m ³)	156000.-
Zeolite 5A (22 m ³)	120000.-
Purchased Cost Equipment(PCE):	1826000.-

Table A10.10 : H₂-PSA + CO VSA system investment cost

	f_a	f_b	PCE	PPC	FC
H ₂ -PSA system	3.25	1.45	1.83*10 ⁶	5.93*10 ⁶	8.61*10 ⁶

COSORB® cost calculations

Requirements	Steam-natural gas reformer CO ₂ recycle
CO [kg/h]	1500
COSORB unit investment [MM \$]	7.3
Feed required [kmol/hr] dry	989.2
Utilities	
Power (kWh)	1,102
Steam (M kg/h)	9.9
Cooling Water (M kg/h)	140
Toluene (kg/h)	20

Table A10.11 : Investment, feedgas, and utility requirements

Costs	Steam-natural gas reformer CO ₂ recycle
	¢/kg CO
Utilities & Chemicals	
Power @ 4.5 ¢/kWh	1.06
Steam @ 9.0 \$/Mkg	0.95
Cooling Water @ 15 ¢/gal	0.06
Toluene @ 1.50 \$/gal	0.08
Catalyst & Chemicals	0.10
Total Variable Cost	2.25
Fixed cost	
Operating manpower	0.22
Maintenance @ 8% investment/yr	0.73
Insurance & Taxes	0.18
Depreciation @ 10 yr Straight life	0.93
Total Fixed Cost	2.06
Investment	
Cost at 30% of investment yr	2.80
Total Recovery Cost	7.11

Table A10.12 : CO Recovery costs

PROGRAM COSORBER [© I. Ketelaars, 1993];

Uses Crt, Iwan, Iwan2, Dos, IwanScr;

CONST

```

Nmax = 400;
R = 8.314E-2;           { bar*M3/kmol*K }
TcCO = 132.9;          TcCH4 = 190.6;
TcCO2 = 304.2;         TcN2 = 126.2;
TcH2 = 33.2;           TcAr = 150.8;
TcTol = 591.7;         { K }
PcCO = 35.0;           PcCH4 = 46.0;
PcCO2 = 73.8;          PcN2 = 33.9;
PcH2 = 13.0;           PcAr = 48.7;
PcTol = 41.1;         { Bar }
MCuAlCl4 = 232.3;      Mtol = 92.14;
MCO = 28.01;           MCH4 = 16.04;
MH2 = 2.016;           MN2 = 28.01;
MCO2 = 44.01;          MAr = 39.948;   { kg/kmol }

```

VAR

```

x1, x2, y1, y2, Value, komma, it           : Byte;
a, Teller1, Teller2, Step, N                : Word;
k1, k2, k3, k4, h, Z, Ze, at, Eff,
Kog, hg_a, Gb, GN2, GCO1,
PTol, GCH4, GH2, GAr, GCO2, LCu,
aw, xCu, P, dp, Tgest,
Enh, Sig_l, F, Hv1, Hv0, Ro_l,
Lab_g, Visc_l, sigc, Visc_Tol,
Ro_g, Visc_g, Keq, corr, kg_CO,
kl_CO, Hos, Hv, Dg_CO, He, Le,
Dl_CO, Dg_Tol, Dg, kg_Tol, TITemp,
YTol2, L2, Tg2, TI2, yCO2, TotError,
KeyError, yN2, yCH4, yAr, yH2, Mavr_g,
CpTol, CpCO2, CpCO, CpH2, CpCH4, CpN2,
Cpb, CpAr, Mavr_l, Prm, Trm,
Tcm, Pcm, epsy, epst, epsg                 : Real;
Y                                             : ARRAY[0..1] OF REAL;
iri                                          : ARRAY[1..8] OF Byte;
YCO, YTol, Tg, TI, XCO, XTol,
YCOe, YTole, L, G, Ra, Lo,
Cpg, Cpq, yb                               : ARRAY[0..Nmax] OF REAL;
Open, Initer, siap, First,
NoInput, SaveProf, Nooit                   : Boolean;

```

(*****)

PROCEDURE Get_Fract; { - } { Gas fraction of component i }

BEGIN

yN2 := GN2/G[i];

yAr := GAr/G[i];

yCH4 := GCH4/G[i];

yH2 := GH2/G[i];

yCO2 := GCO2/G[i];

yb[i] := Gb/G[i];

END;

(*****)

PROCEDURE Get_Mavrg; { kg/kmol } { Average MolWeight of Gas }

BEGIN

yb[i] := yN2 + yCH4 + yAr + yH2 + yCO2;

Mavr_g := yN2*MN2 + yCH4*MCH4 + YCO[i]*yb[i]*MCO + yAr*MAr +
yH2*MH2 + yCO2*MCO2 + yTol[i]*yb[i]*Mtol;

END;

(*****)

PROCEDURE Get_Mavrl; { kg/kmol } { Average MolWeight of Liquid }

BEGIN

Mavr_l := (XCO[i]/Lo[i])*MCuAlCl4 + XTol[i]*MTol + XCO[i]*MCO;

END;

(*****)

PROCEDURE Get_Critical; { K, bar } { Critical Temperature and pressure }

BEGIN

Tcm := TcCO*YCO[i]*yb[i] + TcCH4*yCH4 + TcN2*yN2 + TcH2*yH2 +
TcAr*yAr + TcCO2*yCO2; { K }

Pcm := PcCo*YCO[i]*yb[i] + PcCH4*yCH4 + PcN2*yN2 + PcH2*yH2 + PcAr*yAr
+ PcCO2*yCO2; { Bar }

Trm := Tg[i]/Tcm; Prm := P/Pcm;

END;

(*****)

PROCEDURE Get_Rol; { kg/m3 } { Liquid Density }

BEGIN

Ro_l := 1000*((1.5562/(Ra[i]+1) + 0.8810) - (0.0002614/(Ra[i]+1)
+ 0.0009207)*(Tl[i] - 273.15) - Lo[i]*0.06/Ra[i]);

END;

(*****)

PROCEDURE Get_Rog; { kg/ m3 } { Gas Density }

VAR V1, V2, aRK, bRK : Real;

BEGIN

Get_Critical;

{ Redlich & Kwong Equation }

aRK := (0.42748*R*R*Raise(Tcm,2.5))/Pcm;

bRK := (0.08664*R*Tcm)/Pcm;

V2 := R*Tg[i]/P;

REPEAT

V1 := V2;

V2 := R*Tg[i]/P + bRK - (aRK*(V1-bRK))/(Raise(Tg[i],0.5)*
P*V1*(V1+bRK));

UNTIL Abs(V2-V1) < 1E-6;

Ro_g := Mavr_g/V2

END;

(*****)

PROCEDURE Get_Visc_l; { mNs/M2 = cp } { Liquid Viscosity }

CONST

C2 = -5.08093E-2; D2 = 12.2693E-2;

E2 = -6.78159E-2; F2 = -9.6315E-2;

G2 = 8.026396E-2;

VAR A2, B2 : Real;

BEGIN

A2 := -4.122269 - 2.34034*EXP(C2*Raise(Ra[i],3) + D2*Raise(Ra[i],2) + E2*Ra[i]);

B2 := 1051.795 + 2026.852*EXP(F2*Raise(Ra[i],2) + G2*Ra[i]);

Visc_l := EXP(A2 + B2/Tl[i]);

END;

(*****)

PROCEDURE Get_Visc_g; { mNs/M2 = cp } { Gas Viscosity }

BEGIN

Visc_g := 0.0172 + (Tg[i] - Tg[0])*1E-4;

END;

(*****)

```

PROCEDURE Get_DgCO;  { m2/s }  { Gas Diffusivity of CO }
BEGIN
  Dg_CO := (1.013E-7*Raise(Tg[i],1.75)*
    Raise((1/MCO+1/Mavr_g),0.5))/
    (P*Raise(Raise(18.9,1/3) + Raise(10.5,1/3),2));
END;

```

(*****)

```

PROCEDURE Get_DICO;  { m2/s }  { Liquid Diffusivity of CO }
BEGIN
  DI_CO := (1.173E-13*Raise(Mavr_1,0.5)*TI[i])/
    (Visc_1*Raise(0.0307,0.6));
END;

```

(*****)

```

PROCEDURE Get_DgTol;  { m2/s }  { Gas Diffusivity of Toluene }
BEGIN
  Dg_Tol := (1.013E-7*Raise(Tg[i],1.75)*
    Raise((1/MTol+1/Mavr_g),0.5))/
    (P*Raise(Raise(18.9,1/3) + Raise(10.5,1/3),2));
END;

```

(*****)

PROCEDURE Get_CPg; { kJ/kmol*K } { Specific heat of gas }

FUNCTION Cp(a,b,c,d : Real) : Real;

BEGIN

Cp := a + b*Tg[i] + c*Raise(Tg[i],2) + d*Raise(Tg[i],3);

END;

BEGIN

CpCo := Cp(30.869,-1.285E-2,2.789E-5,-1.271E-8);

CpTol := Cp(-24.355,5.124E-1,-2.765E-4,4.911E-8);

CpH2 := Cp(27.143,9.273E-3,-1.38E-5,7.645E-9);

CpCH4 := Cp(19.251,5.212E-2,1.197E-5,-1.131E-8);

CpN2 := Cp(31.15,-1.356E-2,2.679E-5,-1.168E-8);

CpAr := Cp(20.804,-3.211E-5,5.166E-8,0);

CpB := yN2*CpN2/yb[i] + yH2*CpH2/yb[i] + yAr*CpAr/yb[i] +
yCH4*CpCH4/yb[i];

Cpg[i] := (CpB*yb[i] + YCO[i]*yb[i]*CpCO + YTol[i]*yb[i]*CpTol)*0.986;

END;

(*****)

PROCEDURE Get_Cpl; { KJ/Kmol*K } { Specific heat of liquid }

BEGIN

Cpq[i] := ((36.6809*(Ra[i]-1) + 65.2886)/(92.14*(Ra[i]-1) + 324.48) +
0.000637*(Tl[i] - 273.15) + 0.0263*Lo[i])*Mavr_l*4.183;

END;

(*****)

PROCEDURE Get_aw; { M2/M3 } { Wetted interfacial area }

VAR Lster : REAL;

BEGIN

Lster := L[i]*Mavr_l/F; { kg/s*M2 }

aw := at*(1 - EXP(-1.45*
Raise(sigc/sig_l,0.75)*
Raise(Lster/(at*Visc_l*1E-3),0.1)*
Raise(Raise(Lster,2)*at/(Raise(Ro_l,2)*9.831),-0.05)*
Raise(Raise(Lster,2)/(Ro_l*sig_l*1E-3*at),0.2)));

END;

(*****)

PROCEDURE Get_Sig; { mN/M = Dyne/cm = mJ/M2 } { Liquid Surface Tension }

BEGIN

Sig_l := 30.62 + 3.075/Ra[i] +
(-0.1103 + 0.1267/Ra[i])*(Tl[i]-273.15);

END;

(*****)

PROCEDURE Get_Labg; { kW/m*K } { Coefficient of heat conductivity }

BEGIN

Lab_g := (Visc_g*(Cpg[i]/Mavr_g + 10.4/Mavr_g))/1000;

END;

(*****)

PROCEDURE Get_kgCO; { kmol/m2*s*bar }
{ Mass-transfer coefficient CO in vapor phase }

BEGIN

kg_CO := 5.23*at*
Raise((G[i]*Mavr_g/F)/(at*Visc_g*1E-3),0.7)*
Raise(Visc_g*1E-3/(Ro_g*Dg_CO),1/3)*
Raise(at*dp,-2)*Dg_CO/(R*Tg[i]);

END;

(*****)

PROCEDURE Get_klCO; { kmol/m2*s*(kmol/m3) = m/sm-2 }
{ Mass-transfer coefficient CO in vapor phase }

BEGIN

kl_CO := 0.0051*
Raise(((L[i]*Mavr_l/F)/(aw*Visc_l*1E-3)),2/3)*
Raise(Visc_l*1E-3/(Ro_l*Dl_CO),-1/2)*
Raise(at*dp,0.4)*Raise(Ro_l/(Visc_l*1E-3*9.8319),-1/3);

END;

(*****)

PROCEDURE Get_Kog; { 1/sm-2 }
{ Overall Mass-transfer coefficient CO of gas }

BEGIN

Kog := eff/(1/kg_CO + He/(kl_CO*Enh));

END;

(*****)

```

PROCEDURE Get_kgTol; { m/sm-2 }
                { Mass-transfer coefficient Toluene in vapor phase }
BEGIN
    kg_Tol := kg_CO*Raise((Dg_Tol/Dg_CO),0.5);
END;

```

(*****)

```

PROCEDURE Get_hga; { KJ/Kmol/K } { heat transfer coefficient of gas }

BEGIN
    corr := Gb*(CpCO*(YCO[i+1] - YCO[i]) + CpTol*(YTol[i+1] - YTol[i]));
    Le := Lab_g*Mavr_g/(Ro_g*Cpg[i]*Dg_CO);
    hg_a := Kg_CO*at*Cpg[i]*Raise(Le,2/3);
    hg_a := -corr/(1 - Exp(corr/hg_a));
END;

```

(*****)

```

PROCEDURE Get_Keq; { m3/kmol } { Equilibrium constant }
BEGIN
    Keq := 74*EXP(-67*(1/Tl[i] - 3E-3));
END;

```

(*****)

```

PROCEDURE Get_Enh; { - } { Enhancement factor }
BEGIN
    Enh := 1 + (Keq*(LCu/L[i])/(1 + Keq*(YCOe[i]*yb[i]));
END;

```

(*****)

```

PROCEDURE Get_Henry; { bar*m3/kmol } { Henry's Law constant }
BEGIN
    He := 5.8e-3*(EXP(14.3686 - (530.22/(Tl[i]-13.15)))*1E-5*133.32);
END;

```

(*****)

```

PROCEDURE Get_YCOe; { - } { Equilibrium fraction of CO in gas phase }
BEGIN
    YCOe[i] := He*XCO[i]/(P*yb[i]);
END;

```

(*****)

PROCEDURE Get_YTol; { - } { Equilibrium fraction of Toluene in gas phase }

```

CONST  D6 = -0.056757;
        E6 = -0.0011087;
        F6 = 23.00;
        G6 = -7000.0;

VAR    CF, Pc, Ps : REAL;

BEGIN
  CF := EXP((D6 + E6*(Tl[i]-293.15))*Raise(Ra[i]-1,2));
  Pc := EXP(F6 + G6/Tl[i]);
  Ps := (Ptol*1E5/133.32)*(Ra[i]-CF)/Ra[i] + Pc*CF/Ra[i];
  YTol[i] := 2.15*Ps*XTol[i]*133.32*1E-5/(P*yb[i]);
END;
```

(*****)

PROCEDURE Get_Enthalpie; { KJ/Kmol }

{ Latent heat of evaporation or condensation }
 { Heat of reaction (include heat of solution) }

```

CONST
  ANTAtol = 16.0137;
  ANTBtol = 3096.52;
  ANTCtol = -53.67;

VAR  dz, hv1, hv2 : REAL;

BEGIN
  Hos := 78.3E3;
  Ptol := 133.32E-5*EXP(ANTAtol - ANTBtol/(Tl[i] + ANTCtol));
  dz := Raise(1-(Ptol/PcTol)/Raise(Tg[i]/TcTol,3),0.5);
  Hv := 8.5*ANTBtol*Raise(Tg[i],2)*dz/Raise(Tg[i] + ANTCtol,2);
END;
```

(*****)

PROCEDURE Get_RaLo; { - } { Ratio Tol/Cu & Ratio CO/Cu }

```

BEGIN
  Ra[i] := (L[i]*XTol[i])/LCu;      { Ratio Tol/Cu }
  Lo[i] := (L[i]*XCO[i])/LCu;     { Ratio CO/Cu }
END;
```

(*****)

PROCEDURE Get_FysData;

BEGIN

```

Get_RaLo;          { Ratio Tol/Cu }
                  { Ratio CO/Cu }
Get_Fract;        { Liquid fraction CO }
                  { Liquid fraction Tol }
                  { Gas fraction CO }
                  { Gas fraction Tol }
                  { Gas fraction Inert }
Get_Mavrg;        { Average MolWeight of Gas }
Get_Viscg;        { Gas Viscosity }
Get_Rog;          { Gas Density }
Get_DgCO;         { Gas Diffusivity of CO }
Get_Mavrl;        { Average MolWeight of Liquid }
Get_Viscl;        { Liquid Viscosity }
Get_Rol;          { Liquid Density }
Get_DlCO;         { Liquid Diffusivity of CO }
Get_Sigl;         { Liquid Surfase Tension }
Get_aw;           { Wetted interfacial area }
Get_Henry;        { Henry's Law constant }
Get_Keq;          { Equilibrium constant }
Get_Enthalpie;    { Latent heat of evaporation or condensation }
                  { Heat of reaction (include heat of solution) }
Get_YCOe;         { Equilibrium fraction of CO in gas phase }
Get_YTol;         { Equilibrium fraction of Toluene in gas phase }
Get_Enh;          { Enhancement factor }
Get_kgCO;         { Mass-transfer coefficient CO in vapor phase }
Get_klCO;         { Mass-transfer coefficient CO in liquid phase }
Get_Kog;          { Overall Mass-transfer coefficient CO of gas }
Get_DgTol;        { Gas Diffusivity of Toluene }
Get_kgTol;        { Mass-transfer coefficient Toluene in vapor phase }
Get_Cpg;          { Specific heat of gas }
Get_Cpl;          { Specific heat of liquid }
Get_Labg;         { Coefficient of heat conductivity }
Y[0] := YCO[i];

```

END;

(*****)

```

PROCEDURE SetValue;  { Change iteration key value }
  BEGIN
    Case Value OF

      1 : BEGIN
          IF Abs((G[N] - G[N-1])/G[N]) < epsg THEN Set_Key_Value(2) ELSE
            G[N] := (Abs(G[N-1]+G[N]))/2;
          END;
      2 : BEGIN
          IF Abs((Tg[N] - Tg[N-1])/Tg[N]) < epst THEN
            BEGIN
              IF Abs((G[N] - G[N-1])/G[N]) < epsg THEN Set_Key_Value(3) ELSE
                Set_Key_Value(1);
              END ELSE Tg[N] := (Abs(Tg[N] + Tg[N-1]))/2;
            END;
      3 : BEGIN
          IF Abs((YCO[N] - YCO[N-1])/YCO[N]) < epsy THEN
            BEGIN
              IF Abs((G[N] - G[N-1])/G[N]) < epsg THEN
                BEGIN
                  IF Abs((Tg[N] - Tg[N-1])/Tg[N]) < epst THEN EndCalculations
                    ELSE Set_Key_Value(2);
                  END ELSE Set_Key_Value(1);
                END ELSE YCO[N]:= (Abs(YCO[N] + YCO[N-1]))/2;
            END;
          END;
        END;
  END;

```

(*****)

```

PROCEDURE MasBalance;
  BEGIN

    { Inert Bal } G[i+1] := Gb*(YTol[i+1] + 1 + YCO[i+1]);
    { Flow Bal  } L[i+1] := G[i+1] + L[i] - G[i];
    { CO Bal    } XCO[i+1] := (Gb*(YCO[i+1] - YCO[i]) + L[i]*XCO[i])/L[i+1];
    { Tol Bal   } XTol[i+1] := (Gb*(YTol[i+1] - YTol[i]) + L[i]*XTol[i])/L[i+1];

  END;

```


(*****)

PROCEDURE RungeKutta;

FUNCTION FunctionRK(V : REAL) : REAL;

BEGIN

CASE a OF

1 : FunctionRK := (-Kog*at*P*F*(V-YCOe[i]))/Gb;
 2 : FunctionRK := (-kg_Tol*at*P*F*(V-YTole[i]))/Gb;
 3 : FunctionRK := (-hg_a*F*(V-Tl[i]))/(Gb*Cpg[i]/yb[i]);

END;

END; { FunctionRK }

BEGIN

CASE a OF

1..3 : BEGIN

k1 := h*FunctionRK(Y[0]); Writeln('k1 = ',k1);
 k2 := h*FunctionRK(Y[0] + k1/2); Writeln('k2 = ',k2);
 k3 := h*FunctionRK(Y[0] + k2/2); Writeln('k3 = ',k3);
 k4 := h*FunctionRK(Y[0] + k3); Writeln('k4 = ',k4); Writeln;

Y[1] := Y[0] + (1/6)*(k1 + 2*k2 + 2*k3 + k4);

END;

4 : Y[1] := (Y[0]*L[i] + (G[i+1]*Cpg[i]*Tg[i+1] - G[i]*cpg[i]*Tg[i] +
 Hos*Gb*(YCO[i+1] - YCO[i]) +
 Hv*Gb*(YTol[i+1] - YTol[i]))/Cpq[i])/L[i+1];

END;

END; {RungeKutta}

(*****)

```
PROCEDURE Equation1;  
  BEGIN  
    YCO[i+1] := Y[1];  
    a := 2;  
    Y[0] := YTol[i];  
  END;
```

(*****)

```
PROCEDURE Equation2;  
  BEGIN  
    YTol[i+1] := Y[1];  
    Get_hga;  
    a := 3;  
    Y[0] := Tg[i];  
  END;
```

(*****)

```
PROCEDURE Equation3;  
  BEGIN  
    Tg[i+1] := Y[1];  
    a := 4;  
    Y[0] := Tl[i];  
  END;
```

(*****)

```
PROCEDURE Equation4;  
  BEGIN  
    Tl[i+1] := Y[1];  
    a := 1;  
  END;
```

(*****)

PROCEDURE InitMasBal; { kmol/s }
BEGIN

{ Flow Bal } $L[0] := L[N] + G[0] - G[N];$
 { CO Bal } $XCO[0] := (L[N]*XCO[N] + Gb*(YCO[0] - YCO[N]))/L[0];$
 { Inert Bal } $YTol[N] := G[N]/Gb - YCO[N] - 1;$
 { Tol Bal } $XTol[0] := (L[N]*XTol[N] + Gb*(YTol[0] - YTol[N]))/L[0];$

 i := N;
 Get_Fract;
 Get_RaLo;
 Get_Mavrl;
 Get_Mavrg;
 Get_Cpl;
 Get_Cpg;
 Get_Enthalpie;
 Hv1 := Hv;
 i := 0;
 Get_Fract;
 Get_RaLo;
 Get_Mavrl;
 Get_Mavrg;
 Get_Cpg;
 Get_Enthalpie;
 Hv0 := Hv;
END;

(*****)

PROCEDURE InitHeatBalance;
BEGIN

 TITemp := Tg[0];
 REPEAT
 Tl[0] := (TITemp + Tl[0])/2;
 Get_Cpl;
 Get_Enthalpie;

 { Heat Bal } $TITemp := (G[0]*Cpg[0]*Tg[0] - G[N]*Cpg[N]*Tg[N] +$
 $L[N]*Cpq[N]*Tl[N] - (YTol[N]*Hv1 - YTol[0]*Hv0)*Gb -$
 $(YCO[N] - YCO[0])*Gb*Hos)/(L[0]*Cpq[0]);$

 UNTIL Abs(TITemp - Tl[0]) < 1E-3;
END;

(*****)

PROCEDURE Init;

BEGIN

F := 0.7; { m2 }
 Z := 0; { m }
 Ze := 8; { m }
 a := 1; { - }
 h := 0.04; { m }
 P := 25; { Bar }
 dp := 0.038; { m }
 sigc := 75; { mN/M }
 at := 130; { m2/m3 }

{ GasStream }

Tg[0] := 40 + 273.15;
 YTol[0] := 0; { yTol/yInert }
 GN2 := 67.23/3600; { Kmol/s }
 GCH4 := 1.22/3600; { Kmol/s }
 GH2 := 174.58/3600; { Kmol/s }
 GAr := 0.91/3600; { Kmol/s }
 GCO2 := 0;
 GCO1 := 35.37/3600; { Kmol/s }
 Gb := GN2 + GCH4 + GH2 + GAr; { Kmol/s }
 G[0] := GCO1 + Gb; { Kmol/s }
 YCO[0] := GCO1/Gb; { - }

{ LiquidStream }

XCu {Top} := 0.2774; { - }
 L[N] := 204.28/3600; { Kmol/s }
 LCu := L[N]*XCu; { Kmol/s }
 XCO[N] := 0.0028; { - }
 Tl[N] := 40 + 273.15; { K }
 XTol[N] := 1 - XCu - XCO[N]; { - }

{ Guesses }

G[N] := 240.82/3600; { = ong Gb } { Kmol/s }
 YCO[N] := 0.00024; { yCO/yInert }
 Tg[N] := 43 + 273.15; { K }
 END;

(*****)

```
PROCEDURE CalculateTower;
BEGIN
  REPEAT
    InitMasBal;
    InitHeatBalance;
    REPEAT
      Get_FysData; { at Height i in the Tower }
      RungeKutta; Equation1;
      RungeKutta; Equation2;
      MasBalance;
      RungeKutta; Equation3;
      RungeKutta; Equation4;
      i := i + 1;
      Z := Z + h;
      IF KeyPressed THEN Key;
    UNTIL (i = N-1) OR (ch = 'y');
    IF ch <> 'y' THEN
      BEGIN
        Get_Fract;
        SetValue;
      END;
    UNTIL siap;
  END; { CalculateTower }
```

(*****
*****)

```
BEGIN { HoofdProgramma }
  Init;
  REPEAT
    CalculateTower;
  UNTIL Nooit;
END. { Simulation }
```

APPENDIX 12. WORKING WITH ASPEN PLUS.

Creating an input file with ModelManager.

First you have to start ModelManager; the procedure to do this depends on the host-computer which is used. When this is done ModelManager displays a field on which the flowsheet is to be drawn; on the right of the screen a menu is displayed containing all models to be chosen from. These models are given below along with a description.

Model	Description
FEED	Incoming material-, heat- or work-stream
PROD	Outgoing material-, heat- or work-stream
MIXER	Mixes two or more streams
FSPLIT	Splits a stream into two or more of the same composition
SEP	Splits a stream into two with specified composition
SEP2	Same as SEP, but with specified T & P
HEATER	Simple heater or cooler
FLASH2	Two phase flash
FLASH3	Three phase flash
HEATX	Heatexchanger
MHEATX	Multistream heatexchanger
DSTWU	Shortcut distillation; Win-Underwood-Gilliland method
DISTL	Shortcut distillation; Edmister method
SCFRAC	Shortcut distillation for complex situation
RADFRAC	Rigorous distillation column
MULTIFRAC	Interlinked multistage fractionation units
RATEFRAC	Rate-based non-equilibrium distillation
ABSORBR	Rigorous absorber/stripper
EXTRACT	Liquid-liquid
RSTOIC	Stoichiometric reactor
RYIELD	Reactor with given yield distribution
REQUIL	Reactor with chemical/thermodynamic equilibrium
RGIBBS	Reactor which minimizes Gibbs energy

Block Unit	Description
RCSTR	Continuous stirred tank reactor
RPLUG	Plug-flow reactor
RBATCH	Batch or fed-batch reactor
PUMP	Pump
COMPR	Compressor or turbine/expander
MCOMPR	Multistage compressor or turbine
MULT	Multiplies a stream by a given factor
DUPL	Duplicates a stream any number of times
CLCHNG	Changes stream class
SSPLIT	Substream-splitter
CRUSHER	Crusher
SCREEN	Solids/solids separator
FABFL	Fabric filter
CYCLONE	Cyclone
VSCRUB	Venturi scrubber
ESP	Electrostatic precipitator
HYCYC	Hydrocyclone
CFUGE	Centrifuge filter
FILTEC	Continuous rotary vacuum filter
SWASH	Single stage solids washer
CCD	Countercurrent decanter / multistage solids washer
USER	User-defined block
USER2	Same as USER

On top of the screen there are a number of pull-down menus which represent the basic functions of Aspen; these are FILE (opening or creating files), FLOWSHEET (mainly options on how to display the flowsheet), MODIFY (renaming blocks and streams etc.), VIEW (zoom in/out on the flowsheet), FORMS (entering and reviewing all input and results), RUN (generating input files and running the simulation), PFD (generating process flow diagrams), SETTINGS (settings for the program, e.g. autosave-level).

To create a new file, open the file menu and select new. Then select the models to be used on the flowsheet from the model menu on the right of the screen, by clicking on it with the left mouse button. Put it on the right place in the flowsheet and click the left button again to place it there; interconnecting the blocks by streams is done by selecting the block (clicking on it with the left button) and then pressing the right button to display the menu of in- and outlet streams. You choose the desired stream, and click the left mouse-button on the block where the stream is to go to or come from. When all this is done, open the forms-menu and select 'next input'; Aspen then asks for all required data regarding the blocks and streams and gives a prompt when sufficient data is entered. Then open the run-menu and select 'generate' to create an Aspen input file; after this, 'submit' runs the simulation. If all results are read into ModelManager open 'forms' to display the results.

On the following pages you will find an example of in Aspen input file.

DESCRIPTION "

This is the flowsheet used by the G-group feb. 1993 to simulate the GE-Plastics CO-plant II "

DATABANKS ASPENPCD

PROP-SOURCES ASPENPCD

COMPONENTS

H2 H2 H2 /
CO CO CO /
CO2 CO2 CO2 /
N2 N2 N2 /
O2 O2 O2 /
CH4 CH4 CH4 /
C2H6 C2H6 C2H6 /
C3H8 C3H8 C3H8 /
C4H10 C4H10-1 C4H10 /
C5H12 C5H12-1 C5H12 /
C6H14 C6H14-1 C6H14 /
MEA C2H7NO MEA /
H2O H2O H2O /
AR AR AR

FLOWSHEET S

BLOCK H2-MIX IN=201 202 OUT=202B
BLOCK H2104 IN=3 202B OUT=4 203
BLOCK V2101 IN=203 OUT=142
BLOCK CO2-MIX IN=204 146 OUT=204B
BLOCK H2O-MIX IN=204B 205 150 OUT=206
BLOCK H2103 IN=2 206 OUT=3 207
BLOCK H2101 IN=207 OUT=208
BLOCK H2107 IN=208 G OUT=208B H
BLOCK H2O-MIX2 IN=208B 260 OUT=208C
BLOCK M2102 IN=208C 253 OUT=209
BLOCK H2102 IN=1 A OUT=2 B
BLOCK H2105A IN=4 C OUT=5 D
BLOCK H2105B IN=5 E OUT=6 F
BLOCK H2106 IN=6 251 OUT=7 251B
BLOCK DRUMMIX IN=B D F H OUT=STMFD
BLOCK V2103 IN=STMFD 251B OUT=STM LIQ
BLOCK SPLIT IN=LIQ OUT=A C E G
BLOCK BURNER IN=BIN OUT=1
BLOCK B1 IN=228 OUT=140 202 150

FLOWSHEET SI

BLOCK H2110A IN=209 OUT=209B
BLOCK V2107 IN=209B OUT=210 264
BLOCK H2111 IN=210 OUT=210B
BLOCK V2108 IN=210B OUT=211 265

BLOCK COND-MIX IN=264 265 OUT=266
 BLOCK V2105 IN=281 211 OUT=212 282
 BLOCK V2106 IN=285 282B OUT=213 281C
 BLOCK H2108 IN=281B OUT=281A
 BLOCK H2109A IN=281C OUT=281B
 BLOCK H2109B IN=282 OUT=282B

FLWSHEET SII

BLOCK V2116 IN=212 OUT=218 MEASUMP
 BLOCK K2104 IN=218 OUT=218B
 BLOCK H2115 IN=218B OUT=218C
 BLOCK V2117 IN=218C OUT=219 270
 BLOCK V2118 IN=219 OUT=220 OUT

FLWSHEET SIII

BLOCK H2112 IN=213 OUT=213B
 BLOCK V2109 IN=213B 284 OUT=215 285
 BLOCK VENTTANK IN=215 OUT=215B 214
 BLOCK CO-MIX IN=215B 227 OUT=216
 BLOCK K2103-1 IN=216 OUT=216B
 BLOCK H2113 IN=216B OUT=216C
 BLOCK V2113 IN=216C OUT=216D 268
 BLOCK K2103-3 IN=216D OUT=216E
 BLOCK H2114 IN=216E OUT=216F
 BLOCK V2114 IN=216F OUT=204 269
 BLOCK CONDMIX2 IN=269 268 OUT=284

FLWSHEET SIIII

BLOCK E21 IN=220 L22 L49 L48 225 OUT=L35 L2 224 228 &
 227
 BLOCK SPLIT1 IN=L35 OUT=L38 L36 L39
 BLOCK E24 IN=L38 OUT=L41
 BLOCK V41 IN=L41 OUT=L42
 BLOCK T22MIX IN=L42 L43 L11 OUT=L44
 BLOCK E22B IN=L36 OUT=L37
 BLOCK V37 IN=L37 OUT=L43
 BLOCK V39 IN=L39 OUT=L40
 BLOCK T24 IN=L40 L30 OUT=L31 L11
 BLOCK E22 IN=L2 OUT=L3
 BLOCK T21 IN=L3 LG OUT=LB L6
 BLOCK CO1 IN=LF OUT=LC LD
 BLOCK MIX1 IN=LB LE OUT=LF
 BLOCK CO2 IN=LC OUT=L4 LE
 BLOCK E23A IN=L4 OUT=L22
 BLOCK T23 IN=L28 OUT=L29 L8
 BLOCK V6 IN=L6 OUT=L28
 BLOCK V8 IN=L8 OUT=L33
 BLOCK T22 IN=L33 L46 OUT=L49 L47
 BLOCK MIX2 IN=L13 L7 L10 OUT=L48
 BLOCK V47 IN=L47 OUT=L13
 BLOCK E25-27 IN=L44 OUT=L46
 BLOCK V29 IN=L29 OUT=L7
 BLOCK V31 IN=L31 OUT=L10
 BLOCK SPLIT2 IN=LD OUT=L9 LG
 BLOCK V9 IN=L9 OUT=L30

FLWSHEET SIIIII
BLOCK K2105-1 IN=224 OUT=224A
BLOCK H2118 IN=224A OUT=224B
BLOCK K2105-2 IN=224B OUT=224C
BLOCK CO-SPLIT IN=224D OUT=226 225
BLOCK H2119 IN=224C OUT=224D
BLOCK B2 IN=142 144 OUT=143
BLOCK B3 IN=147 OUT=146
BLOCK B5 IN=143 OUT=147

PROPERTIES SYSOP3
PROPERTIES AMINES

STREAM 144
SUBSTREAM MIXED TEMP=220 PRES=23.2
MASS-FLOW H2O 450

STREAM 201
SUBSTREAM MIXED TEMP=15 PRES=21
MASS-FLOW CO 0.31 / N2 203.8 / CH4 660 / C2H6 43.4 / &
C3H8 8.3 / C4H10 4.1 / C5H12 1.44 / C6H14 2.05

STREAM 202
SUBSTREAM MIXED TEMP=120 PRES=21.8
MASS-FLOW H2 10.7 / CO 2.5 / N2 2.2

STREAM 205
SUBSTREAM MIXED TEMP=220 PRES=23.2
MASS-FLOW CO2 610 / H2O 1800

STREAM 227
SUBSTREAM MIXED TEMP=31 PRES=2.0 NPHASE=1
MASS-FLOW H2 5.8 / CO 197.3 / N2 29.4 / CH4 68.6

STREAM 251
SUBSTREAM MIXED TEMP=104 PRES=27
MASS-FLOW H2O 5400

STREAM 253
SUBSTREAM MIXED TEMP=104 PRES=27
MASS-FLOW H2O 550

STREAM 260
SUBSTREAM MIXED TEMP=208 PRES=18.3
MASS-FLOW H2O 1900

STREAM 281
SUBSTREAM MIXED TEMP=40 PRES=14.7 NPHASE=1 PHASE=L
MASS-FLOW CO2 390 / MEA 4921.1 / H2O 12915

STREAM BIN
SUBSTREAM MIXED TEMP=40 PRES=2 NPHASE=1
MASS-FLOW H2 245.6 / CO 64.5 / N2 7301.5 / O2 2284 / &
CH4 0.39 / AR 124.7

BLOCK B2 MIXER
PARAM PRES=18.5

BLOCK CO-MIX MIXER
PARAM PRES=1.5 NPHASE=1

BLOCK CO2-MIX MIXER
PARAM PRES=16.2

BLOCK COND-MIX MIXER
PARAM PRES=3.5 NPHASE=1 PHASE=L

BLOCK CONDMIX2 MIXER
PARAM PRES=1.5 NPHASE=1 PHASE=L

BLOCK DRUMMIX MIXER
PARAM PRES=0

BLOCK H2-MIX MIXER
PARAM PRES=21

BLOCK H2O-MIX2 MIXER
PARAM PRES=11

BLOCK H2O-MIX MIXER
PARAM PRES=15.6
BLOCK-OPTION FREE-WATER=NO

BLOCK M2102 MIXER
PARAM PRES=11

BLOCK MIX1 MIXER
PARAM PRES=0

BLOCK MIX2 MIXER
PARAM PRES=0

BLOCK T22MIX MIXER
PARAM PRES=0

BLOCK B1 FSPLIT
MASS-FLOW 202 10.7 1 / 150 80 2
DEF-KEY KEYNO=1 SUBSTREAM=MIXED COMPS=H2
DEF-KEY KEYNO=2 SUBSTREAM=MIXED COMPS=H2

BLOCK CO-SPLIT FSPLIT
FRAC 225 0.16347

BLOCK SPLIT FSPLIT
PARAM PRES=0
FRAC A 0.10 / C 0.10 / G 0.65

BLOCK SPLIT1 FSPLIT
FRAC L38 0.546 / L36 0.156

BLOCK SPLIT2 FSPLIT
FRAC L9 0.2564

BLOCK V2118 SEP
FRAC STREAM=220 SUBSTREAM=MIXED COMPS=H2 CO CO2 N2 O2 &
CH4 C2H6 C3H8 C4H10 C5H12 C6H14 MEA H2O AR FRACS= &
0.994 0.994 0 0.994 0 0.994 0 0 0 0 0 0 0 0 .

BLOCK VENTTANK SEP
FRAC STREAM=214 SUBSTREAM=MIXED COMPS=H2 CO CO2 N2 O2 &
CH4 C2H6 C3H8 C4H10 C5H12 C6H14 MEA H2O FRACS=0 0 &
0.04761905 0.04832633 0 0 0 0 0 0 0 0 1 0.04761950

BLOCK B5 HEATER
PARAM TEMP=500 PRES=18.2

BLOCK E22 HEATER
PARAM PRES=-0.1 DELT=-7.2

BLOCK E22B HEATER
PARAM PRES=-0.1 DELT=-5.7

BLOCK E23A HEATER
PARAM TEMP=-176.1 PRES=3.9

BLOCK E24 HEATER
PARAM PRES=-0.1 DELT=-2.5

BLOCK E25-27 HEATER
PARAM TEMP=-184.5 PRES=-0.175 NPHASE=1 PHASE=L

BLOCK H2108 HEATER

PARAM TEMP=40 PRES=14.7 NPHASE=1 PHASE=L

BLOCK H2109A HEATER
PARAM TEMP=88 PRES=0 NPHASE=1 PHASE=L

BLOCK H2109B HEATER
PARAM TEMP=104 PRES=0

BLOCK H2110A HEATER
PARAM TEMP=132 PRES=10.7

BLOCK H2111 HEATER
PARAM TEMP=40 PRES=10.4

BLOCK H2112 HEATER
PARAM TEMP=40 PRES=1.5

BLOCK H2113 HEATER
PARAM TEMP=40 PRES=-0.2

BLOCK H2114 HEATER
PARAM TEMP=40 PRES=16.2

BLOCK H2115 HEATER
PARAM TEMP=35 PRES=-0.1

BLOCK H2118 HEATER
PARAM TEMP=30 PRES=-0.2

BLOCK H2119 HEATER
PARAM TEMP=35 PRES=-0.2

BLOCK V2101 HEATER
PARAM TEMP=500 PRES=18.2
BLOCK-OPTION FREE-WATER=NO

BLOCK V29 HEATER
PARAM PRES=-3.1 DUTY=0

BLOCK V31 HEATER
PARAM PRES=-5.3 DUTY=0

BLOCK V37 HEATER
PARAM PRES=-6.6 DUTY=0

BLOCK V39 HEATER
PARAM PRES=-1.25 DUTY=0

BLOCK V41 HEATER
PARAM PRES=-6.6 DUTY=0

BLOCK V47 HEATER
PARAM PRES=-0.1 DUTY=0

BLOCK V6 HEATER
PARAM PRES=-21.83 DUTY=0

BLOCK V8 HEATER
PARAM PRES=-3.017 DUTY=0

BLOCK V9 HEATER
PARAM PRES=-19.497 DUTY=0

BLOCK CO1 FLASH2
PARAM TEMP=-171.5 PRES=27.347

BLOCK CO2 FLASH2
PARAM TEMP=-201.5 PRES=27.3

BLOCK V2103 FLASH2
PARAM TEMP=221 VFRAC=0.0421

BLOCK V2107 FLASH2
PARAM TEMP=40 PRES=10.7

BLOCK V2108 FLASH2
PARAM TEMP=40 PRES=10.4

BLOCK V2109 FLASH2
PARAM TEMP=40 PRES=1.5

BLOCK V2113 FLASH2
PARAM TEMP=40 PRES=9.6

BLOCK V2114 FLASH2
PARAM TEMP=40 PRES=16.2

BLOCK V2116 FLASH2
PARAM TEMP=40 PRES=10.3

BLOCK V2117 FLASH2
PARAM TEMP=35 PRES=28.3

BLOCK H2102 HEATX
PARAM VFRAC-COLD=0.048
FEEDS HOT=1 COLD=A
PRODUCTS HOT=2 COLD=B
HEAT-TR-COEF U=97.2 <KCAL/HR-SQM-K>

BLOCK H2103 HEATX
PARAM T-COLD=487 TYPE=COCURRENT PRES-COLD=-0.5
FEEDS HOT=2 COLD=206
PRODUCTS HOT=3 COLD=207
HEAT-TR-COEF U=63.2 <KCAL/HR-SQM-K>

BLOCK H2104 HEATX
PARAM T-COLD=550 TYPE=COCURRENT PRES-COLD=-0.3
FEEDS HOT=3 COLD=202B
PRODUCTS HOT=4 COLD=203
HEAT-TR-COEF U=54.7 <KCAL/HR-SQM-K>

BLOCK H2105A HEATX
PARAM VFRAC-COLD=0.048
FEEDS HOT=4 COLD=C
PRODUCTS HOT=5 COLD=D
HEAT-TR-COEF U=152 <KCAL/HR-SQM-K>

BLOCK H2105B HEATX
PARAM VFRAC-COLD=0.048
FEEDS HOT=5 COLD=E
PRODUCTS HOT=6 COLD=F
HEAT-TR-COEF U=250 <KCAL/HR-SQM-K>

BLOCK H2106 HEATX
PARAM T-COLD=180 PRES-COLD=-0.2
FEEDS HOT=6 COLD=251
PRODUCTS HOT=7 COLD=251B
HEAT-TR-COEF U=258 <KCAL/HR-SQM-K>

BLOCK H2107 HEATX
PARAM VFRAC-COLD=0.048
FEEDS HOT=208 COLD=G

PRODUCTS HOT=208B COLD=H
HEAT-TR-COEF U=275 <KCAL/HR-SQM-K>

BLOCK E21 MHEATX

HOT-SIDE IN=220 OUT=L2 DELT=-192.6 PRES=-0.2 MAXIT=250
COLD-SIDE IN=L22 OUT=228 PRES=-0.4 MAXIT=250
COLD-SIDE IN=L49 OUT=224 DELT=205.5 PRES=-0.2 MAXIT=250
COLD-SIDE IN=L48 OUT=227 DELT=197.9 PRES=-0.2 MAXIT=250
HOT-SIDE IN=225 OUT=L35 DELT=-190.1 PRES=-0.2 MAXIT=250

BLOCK T21 RADFRAC

PARAM NSTAGE=20
FEEDS L3 21 / LG 1 ON-STAGE
PRODUCTS LB 1 V / L6 20 L
P-SPEC 1 27.3
COL-SPECS Q1=0 QN=0 MASS-RDV=1
T-EST 1 -173.6 / 20 -172

BLOCK T22 RADFRAC

PARAM NSTAGE=12
FEEDS L33 9 / L46 1
PRODUCTS L49 1 V / L47 12 L
P-SPEC 1 2.25
COL-SPECS Q1=0 MASS-RDV=1 MOLE-BR=5.7

BLOCK T23 RADFRAC

PARAM NSTAGE=4
FEEDS L28 1
PRODUCTS L8 4 L / L29 1 V
P-SPEC 1 5.4
COL-SPECS Q1=-6000 MASS-RDV=1 MOLE-RR=6.55

BLOCK T24 RADFRAC

PARAM NSTAGE=5
FEEDS L40 6 / L30 1
PRODUCTS L11 5 L / L31 1 V
P-SPEC 1 7.6
COL-SPECS QN=0 MASS-RDV=1 MASS-RR=10.35496

BLOCK V2105 RADFRAC

PARAM NSTAGE=20 EFF=MURPHREE
FEEDS 281 1 ON-STAGE / 211 20 ON-STAGE
PRODUCTS 212 1 V / 282 20 L
P-SPEC 1 10.3 / 20 10.4
COL-SPECS Q1=0 QN=0 MASS-RDV=1
COMP-EFF 10 CO2 0.2
T-EST 1 40 / 20 72
SPEC 1 MASS-FLOW 0.5 PHASE=V COMPS=CO2 STREAMS=212
VARY 1 MURPHREE 0 0.5 STAGE1=1 STAGE2=20 COMPS=CO2
PROPERTIES AMINES

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BLOCK V2106 RADFRAC
PARAM NSTAGE=18
FEEDS 285 1 / 282B 5
PRODUCTS 213 1 V / 281C 18 L
P-SPEC 1 1.7
COL-SPECS Q1=0 QN=2.85E6 MASS-RDV=1
T-EST 16 120 / 1 102
SPEC 1 MASS-FLOW 1250 PHASE=V COMPS=CO2 STREAMS=213
VARY 1 QN 1E4 1E8
PROPERTIES AMINES

BLOCK B3 RGIBBS
PARAM TEMP=450 PRES=15

BLOCK BURNER RGIBBS

PARAM TEMP=1005 PRES=-0.5 NPHASE=1 MAXIT=100

BLOCK H2101 RGIBBS
PARAM TEMP=860 PRES=11.5 NREAC=9
STOIC 1 CH4 -1 / H2O -1 / CO 1 / H2 3
STOIC 2 H2 -1 / CO2 -1 / CO 1 / H2O 1
STOIC 3 C2H6 -1 / H2 -1 / CH4 2
STOIC 4 C3H8 -1 / H2 -2 / CH4 3
STOIC 5 C4H10 -1 / H2 -3 / CH4 4
STOIC 6 C5H12 -1 / H2 -4 / CH4 5
STOIC 7 C6H14 -1 / H2 -5 / CH4 6
STOIC 8 O2 -1 / CO -2 / CO2 2
STOIC 9 N2 -1 / C4H10 -1 / H2O -2 / MEA 2
TAPP-SPEC 1 -20 / 8 -860 / 9 2000

BLOCK K2103-1 COMPR
PARAM TYPE=POLYTROPIC PRES=9.6 TEMP=145

BLOCK K2103-3 COMPR
PARAM TYPE=POLYTROPIC PRES=15.4 TEMP=145

BLOCK K2104 COMPR
PARAM TYPE=POLYTROPIC PRES=28.5 TEMP=145

BLOCK K2105-1 COMPR
PARAM TYPE=POLYTROPIC PRES=6 TEMP=100

BLOCK K2105-2 COMPR
PARAM TYPE=POLYTROPIC PRES=9.6 TEMP=90

DESIGN-SPEC PRE
DEFINE PRE BLOCK-VAR BLOCK=B3 VARIABLE=QCALC SENTENCE=PARAM
SPEC "PRE" TO "0"
TOL-SPEC "0.1"
VARY BLOCK-VAR BLOCK=B3 VARIABLE=TEMP SENTENCE=PARAM
LIMITS "250" "800"

CONV-OPTIONS
WEGSTEIN MAXIT=150

