

UNIVERSITY OF DAYALA

COLLEGE OF ENGINEERING

CHEMICAL ENGINEERING DEPARTMENT



PRODUCTION OF AMMONIA

**A PROJECT SUBMITTED TO THE COLLEGE OF ENGINEERING
OF THE UNIVERSITY OF DIYALA IN PARTIAL FULFILLMENT OF
THE REQUIREMENT FOR THE DEGREE OF BACHLOR OF
SCIENCE IN CHEMICAL ENGINEERING**

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بِسْمِ اللَّهِ الرَّحْمَنِ الرَّحِيمِ

"يَرْفَعِ اللَّهُ الَّذِينَ آمَنُوا مِنْكُمْ وَالَّذِينَ أُوتُوا الْعِلْمَ دَرَجَاتٍ

وَاللَّهُ بِمَا تَعْمَلُونَ خَبِيرٌ "

صدق الله العظيم

سورة المجادلة الآية رقم ١١ .

الإهداء

إلى من خلقنا فسوانا الرحمن الرحيم.

إلى معلم الإنسانية ونبراس الهدى سيدنا محمد (صلى الله عليه واله وسلم).

إلى التي ترتاح لرؤيتها نفسنا وتسهل دعواتها كل أمورنا، إلى التي بسط الله الجنة تحت قدميها الوالدة الحبيبة .

إلى من نحت في صخر الحياة ليعبد لنا درب العلم والمعرفة ،الذي كلامه يذلل لنا كل مصاعب الدنيا وهمومها الوالد الحبيب .

إلى من لهما الفضل بعد الله تعالى ،إلى من ساندنا ولم يدخرا جهدا في تعليمنا أساتذتنا الأعزاء.

إلى الذين كانوا لنا خير سند وعون إخوتي وأخواتي .

إلى كل من يحب هذه الأرض الطاهرة، ارض الأنبياء والأولياء لان جذوره غرست وامتدت فيها جميع العراقيين

نهدي ثمرة جهدنا المتواضع متمنيا القبول والحمد لله رب العالمين

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I wish also express my faithful and sincere respect to my honorable supervisor Lecture .Staar Ibrahem for his continuous guidance and permanent concern during my research.

Above all .thanks to God for the mercy and blesses he showed.

SUMMARY

Ammonia is one of the most highly produced inorganic chemicals, is alkaline gas has no color. Ammonia is also produced naturally from decomposition of organic matter, including plants, animals and animal wastes.

There is more than one way for the production of ammonia. In present project, A detailed literature survey, physical and chemical properties, and other specification were addressed.

we have to choose the way Haber in the production of ammonia to be the most common way as it is much easier than other economic. The Haber Process combines nitrogen from the air with hydrogen derived mainly from natural gas (methane) into ammonia. The reaction is reversible and the production of ammonia is exothermic.

the production capacity of the unit was 10000 ton/ yr. this capacity is used as a basis in material & energy balance. Absorber column, heat exchanger & reactor were designed. the dimensions of absorber column were 5m height & 0.87 diameter. the overall coefficient based on the outside area of the tube $=891 \text{ w/m}^2 \cdot \text{c}$ Shell diameter $D_s = 796 \text{ mm}$ in heat exchanger. Volume : 27.81 m^3 , Thickness of shell: 34 mm in reactor design.

Control system for CO_2 Absorber and Methanator were also designed.

Supervisor Certification

I bear witness that this project entitled " PRODUCTION OF AMMONIA " prepared under my supervision in the Chemical Engineering Department , Faculty of Engineering , University of Diyala , by (Shatha Mohamed , Shahad Ali, Eman Mahmood) as part of the requirements for obtaining a bachelor's degree in chemical engineering.

Signature :

Name :staar Ibrahim

Date : 2016 / 5 /10

A certificate from the Commission on coal

We are seeing that we read the project entitled "production " and as a committee examining We tested students (Shatha Mohamed , Shahad Ali , Eman Mahmood) contents which saw it meet the standards of a project for a bachelor's degree (Chemical Engineering).

Signature :

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Date : 2016 / 5 /10

CONTENTS

Item

Page

ACKNOWLEDGMENT
CERTIFICATION OF COMMITTEE
SUMMARY
CONTENT

CHAPTER ONE INTRODUCTION

1.1	Introduction	1
1.2	Ammonia production	1
1.3	History	2
1.4	Uses of ammonia	2
1.5	Ammonia damage	3
1.6	Physical Properties	3
1.7	Chemical Properties	4
1.8	Methods of production	6
1.8.1	The first method	6
1.8.2	The second method	6

CHAPTER TWO MATERIAL & ENERGY BALANCE

2.1	Material balance	10
2.2	Energy balance	36

CHAPTER THREE
EQUIPMENT DESIGN

3.1 Introduction to <i>Equipment design</i>	65
3.2 Absorber design	66
3.3 heat exchanger design	68
3.4 reactor design	73

CHAPTER FOUR
PROCESS CONTROL

4.1 introduction to process control	78
4.2 Primary and Secondary Reformer Effluent	81
4.3 CO₂ Absorber and Methanator	82

CHAPTER FIVE
LOCATION, ECONOMIC & SAFETY CONSIDERATION

5.1 Location & Site Consideration	84
5.2 Safety & environmental consideration	87
5.3 Economic & Feasibility Consideration	88

Chapter one

Introduction

1.1 Introduction:

Ammonia (NH_3) is one of the most commonly produced industrial chemicals in the United States. It is used in industry and commerce, and also exists naturally in humans and in the environment. Ammonia is essential for many biological processes and serves as a precursor for amino acid and nucleotide synthesis. In the environment, ammonia is part of the nitrogen cycle and is produced in soil from bacterial processes. Ammonia is also produced naturally from decomposition of organic matter, including plants, animals and animal wastes⁽¹⁾.

Ammonia is alkaline gas has no color. Made up of one nitrogen and three parts hydrogen portion, which is lighter than air and has a characteristic pungent odor. It is the chemical symbol NH_3 and attend the distillation of coal or some nitrogenous substances. Ammonia gas does not burn in the air, but it ignites in oxygen occurs flame yellow weak Ammonia forms a minute proportion of the atmosphere; it is found in volcanic gases and as a product of decomposition of animal and vegetable matter. Because ammonia was formerly obtained by destructive distillation of horns and hooves of animals. Ammonia has also been called alkaline air and volatile alkali⁽²⁾.

1.2 Ammonia production:

facilities provide the base anhydrous liquid ammonia used predominantly in fertilizers supplying usable nitrogen for agricultural productivity. Ammonia is one of the most abundantly-produced inorganic chemicals. There are literally dozens of large-scale ammonia production plants throughout the industrial world, some of which produce as much as 2,000 to 3,000 tons per day of anhydrous ammonia in liquid form. The worldwide production in 2006 was 122,000,000 metric tons. China produced 32.0% of the worldwide production followed by India with 8.9%, Russia with 8.2%, and the United States with 6.5%. Without such massive production, our agriculturally-dependent civilization would face serious challenges .

1.3 History:

Before the start of World War I, most ammonia was obtained by the dry distillation of nitrogenous vegetable and animal products; the reduction of nitrous acid and nitrites with hydrogen; and the decomposition of ammonium salt] by alkaline hydroxides or by quicklime, the salt most generally used being the ammonium chloride (sal-ammoniac).

There are scientific discoveries play an important role in the history of the world and of human activity and led to the growth of factories and communities. Thanks to these discoveries, the global community moved from one community to another simple scientifically advanced seeks to secure human needs through the rapid and massive production. Before the birth of the twentieth century a few years, the time factor as a key factor in the manufacturing process afternoon. Then scientists began in research to devise new ways of generating managed factory manufactured products at a faster pace than those adopted. Diversified products between the cars, chemicals and furniture and clothes and others. Between discoveries which aimed to increase the production of chemicals is one of the Haber process designed by the world, "Fritz Haber".

Haber sought, then, to develop a process for the manufacture of ammonia in large quantities of agricultural and military imperatives. This seemed impossible, but he managed to achieve this dream⁽³⁾.

1.4 Uses of ammonia:.

About 80% or more of the ammonia produced is used for fertilizing agricultural crops in the form of aqua ammonia (an aqueous solution of ammonia), ammonium sulfate $(\text{NH}_4)_2\text{SO}_4$, ammonium phosphate $(\text{NH}_4)_3\text{PO}_4$, ammonium nitrate NH_4NO_3 and urea $(\text{NH}_2)_2\text{CO}$. Some anhydrous liquid ammonia is also used directly as a fertilizer⁽⁴⁾. Ammonia is also used for:

- Manufacture of nitric acid (HNO_3)
- Manufacture of nylon and other polyamides
- Refrigerant in household, commercial and industrial refrigeration systems
- Manufacture of dyes
- Manufacture of explosive
- Cleaning solutions

1.5 Ammonia damage:

can be summarized ammonia damage are as follows:

If the inhalation of this gas, it can cause severe allergic respiratory and burning eyes with a severe cough, and may lead to the closure through the air and inflammation in the lungs with hoarseness, and if it is inhaled, a center may cause suffocation and then death. When the amount of it swallow it cause burns in the mouth, esophagus, stomach, and accompanied by severe abdominal pain and difficulty in swallowing, may be exposed to infected blood, accompanied by vomiting, and may cause a puncture in the esophagus and stomach. If this gas touched the skin it causes severe and serious burns. Ammonia intervention in creams for skin care industry, but these creams cause many damages including: skin color changed to yellow, and the appearance of dark spots, and causes blisters in the skin for those who suffer from allergies, as it causes hair loss; so it is advisable to stay away from all products contain ammonia. Other damage is still under study is that the ammonia gas cause cancer and infertility, but this has not been confirmed damage, and in spite of all the damage caused by ammonia gas which, however, has beneficial uses in human life⁽⁵⁾.

1.6 Physical Properties of Ammonia:

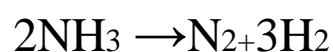
- Ammonia is a colorless gas.
- It has a pungent odor with and an alkaline or soapy taste. When inhaled suddenly, it brings tears into the eyes.
- It is lighter than air and is therefore collected by the downward displacement of air.
- It is highly soluble in water: One volume of water dissolves about 1300 volumes of ammonia gas. It is due to its high solubility in water that the gas cannot be collected over water.
- Liquid ammonia boils at 239.6 K (- 33.5°C) under one atmosphere pressure. It has a high latent heat of vaporization (1370 J per gram) and is therefore used in refrigeration plants of ice making machines.

- Liquid ammonia freezes at 195.3 K (-77.8°C) to give a white crystalline solid⁽⁶⁾.

1.7 Chemical Properties of Ammonia:

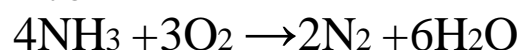
Thermal stability

Ammonia is highly stable. However, it can be decomposed into hydrogen and nitrogen by passing over heated metallic catalysts or when electric discharge is passed through it.

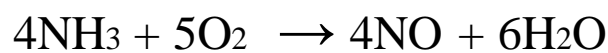


Combustibility

Ammonia is combustible in air. However, it will burn in an atmosphere of oxygen

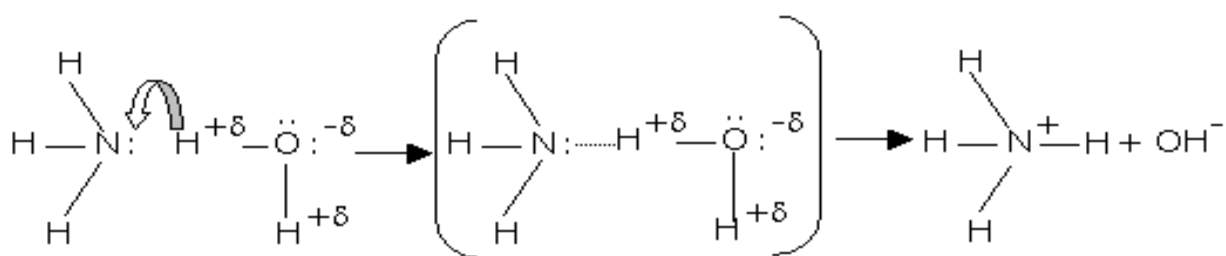


Nitric oxide is obtained when a mixture of ammonia and air is passed over platinum - rhodium catalyst at 800°C



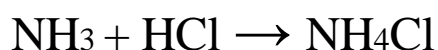
Basic character

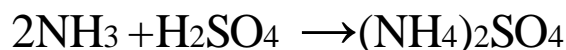
Ammonia molecule has a strong tendency to donate its lone pair of electrons of nitrogen to other molecules. Thus, it acts like a strong Lewis base. In aqueous solutions, NH_3 ionizes in accordance with the reaction.



The equilibrium constant for this reaction at 298 K is 1.8×10^{-5} . Thus, ammonia ionizes to a very small extent in aqueous solution. The aqueous solution of ammonia acts as a weak base due to the presence of OH^- ions therein. Therefore, ammonia turns red litmus blue and reacts with acids to form salts.

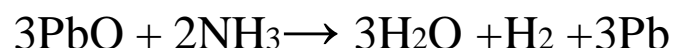
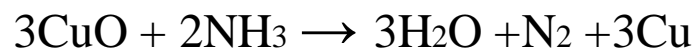
For example,





With metal oxides

Ammonia gets oxidized to nitrogen, when passed over heated metal oxides.



With halogens

Ammonia reacts with halogens but the type of halogen and reaction conditions determine the nature of products.

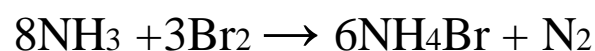
Chlorine

Nitrogen and ammonium chloride are formed with a limited amount of chlorine. In the presence of excess of chlorine, nitrogen trichloride is formed.



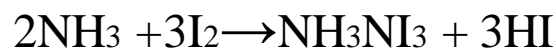
Bromine

It gives ammonium bromide and nitrogen



Iodine

When rubbed with solid iodine, a dark colored precipitate of nitrogen tri-iodide is obtained



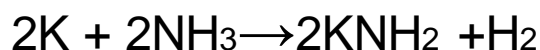
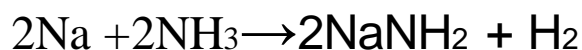
With carbon dioxide (formation of urea)

Ammonia when heated under pressure with CO_2 gives urea.

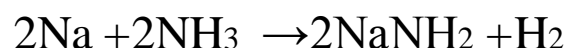


With alkali metals

When ammonia is passed over heated sodium or potassium, amides are formed and hydrogen is set free.



Alkali metal dissolved in liquid ammonia gives a blue solution, which upon standing slowly, liberates hydrogen. The blue color of such solutions is due to the presence of solvated electrons ($e^- (\text{NH}_3)_n$). For instance, with sodium⁽⁶⁾.



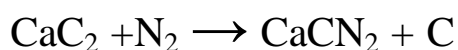
1.8 Methods production of Ammonia :

Ammonia is prepared in the industry in two ways:

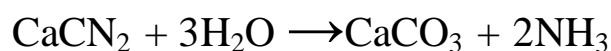
- **1.8.1 The first method (Way cyanamide) :**

Calcium cyanamide: is the inorganic compound with the formula CaCN_2 . This calcium derivative of cyanamide (CN_2^{2-}) is used as fertilizer⁽⁷⁾. First synthesized in 1898 by Adolph Frank. It is commercially known as nitrolime.

Ammonia is produced this way by reacting CaC_2 calcium carbide powder and heated to 1100°C with nitrogen gas, was made up of calcium cyanamide CaCN_2 mixed with graphite



Heating calcium cyanamide with water under high pressure, ammonia and calcium carbonate consists.

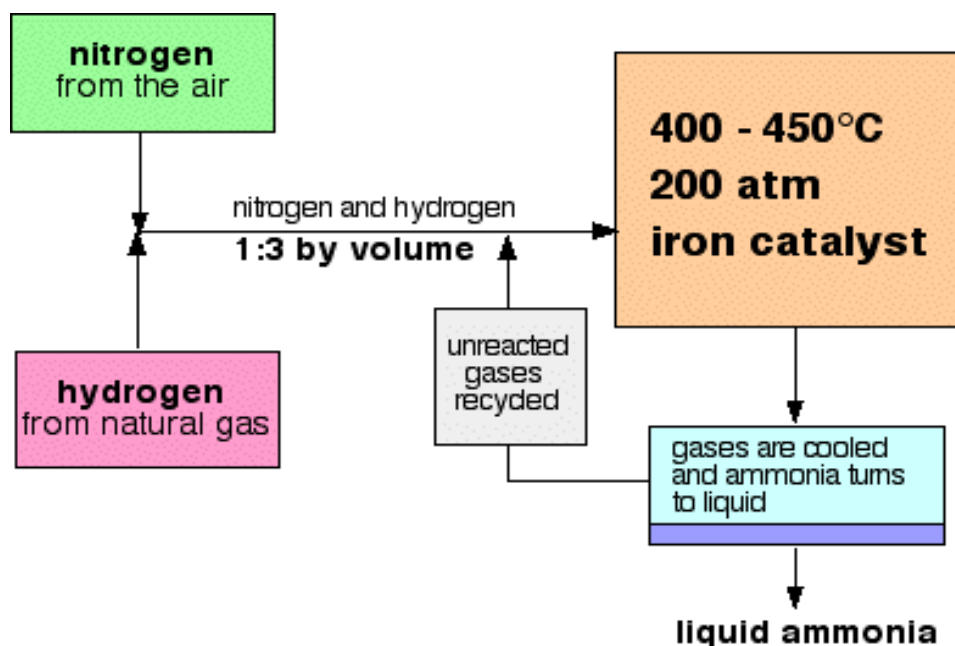


1.8.2 The second method (Haber process):

The Haber Process combines nitrogen from the air with hydrogen derived mainly from natural gas (methane) into ammonia. The reaction is reversible and the production of ammonia is exothermic .



A flow scheme for the Haber Process looks like this:



Some notes on the conditions:

The catalyst

The catalyst is actually slightly more complicated than pure iron. It has potassium hydroxide added to it as a promoter - a substance that increases its efficiency.

The pressure

The pressure varies from one manufacturing plant to another, but is always high. You can't go far wrong in an exam quoting 200 atmospheres.

Recycling

At each pass of the gases through the reactor, only about 15% of the nitrogen and hydrogen converts to ammonia. (This figure also varies from plant to plant.) By continual recycling of the unreacted nitrogen and hydrogen, the overall conversion is about 98%.

The temperature

In order to get as much ammonia as possible in the equilibrium mixture, you need as low a temperature as possible. However, 400 - 450°C isn't a low temperature!

Rate considerations

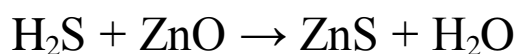
In the absence of a catalyst the reaction is so slow that virtually no reaction happens in any sensible time. The catalyst ensures that the reaction is fast enough for a dynamic equilibrium to be set up within the very short time that the gases are actually in the reactor.

Starting with a natural gas feedstock, the processes used in producing the hydrogen are:

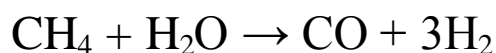
- The first step in the process is to remove sulfur compounds from the feedstock because sulfur deactivates the catalysts used in subsequent steps. Sulfur removal requires catalytic hydrogenation to convert organic sulfur compounds (RSH) in the feed stocks to gaseous hydrogen sulfide (H₂S):



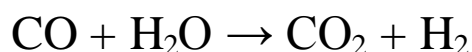
- The gaseous hydrogen sulfide is then passed through a reactor containing zinc oxide (ZnO) where it is converted to solid zinc sulfide (ZnS) in a batch process (i.e., when all of the zinc oxide has been converted to zinc sulfide, it is removed and replaced with fresh zinc oxide):



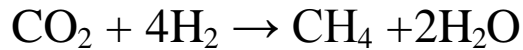
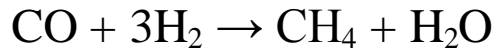
- Catalytic steam reforming of the sulfur-free methane (CH₄) feedstock is then used to form carbon monoxide (CO) plus hydrogen (H₂):



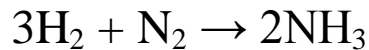
- The next step then uses catalytic shift conversion to convert the carbon monoxide to carbon dioxide (CO₂) and more hydrogen:



- The carbon dioxide is then removed either by absorption in aqueous ethanolamine solutions or by adsorption in pressure swing adsorbers (PSA) using proprietary solid adsorption media.
- The final step in producing the hydrogen is to use catalytic methanation to remove any small residual amounts of carbon monoxide or carbon dioxide from the hydrogen by converting them into methane:

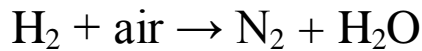


- To produce the desired end-product ammonia, the hydrogen is then catalytically reacted with nitrogen (N_2) derived from process air to form anhydrous liquid ammonia (NH_3). This step is known as the "ammonia synthesis loop" (also referred to as the Haber-Bosch process):

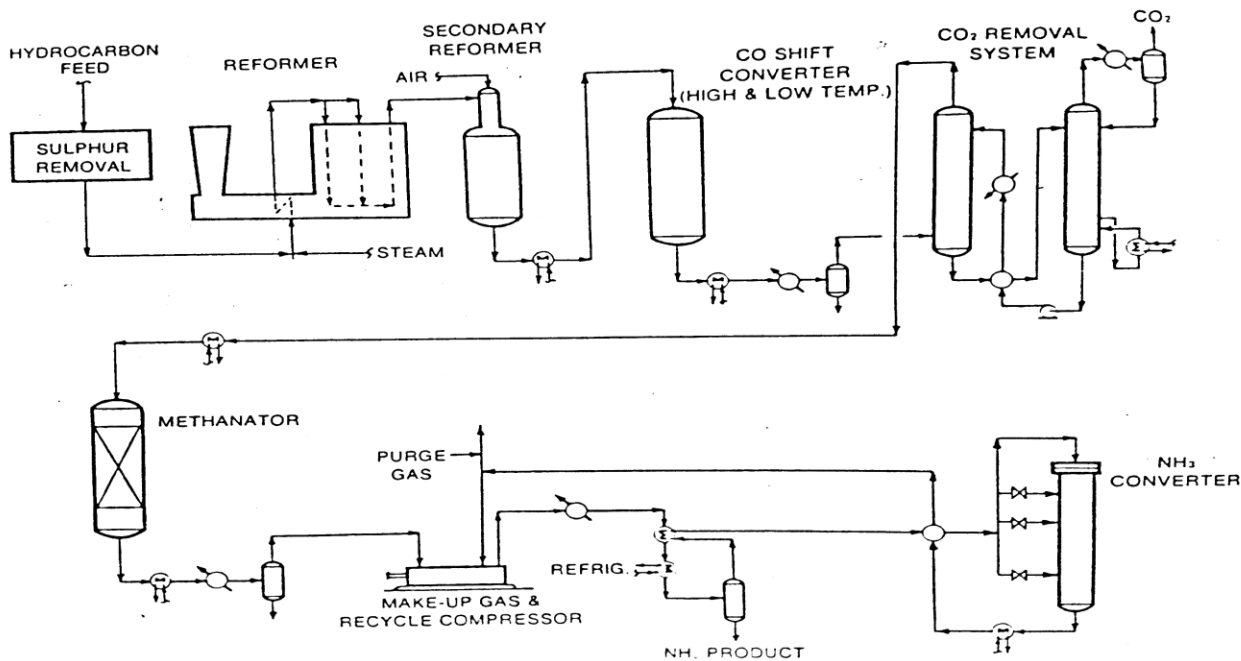


The nitrogen required for the Haber-Bosch process is obtained in either of these two ways:

- The catalytic steam reformer has two sections: a primary reformer and a secondary reformer. By the introduction of air into the secondary reformer, the air reacts with some of the hydrogen produced in the primary reformer to form nitrogen and water :



- By the low-temperature distillation of air



Chapter two

Material & energy balance

2.1 Material balance:

production capacity of $\text{NH}_3 = 10000 \text{ ton / yr}$

yr = 300 day

day = 24 hr

ton = 1000 kg

$$P = (10000 \text{ ton / yr}) * (1000 \text{ kg / ton}) * (\text{yr / 300 day}) * (\text{day / 24 hr})$$

$$P = 1389 \text{ kg /hr NH}_3$$

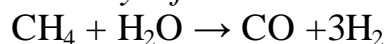
$$\text{M.W of NH}_3 = 17 \text{ kg / k mol}$$

$$P = 1389 / 17$$

$$= 81.7059 \text{ k mol /hr}$$

Reaction :

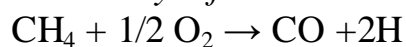
Primary reformer



conversion

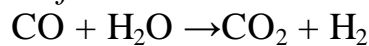
$$x_1 = 70\%$$

Secondary reformer



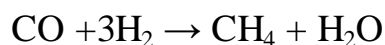
$$x_2 = 29\% \text{ }^{(8)}$$

Shift converter

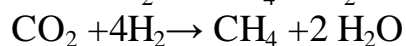


$$x_3 = 95\%$$

Methanator

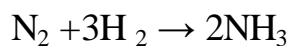


$$x_4 = 4\%$$



$$x_5 = 98\%$$

Ammonia converter



$$x_6 = 85\%$$

Let:

Methane inlet to the process = $F_1 \text{ kmol /hr}$

Conversion of CH_4 in reaction (1) = x_1

Conversion of CH_4 in reaction (2) = x_2

Conversion of CO in reaction (3) = x_3

Conversion of CO in reaction (4) = x_4

Conversion of CO_2 in reaction (5) = x_5

Conversion of H_2 in reaction (6) = x_6

H₂ produced

$$\text{Primary reformer} = 3F_1x_1 \text{ kmol/hr}$$

$$\text{Secondary reformer} = 2F_1x_2$$

$$\text{Shift converter} = (F_1x_1 + F_1x_2) x_3$$

H₂ reacted

$$\text{Methanator} = 3(F_1x_1 + F_1x_2)x_4$$

99% of CO₂ is removed in the CO₂ recovery unit

CO₂ inlet to the Methanator = CO₂ produced - CO₂ removed from *Shift* converter

$$= (F_1x_1 + F_1x_2) x_3 - 0.99(F_1x_1 + F_1x_2) x_3$$

$$= 0.01(F_1x_1 + F_1x_2) x_3 \text{ kmol/hr}$$

= CO₂ inlet to the methanator

H₂ reacted with CO₂ in methanator = 4(0.01)(F₁x₁ + F₁x₂) x₃x₅

$$= 0.04(F_1x_1 + F_1x_2) x_3x_5$$

H₂ inlet to the Ammonia converter = H₂ produced - H₂ reacted

$$= 3F_1x_1 + 2F_1x_2 + (F_1x_1 + F_1x_2) x_3$$

$$= 3(F_1x_1 + F_1x_2) x_4 - 0.04(F_1x_1 + F_1x_2) x_3x_5$$

$$= F_1[(3x_1 + 2x_2) + (x_1 + x_2) x_3 - 3(x_1 + x_2) x_4 - 0.04(x_1 + x_2) x_3x_5]$$

$$= F_1[(3*0.7 + 2*0.29) + (0.7 + 0.29)(0.95) - 3(0.7 + 0.29)(0.04) - 0.04(0.7 + 0.29)(0.95*0.98)]$$

H₂ inlet to the Ammonia converter = 3.4648324 F₁ kmol/hr

Ammonia converter:

NH₃ produced = 1389 kg/hr = 81.7059 kmol/hr

H₂ reacted = (81.7059)*(3/2) = 122.5589 kmol/hr

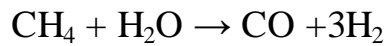
Conversion of $H_2 = \text{reacted } H_2 / \text{input } H_2$

$$0.85 = 122.5589 / 3.4648324 F_1$$

$$F_1 = 41.6144 \text{ kmol /hr } CH_4 * 16$$

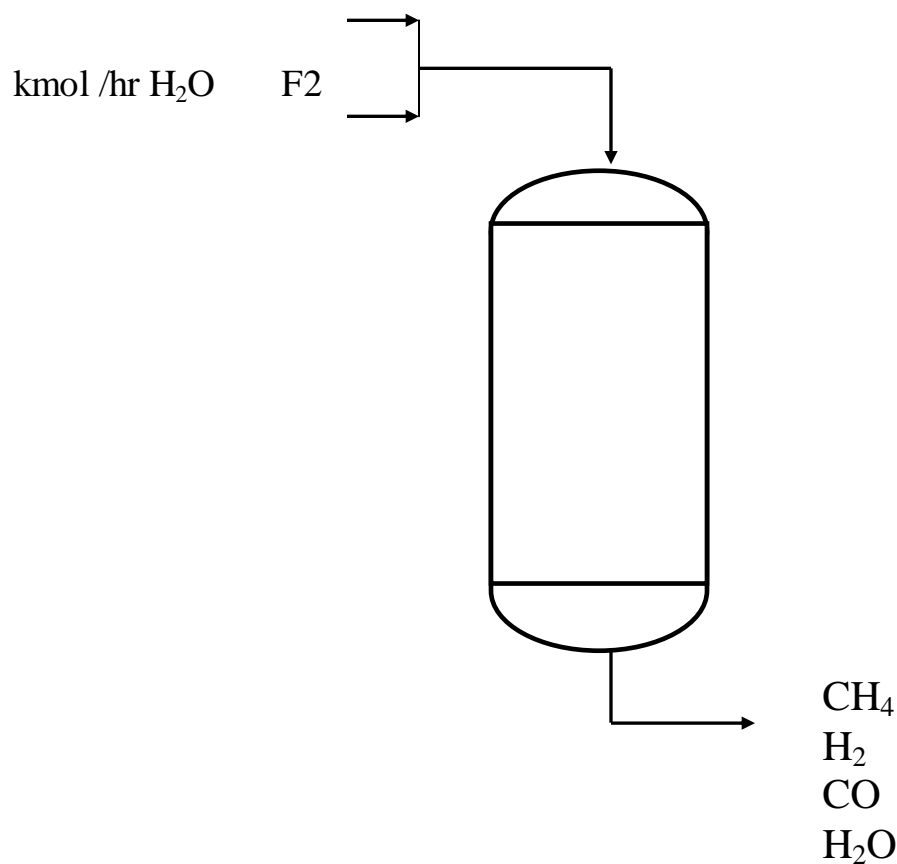
$$= 665.83 \text{ kg/hr } CH_4$$

Material balance on Primary reformer :



$$x_1 = 70\%$$

$$41.6144 \text{ kmol /hr } CH_4 \text{ } F_1$$



Steam to methane ratio = 3/1

$$F_2/F_1 = 3/1$$

$$F_2/41.6144 = 3/1$$

$$F_2 = 124.8432 \text{ kmol /hr } H_2O * 18$$

$$= 2247.18 \text{ kg/hr}$$

$$CH_4 \text{ reacted} = F_1 x_1$$

$$= 41.6144 * (0.7)$$

$$= 29.1301 \text{ kmol /hr } * 16$$

=466.08 kg/hr
 H_2O reacted= 29.1301 kmol /hr *18
 =524.34 kg/hr
 CO Produce = 29.1301 kmol /hr * 28
 =615.64 kg/hr

H_2 Produce =3*(29.1301)
 =87.3903 kmol /hr *2
 = 174.78 kg/hr

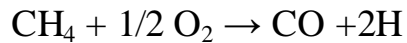
Exit product
 CH_4 = inlet – reacted
 =41.6144 -29.1301
 =12.4843 kmol /hr *16
 =199.75 kg/hr

H_2O = inlet – reacted
 =124.8432 – 29.1301
 =95.7131 kmol /hr *18
 =1722.84 kg/hr

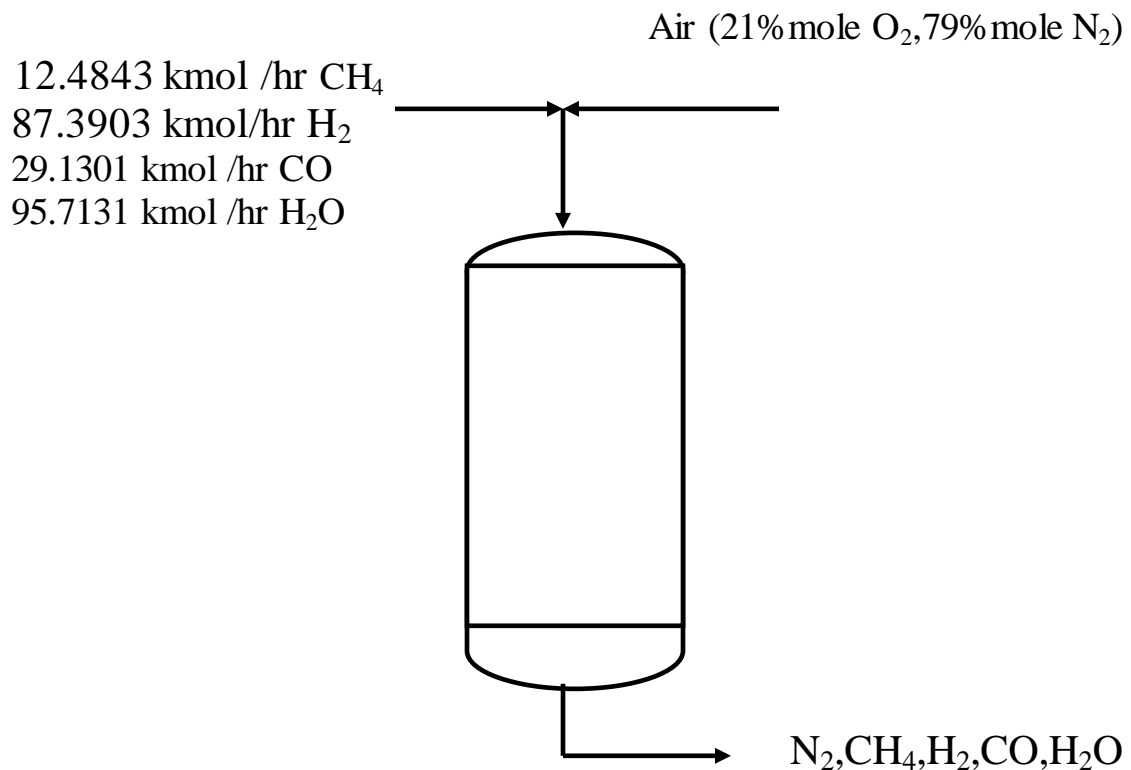
CO =29.1301 kmol /hr *28
 = 615.64 kg/hr
 H_2 = 87.3903 kmol /hr *2
 =174.78 kg/hr

Input			output	
Comp.	Kmol/hr	Kg/hr	Kmol/hr	Kg/hr
CH_4	41.6144	665.83	12.4843	199.75
H_2	-----	-----	87.3903	174.78
CO	-----	-----	29.1301	815.64
H_2O	124.8432	2247.18	95.7131	1722.84
	166.4576	2913.01	224.7648	2913.01

Material balance on secondary reformer:



$$x_2 = 29\%$$



$$\text{H}_2/\text{N}_2 = 3/1 (\text{in ammonia converter})$$

$$\text{H}_2 \text{ inlet to ammonia converter} = 3.4648324 F_1$$

$$3.4648324(41.6144) =$$

$$144.1869 \text{ kmol /hr} =$$

$$144.1869/\text{N}_2 = 3/1$$

$$\text{N}_2 = 48.0623 \text{ kmol /hr inlet to secondary reformer}$$

$$= 48.0623 * 18$$

$$1345.74 \text{ kg/hr} =$$

$$\text{O}_2 \text{ inlet to secondary reformer}$$

$$= 48.0623(0.21/0.79)$$

$$= 12.7761 \text{ kmol /hr} * 32$$

$$= 408.84 \text{ kg /hr}$$

$$\text{CH}_4 \text{ reacted} = F_1 x_2$$

$$= 41.6144 (0.29)$$

$$=12.0682 \text{ kmol /hr}$$

$$\text{O}_2 \text{ reacted} = 1/2(12.0682)$$

$$=6.0341 \text{ kmol /hr} *32$$

$$=193.09 \text{ kg /hr}$$

$$\text{CO produce} = 12.0682 \text{ kmol /hr} *23$$

$$=337.91 \text{ kg /hr}$$

$$\text{H}_2 \text{ produce} = 2*(12.0682)$$

$$=24.1364 \text{ kmol /hr} *2$$

$$=4827\text{kg /hr}$$

Exit product

$$\text{CH}_4 = \text{inlet} - \text{reacted}$$

$$=12.4843 - 12.0682$$

$$=0.4161 \text{ kmol /hr} *16$$

$$=6.66 \text{ kg /hr}$$

$$\text{CO} = \text{inlet} + \text{produced}$$

$$=29.1301 + 12.0682$$

$$=41.1983 \text{ kmol /hr} *23$$

$$=1153.55 \text{ kg /hr}$$

$$\text{H}_2 = \text{inlet} + \text{produced}$$

$$=87.3903 + 24.1364$$

$$=111.5267 \text{ kmol/hr} *2$$

$$=223.05 \text{ kg /hr}$$

$$\text{O}_2 = \text{inlet} - \text{reacted}$$

$$=12.7761 - 6.0341$$

$$=6.7420 \text{ kmol/hr} *32$$

$$=215.74 \text{ kg /hr}$$

$$\text{N}_2 \text{ in} = \text{N}_2 \text{ out}$$

$$=22.6997 \text{ kmol /hr} *28$$

$$=635.59 \text{ kg /hr}$$

$H_2O \text{ in} = H_2O \text{ out}$
 $= 95.7131 \text{ kmol/hr} * 18$
 $= 1722.8 \text{ kg/hr}$

<i>Input</i>			<i>output</i>	
<i>Comp.</i>	<i>kmol/hr</i>	<i>kg/hr</i>	<i>kmol/hr</i>	<i>kg/hr</i>
N₂	48.0623	1345.74	48.0623	1345.74
O₂	12.7761	408.84	6.7420	215.74
CH₄	12.4843	199.75	0.4161	6.66
H₂	87.3903	174.78	111.5267	223.05
CO	29.1301	815.64	41.1983	1153.55
H₂O	95.7131	1722.84	95.7131	1722.84
	285.5562	4667.59	303.6585	4667.55

Mol%
 of CH₄ dry basis $= 0.4161 / (303.6585 - 95.7131) * 100\%$
 $= 0.2\%$

Material balance on shift conversion:

48.0623Kmol/hr N₂

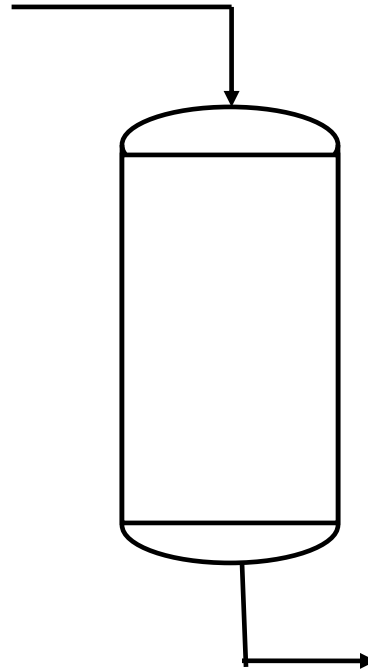
6.7420 Kmol/hr O₂

0.4161Kmol/hr CH₄

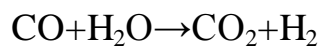
111.5267Kmol/hr H₂

41.1983Kmol/hr CO

95.7131Kmol/hr H₂O



N₂
O₂
CH₄
H₂
CO
CO₂
H₂O



X₃=95%

$$\begin{aligned} \text{Co reacted} &= 41.1983(0.95) \\ &= 39.1384 \text{Kmol/hr} * 28 \\ &= 1095.88 \text{Kg/hr} \end{aligned}$$

$$\begin{aligned} \text{H}_2\text{O reacted} &= 39.1384 \text{Kmol/hr} * 18 \\ &= 704.49 \text{kg/hr} \end{aligned}$$

$$\begin{aligned} \text{CO}_2 \text{ produced} &= 39.1384 \text{Kmol/hr} * 44 \\ &= 1722.09 \text{Kg/hr} \end{aligned}$$

$$\begin{aligned} \text{H}_2 \text{ produced} &= 39.1384 \text{Kmol/hr} * 2 \\ &= 78.27 \text{Kg/hr} \end{aligned}$$

Exit products:

$$\text{N}_2 \text{ in} = \text{N}_2 \text{ out}$$

$$= 48.0623 \text{ Kmol/hr} * 28$$

$$= 1345.74 \text{ Kg/hr}$$

$$\text{O}_2 \text{ in} = \text{O}_2 \text{ out}$$

$$= 6.7420 \text{ Kmol/hr} * 32$$

$$= 215.74 \text{ Kg/hr}$$

$$\text{CH}_4 \text{ in} = \text{CH}_4 \text{ out}$$

$$= 0.4161 \text{ Kmol/hr} * 16$$

$$= 6.66 \text{ Kg/hr}$$

$$\text{H}_2 = \text{inlet} + \text{produced}$$

$$= 111.5267 + 39.1384$$

$$= 150.6651 \text{ Kmol/hr} * 2$$

$$= 301.33 \text{ Kg/hr}$$

$$\text{CO} = \text{inlet} - \text{reacted}$$

$$= 41.1983 - 39.1384$$

$$= 2.0599 \text{ Kmol/hr} * 28$$

$$= 57.68 \text{ Kg/hr}$$

$$\text{CO}_2 = \text{inlet} + \text{produced}$$

$$= 0 + 39.1384 \text{ Kmol/h} * 44$$

$$= 1722.09 \text{ Kg/hr}$$

$$\text{H}_2\text{O in} = \text{inlet} - \text{reacted}$$

$$= 95.7131 - 39.1384$$

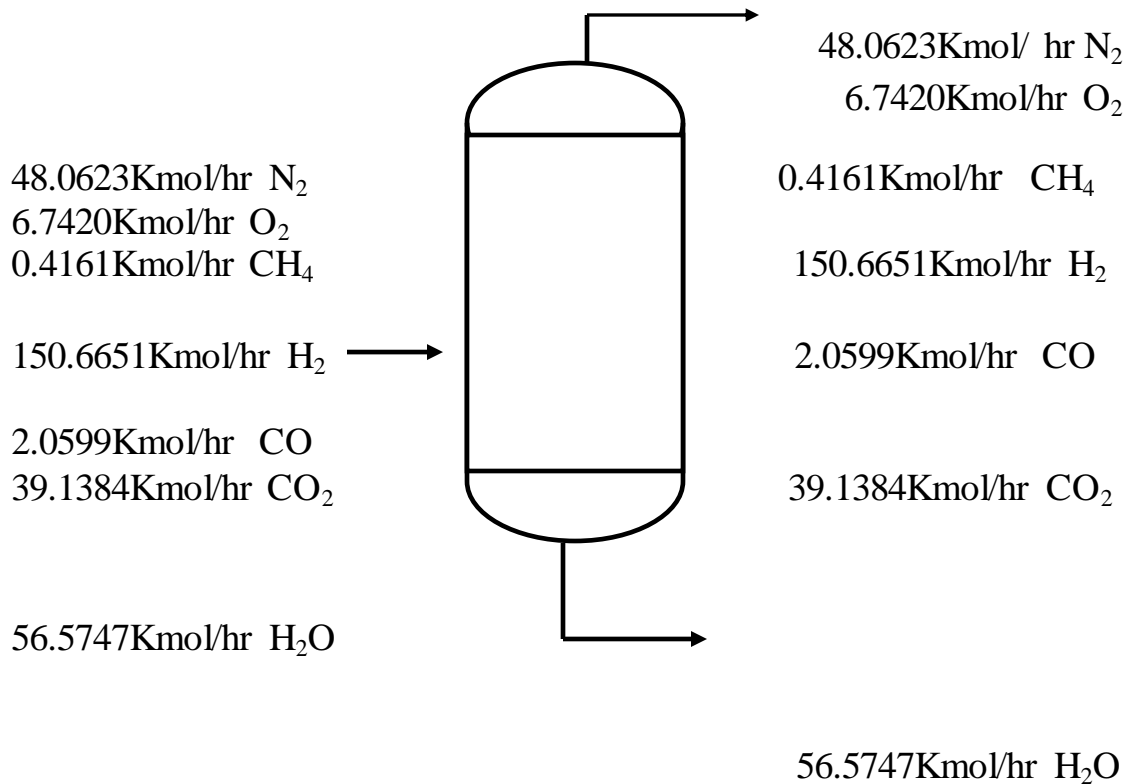
$$= 56.5747 \text{ Kmol/hr} * 18$$

$$= 1018.34 \text{ Kg/hr}$$

<i>Comp.</i>	<i>Input</i>		<i>output</i>	
	<i>kmol /hr</i>	<i>kg/hr</i>	<i>kmol /hr</i>	<i>kg/hr</i>
N₂	48.0623	1345.74	48.0623	1345.74
O₂	6.7420	215.74	6.7420	215.74
CH₄	0.4161	6.66	0.4161	6.66
H₂	111.5267	223.05	150.6651	301.33
CO	41.1983	1153.55	2.0599	57.68
CO₂	-----	-----	39.1384	1722.09

H₂O	95.7131	1722.84	56.5747	1018.34
	303.6585	4667.58	303.6585	4667.58

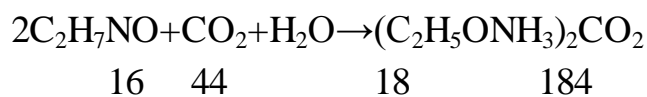
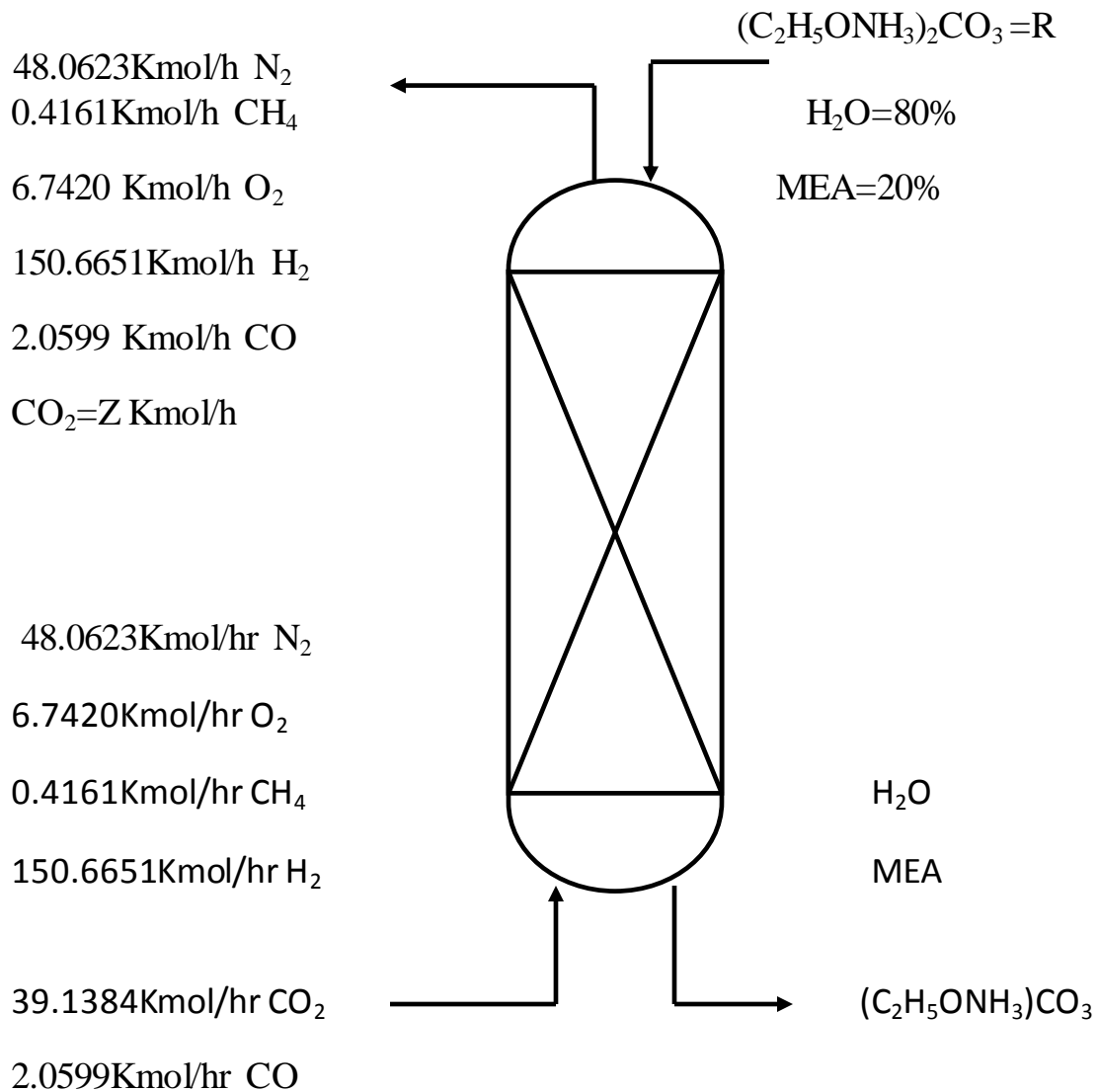
Material balance on gas liquid separator:



Material balance on absorber:

use 20 wt% MEA solution as solvent.

let recycle of $(C_2H_5ONH_3)_2CO_3=R$



1 Kmol of MEA per 0.40 Kmol of CO₂ is used

$$\begin{aligned} \text{Kmol MEA /Kmol CO}_2 &= 1/0.40 \\ \text{Kmol} / 39.1384 &= 1/0.40 \end{aligned}$$

$$\begin{aligned} \text{MEA} &= 97.8460 \text{ Kmole/hr inlet to absorber} * 61 \\ &= 5968.61 \text{ Kg/hr} \end{aligned}$$

$$\begin{aligned} \text{H}_2\text{O inlet to absorber} &= (5968.61) (0.80/0.20) \\ &= 23874.44 \text{ Kg/hr} \div 18 \\ &= 1326.3578 \text{ Kmole/hr} \end{aligned}$$

$$\begin{aligned} &99\% \text{ recovery of CO}_2 \\ \text{CO}_2 \text{ recovery} &= 0.99(39.1384) \\ &= 38.7470 \text{ Kmole/hr} \end{aligned}$$

Un reacted CO₂ exit with gases :

$$\begin{aligned} Z &= 39.1384 - 38.7470 \\ &= 0.3914 \text{ Kmole/hr} * 44 \\ &= 17.22 \text{ Kg/hr} \end{aligned}$$

$$\begin{aligned} \text{MEA reacted} &= 2(38.7470) \\ &= 77.4940 \text{ Kmole/hr} * 61 \\ &= 4727.13 \text{ Kg/hr} \end{aligned}$$

$$\begin{aligned} \text{H}_2\text{O reacted} &= 38.7470 \text{ Kmole/h} * 18 \\ &= 697.446 \text{ Kg/hr} \end{aligned}$$

$$\begin{aligned} (\text{C}_2\text{H}_5\text{ONH}_3)_2\text{CO}_3 \text{ produced} &= 38.7470 \text{ Kmole/hr} * 184 \\ &= 7129.45 \text{ Kg/hr} \end{aligned}$$

Exit liquid :

$$\begin{aligned} \text{H}_2\text{O} &= \text{inlet} - \text{reacted} \\ &= 1326.3578 - 38.7470 \\ &= 1287.6108 \text{ Kmole/hr} * 18 \\ &= 23176.99 \text{ Kg/hr} \end{aligned}$$

$$\begin{aligned} \text{MEA} &= \text{inlet} - \text{reacted} \\ &= 97.8460 - 77.4940 \\ &= 20.3520 \text{ Kmole/hr} * 16 \\ &= 1241.47 \text{ Kg/hr} \end{aligned}$$

$$\begin{aligned} (\text{C}_2\text{H}_5\text{ONH}_3)_2\text{CO}_3 &= \text{inlet} + \text{produced} \\ &= R + 38.7470 \text{ Kmole/hr} * 184 \\ &= 184R + 7129.45 \text{ Kg/hr} \end{aligned}$$

<i>Input</i>			<i>output</i>	
<i>Comp.</i>	<i>kmol /hr</i>	<i>kg/hr</i>	<i>kmol /hr</i>	<i>kg/hr</i>
N₂	48.0623	1345.74	48.0623	1345.74
O₂	6.7420	215.74	6.7420	215.74
CH₄	0.4161	6.66	0.4161	6.66
H₂	150.6651	301.33	150.6651	301.33
CO	2.0599	57.68	2.0599	57.68
CO₂	39.1384	1722.09	0.3914	17.22
H₂O	1326.3578	23874.44	1287.6108	23176.9
MEA	97.846	5968.61	20.3520	1241.4
(C₂H₅ONH₃)₂CO₃	R	184R	R+38.7470	184R+7129.45
		33492.29+184R		33492.28+184R

Material balance on stripper:

1287.6108Kmol/hr H₂O

20.3520Kmol/h MEA

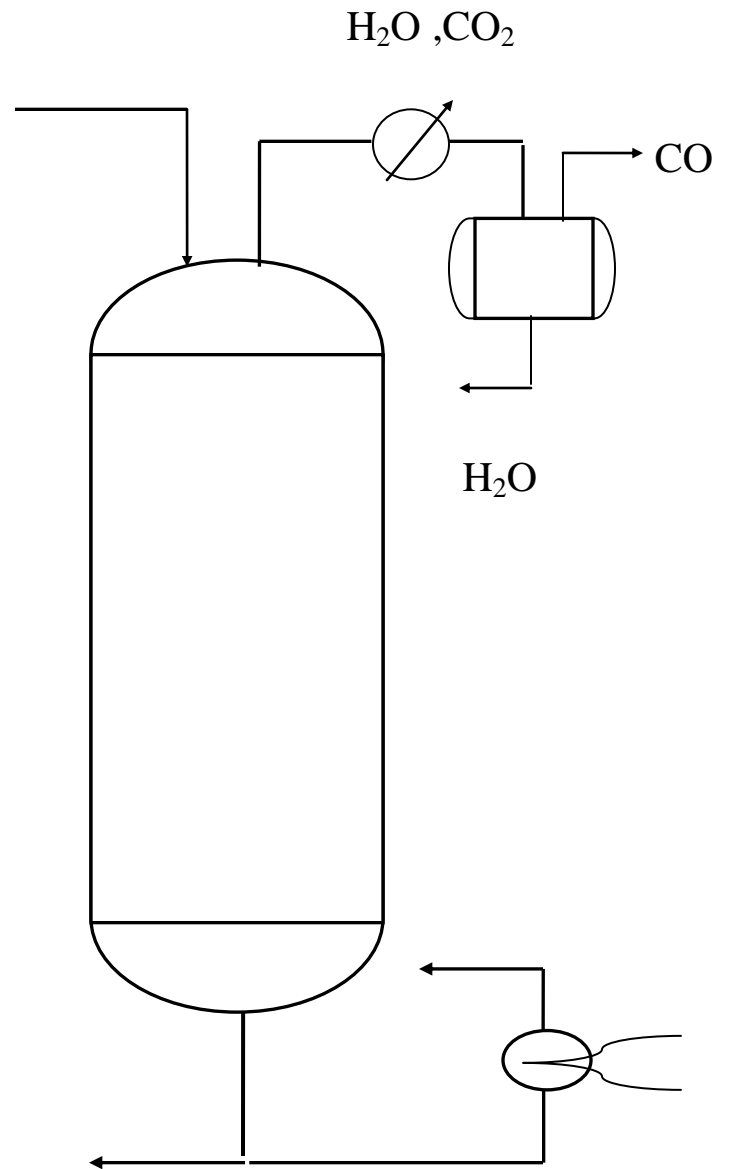
R+38.7470Kmol/hr

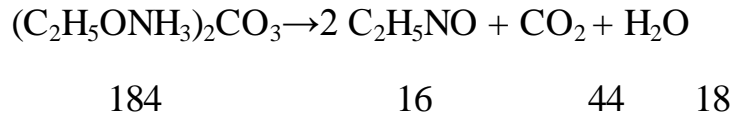
(C₂H₅ONH₃)₂CO₃

H₂O

MEA

(C₂H₅ONH₃)₂CO₃





$$\begin{aligned}
 \text{CO}_2 \text{ produced} &= 38.7470 \text{ Kmol/hr} * 44 \\
 &= 1704.87 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{H}_2\text{O produced} &= 38.7470 \text{ Kmol/hr} * 18 \\
 &= 697.45 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{MEA produced} &= 2(38.7470) \\
 &= 77.4940 \text{ Kmol/hr} * 61 \\
 &= 4727.13 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 (\text{C}_2\text{H}_5\text{ONH}_3)_2\text{CO}_2 \text{ reacted} &= 38.7470 \text{ Kmol/hr} * 184 \\
 &= 7129.45 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 &99\% \text{ recovery of } (\text{C}_2\text{H}_5\text{ONH}_3)_2\text{CO}_3 \\
 &0.99 (\text{R} + 38.7470) = 38.7470 \\
 &\text{R} = 0.3914 \text{ Kmol/hr } (\text{C}_2\text{H}_5\text{ONH}_3)_2\text{CO}_3 * 184 \\
 &= 72.02 \text{ Kg/hr}
 \end{aligned}$$

Exit liquid :

$$\begin{aligned}
 \text{H}_2\text{O} &= \text{inlet} + \text{produced} \\
 &= 1287.6108 + 38.7470 \\
 &= 1326.3578 \text{ Kmol/hr} * 18 \\
 &= 23874.44 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{MEA} &= \text{inlet} + \text{produced} \\
 &= 20.3520 + 77.4940 \\
 &= 97.8460 \text{ Kmol/hr} * 61 \\
 &= 5968.61 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 (\text{C}_2\text{H}_5\text{ONH}_3)_2\text{CO}_3 &= \text{inlet} - \text{reacted} \\
 &= (0.3914 + 38.7470) - 38.7470 \\
 &= 0.3914 \text{ Kmol/hr} * 184 \\
 &= 72.02 \text{ Kg/hr}
 \end{aligned}$$

<i>Input</i>			<i>Output</i>	
<i>Comp.</i>	<i>kmol /hr</i>	<i>kg/hr</i>	<i>kmol /hr</i>	<i>kg/hr</i>
CO₂	-----	-----	38.7470	1704.87
H₂O	1287.6108	23176.99	1326.357	23874.44
MEA	20.3520	1241.47	97.8460	5968.61
(C₂H₅ONH₃)₂CO₃	39.1384	7201.47	0.391	72.02
		31619.93		31619.93

Material balance on methanator :

48.0623Kmol/h N₂

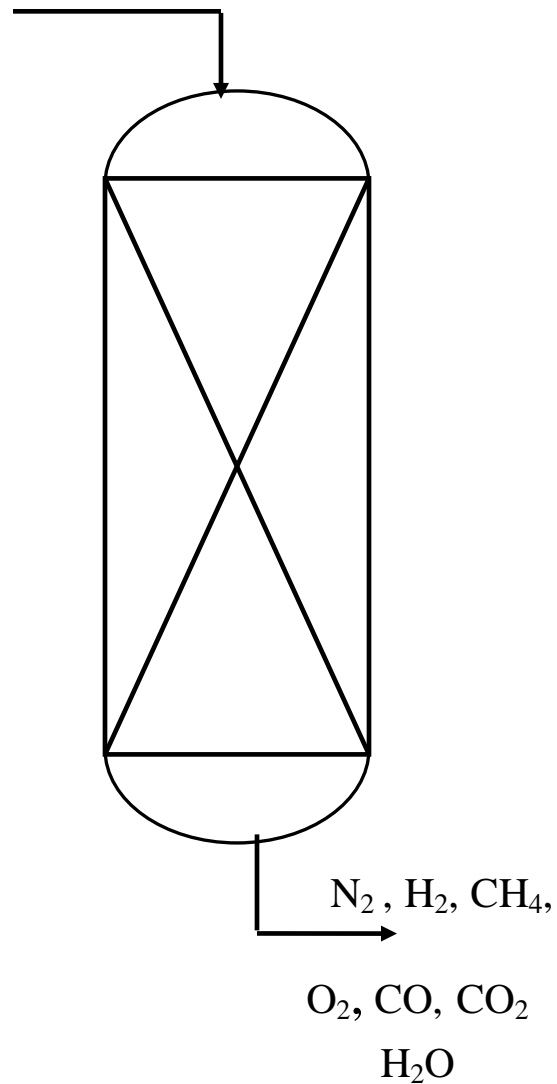
6.7420Kmol/h O₂

0.4161Kmol/h CH₄

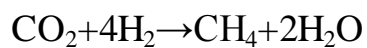
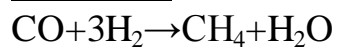
150.6651Kmol/h H₂

2.0599Kmol/h CO

0.3914 Kmol/h CO₂



Reaction 4



$$\begin{aligned} \text{Co reacted} &= 41.1983 \times 4 \\ &= 41.1983 \text{ (o.o4)} \\ &= 1.6479 \text{ Kmol/hr} \times 28 = 46.14 \text{ Kg/hr} \end{aligned}$$

$$\begin{aligned} \text{H}_2 \text{ reacted} &= 3(1.6479) \\ &= 4.9437 \text{ Kmol/hr} \times 2 \\ &= 9.89 \text{ Kg/hr} \end{aligned}$$

$$\text{CH}_4 \text{ produced} = 1.6479 \text{ Kmol/hr} \times 16$$

$$\begin{aligned}
 &= 26.37 \text{ Kg/hr} \\
 \text{H}_2\text{O produced} &= 1.6479 \text{ Kmol/hr} * 44 \\
 &= 29.66 \text{ Kg/hr}
 \end{aligned}$$

Reaction 5

$$\begin{aligned}
 \text{CO}_2 \text{ reacted} &= 0.3914 \times 5 \\
 &= 0.3914 (0.98) \\
 &= 0.3836 \text{ Kmol/hr} * 44 \\
 &= 16.88 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{H}_2 \text{ reacted} &= 4(0.3836) \\
 &= 1.5344 \text{ Kmol/hr} * 2 \\
 &= 3.07 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{CH}_4 \text{ produced} &= 0.3836 \text{ Kmol/hr} * 16 \\
 &= 6.14 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{H}_2\text{O produced} &= 2(0.3836) \\
 &= 0.7672 \text{ Kmol/hr} * 18 \\
 &= 13.81 \text{ Kg/hr}
 \end{aligned}$$

Exit Products

$$\begin{aligned}
 \text{N}_2 &= 48.0623 \text{ Kmol/hr} \\
 &= 1345.74 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{O}_2 &= 6.7420 \text{ Kmol/hr} \\
 &= 215.74 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{CH}_4 &= \text{inlet} + \text{produced} \\
 &= 0.4161 + (1.6479 + 0.3836) \\
 &= 2.4476 \text{ Kmol/hr} * 16 \\
 &= 39.16 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{H}_2 &= \text{inlet} - \text{reacted} \\
 &= 150.6651 - (4.9437 + 1.5344) \\
 &= 144.1870 \text{ Kmol/hr} * 2 \\
 &= 288.37 \text{ Kg/hr}
 \end{aligned}$$

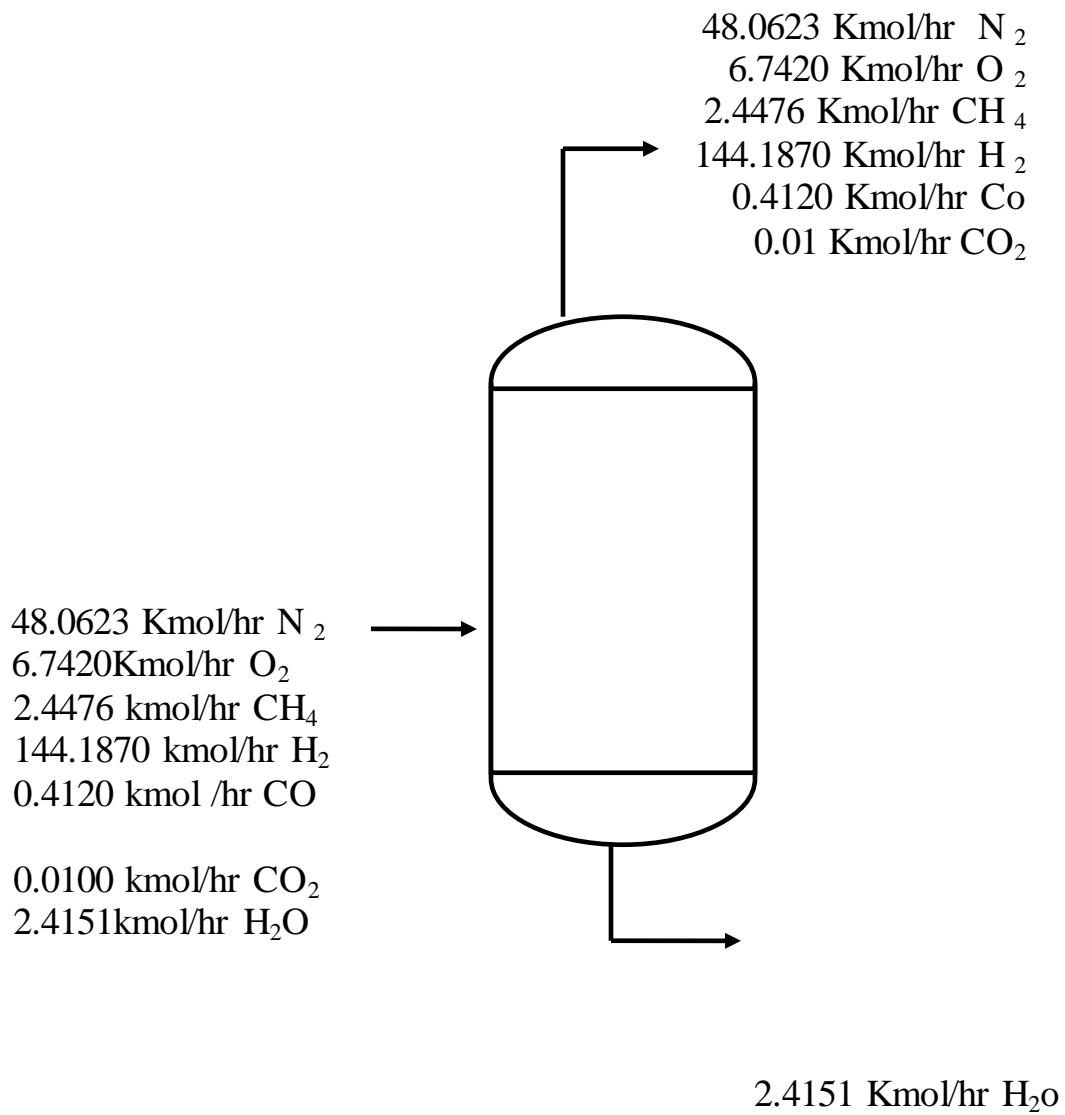
$$\begin{aligned}
 \text{CO} &= \text{inlet} - \text{reacted} \\
 &= 2.0599 - 1.6479 \\
 &= 0.4120 \text{ Kmol/hr} * 28 \\
 &= 11.54 \text{ Kg/hr}
 \end{aligned}$$

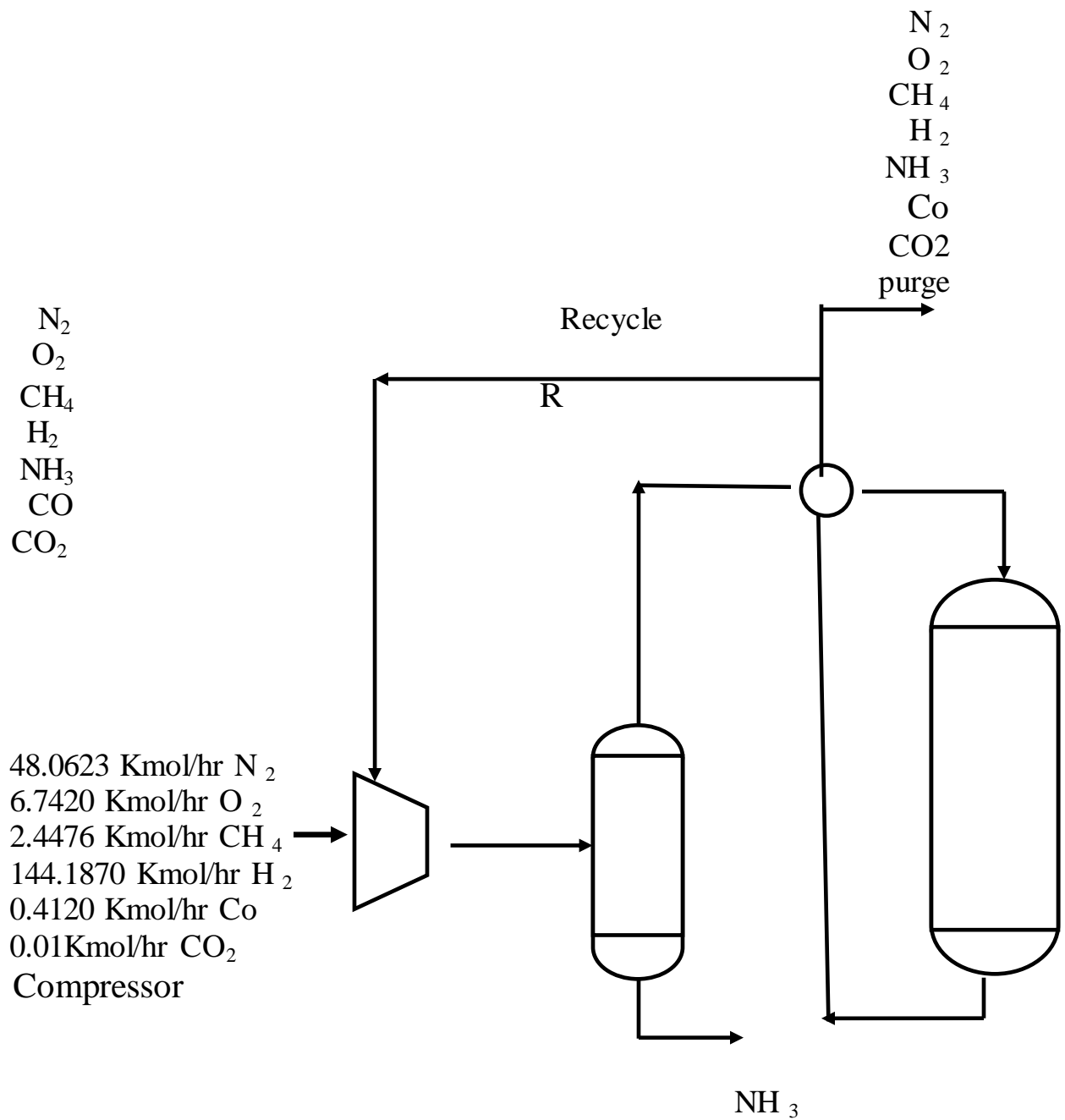
$$\begin{aligned}
 \text{CO}_2 &= \text{inlet} - \text{reacted} \\
 &= 0.3914 - 0.3814 \\
 &= 0.0100 \text{ Kmol/hr} * 44 \\
 &= 0.44 \text{ Kg/hr}
 \end{aligned}$$

$$\begin{aligned}
 \text{H}_2\text{O} &= \text{inlet} + \text{produced} \\
 &= 0 + (1.6479 + 0.7672) \\
 &= 2.4151 \text{ Kmol/hr} \cdot 18 \\
 &= 43.47 \text{ Kg/hr}
 \end{aligned}$$

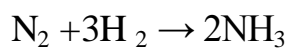
input			Output	
Comp.	Kmol/hr	Kg/hr	Kmol/hr	Kg/hr
N₂	48.0623	1345.74	48.0623	1345.74
O₂	6.7420	215.74	6.7420	215.74
CH₄	0.4161	6.66	2.4476	39.16
H₂	150.6651	301.33	144.1870	288.37
Co	2.0599	57.68	0.4120	11.54
Co₂	0.3914	17.22	0.0100	0.44
H₂o	-----	-----	2.4151	43.47
		<i>1944.37</i>		<i>1944.46</i>

Material balance on gas liquid separator 2:





REACTION



$$x_6 = 85\%$$

Overall conversion of $H_2 = 85\%$

$$\begin{aligned}
 H_2 \text{ reacted} &= 144.1870 \times 0.85 \\
 &= 122.55895
 \end{aligned}$$

$$\begin{aligned}
&= 122.5589 \text{ Kmol/hr} \cdot 2 \\
&= 245.12 \text{ Kg/hr} \\
\text{N}_2 \text{ reacted} &= 1/3(122.5589) \\
&= 40.8530 \text{ Kmol/hr} \cdot 28 \\
&= 1143.88 \text{ Kg/hr} \\
\text{NH}_3 \text{ produced} &= 2/3 (122.5589) \\
&= 81.7059 \text{ Kmol/hr} \cdot 17 \\
&= 1389 \text{ Kg/hr}
\end{aligned}$$

PURGE

$$\begin{aligned}
\text{N}_2 &= \text{inlet} - \text{reacted} \\
&= 48.0623 - 40.8530 \\
&= 7.2093 \text{ Kmol/hr} \\
\text{O}_2 &= 6.7420 \text{ Kmol/hr} \quad (\text{inert}) \\
\text{CH}_4 &= 2.4476 \text{ Kmol/hr} \quad (\text{inert}) \\
\text{H}_2 &= \text{inlet} - \text{reaction} \\
&= 144.1870 - 122.5589 \\
&= 21.6281 \text{ Kmol/hr} \\
\text{Co} &= 0.4120 \text{ Kmol/hr} \quad (\text{inert}) \\
\text{Co}_2 &= 0.0100 \text{ Kmol/hr} \quad (\text{inert}) \\
\text{NH}_3 &= e \text{ Kmol/hr}
\end{aligned}$$

Single pass conversion of $\text{H}_2 = 20\%$

Single pass conversion = $\text{H}_2 \text{ reacted} / \text{H}_2 \text{ input to reactor}$

Let H_2 in recycle stream = r Kmol/hr

H_2 input to reactor = $144.1870 + r$ Kmol/hr

H_2 reacted = 122.5589 Kmol/hr

$$0.2 = 122.5589 / (144.1870 + r)$$

$$r = 468.6075 \text{ Kmol/hr H}_2$$

Recycle to purge ratio R/B

$$R/B = 468.6075 / 21.6281 = 21.6666$$

Recycle stream

$$\begin{aligned}
\text{N}_2 &= 21.6666 (7.2093) \\
&= 156.2010 \text{ Kmol/hr} \\
\text{O}_2 &= 21.6666 (6.7420) \\
&= 146.0762 \text{ Kmol/hr} \\
\text{CH}_4 &= 21.6666 (2.4476) \\
&= 53.0312 \text{ Kmol/h}
\end{aligned}$$

$$H_2 = r = 468.6075 \text{ Kmol/hr}$$

$$C_o = 21.6666 (0.4120)$$

$$= 8.9266 \text{ Kmol/hr}$$

$$C_{o_2} = 21.6666 (0.0100)$$

$$= 0.2167 \text{ Kmol/hr}$$

$$NH_3 \text{ produced from reactor} = 81.7059 \text{ Kmol/hr}$$

$$NH_3 \text{ in recycle stream} = 81.7059 * 21.6666 / (21.6666 + r)$$

$$= 78.1012 \text{ Kmol/hr}$$

$$NH_3 \text{ in purge stream } e = 81.7059 - 78.1012$$

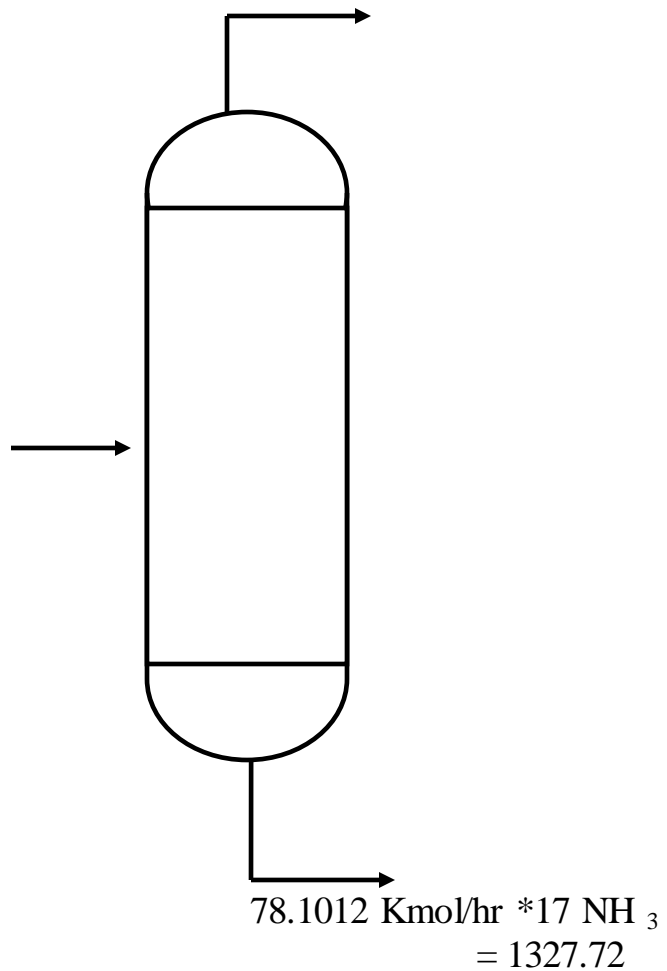
$$= 3.6047 \text{ Kmol/hr}$$

Material balance on separator :

To reactor
204.2633b Kmol/hr N₂
152.8182 Kmol/hr O₂
55.4788 Kmol/hr CH₄
612.7945 Kmol/hr H₂
9.3386 Kmol/hr Co
0.2267 Kmol/hr Co₂

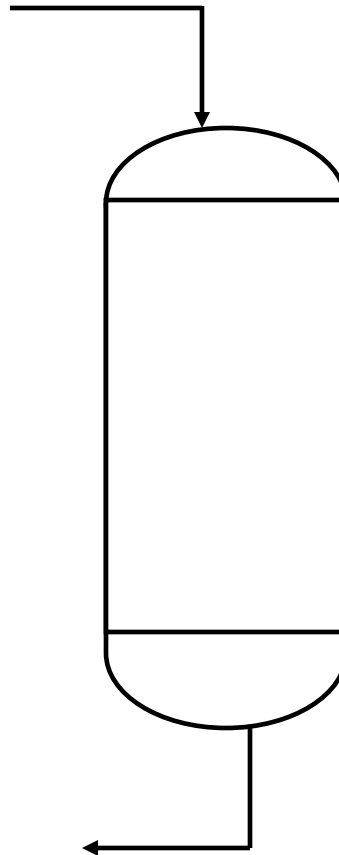
Fresh feed + recycle
204.2633 kmol/hr N₂

152.8182 kmol/hr O₂
55.4788 kmol/hr CH₄
612.7945 kmol/hr H₂
9.3386 kmol/hr Co
0.2267 kmol/hr Co₂
78.1012 kmol/hr NH₃



Material balance on reactor:
All stream are calculated

204.2633 Kmol/hr N₂
152.8182 Kmol/hr O₂
55.4788 Kmol/hr CH₄
612.7945 Kmol/hr H₂
9.3386 Kmol/hr Co
0.2267 Kmol/hr Co₂



163.4103 Kmol/hr N₂
152.8182 Kmol/hr O₂
55.4788Kmol/hr CH₄
490.2356 Kmol/hr H₂
9.3386 Kmol/hr Co
0.2267 Kmol/hr Co₂
81.7059 Kmol/hr NH₃

Input			output	
Comp.	Kmol/hr	Kg/hr	Kmol/hr	Kg/hr
N₂	204.2633	5719.37	163.4103	4575.49
O₂	152.8182	4890.18	152.8182	4890.18
CH₄	55.4788	887.66	55.4788	887.66
H₂	612.7945	1225.59	490.2356	980.47
Co	9.3386	261.48	9.3386	261.48
Co₂	0.2267	9.97	0.2267	9.97
NH₃	-----	-----	81.7059	1389
		12994.25		122994.25

2.2 Energy Balance:

Gas heat capacity :

$$C_p = a + bT + cT^2 + dT^3$$

Where : C_p Gas heat capacity KJ/K mol .k

T: temperature k

a, b , c and d constants

component	A	B	c	d
N ₂	31.150	-1.356*10 ⁻²	2.679*10 ⁻⁵	-1.168*10 ⁻⁸
O ₂	28.106	-3.680*10 ⁻⁶	1.745*10 ⁻⁵	-1.065*10 ⁻⁸
CH ₄	19.251	5.212*10 ⁻²	1.197*10 ⁻⁵	-1.131*10 ⁻⁸
H ₂	27.143	9.273*10 ⁻³	-1.38*10 ⁻⁵	7.645*10 ⁻⁹
CO	30.869	-1.285*10 ⁻²	2.789*10 ⁻⁵	-1.271*10 ⁻⁸
CO ₂	19.795	7.343*10 ⁻²	-5.6*10 ⁻⁵	1.715*10 ⁻⁸
NH ₃	27.315	2.383*10 ⁻²	1.707*10 ⁻⁵	-1.184*10 ⁻⁸
H ₂ O	32.243	1.923*10 ⁻³	1.055*10 ⁻⁵	-3.596*10 ⁻⁹

component	ΔH_f ₂₉₈ KJ/mol
CH ₄	-74.86
CO	-110.62
CO ₂	-393.77
NH ₃	-45.72
H ₂ O	-242

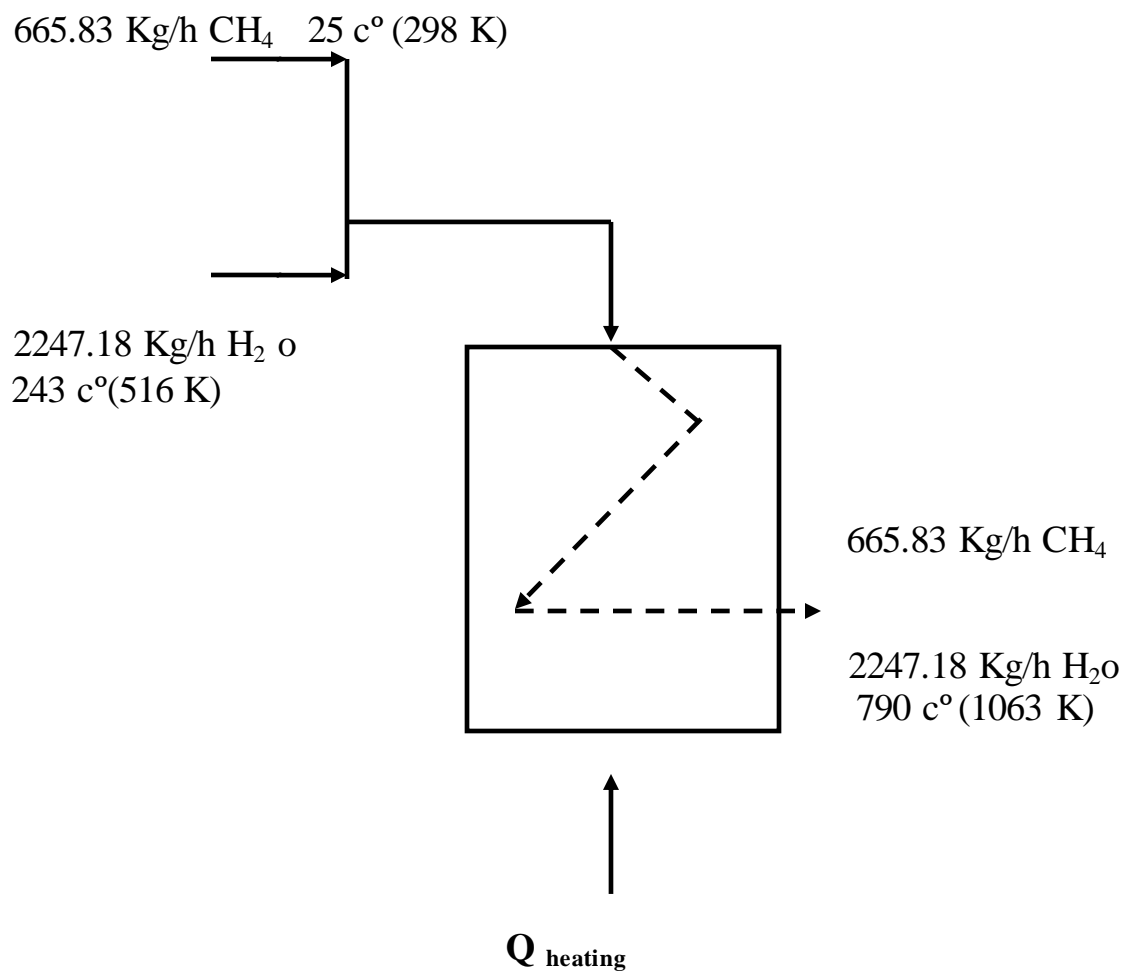
Energy balance furnace

Methane at 25 °C & 35 Kg/cm² (35 bar)
saturated steam at (35bar)

(243 °C) from steam table.

The two feeds are heated to 790 °C .

(Reaction temperature of primary reformer)



Energy balance

$$Q_{in} + Q_{heating} = Q_{out}$$

Q_{in} = inlet enthalpy KJ/h

Q_{out} = outlet enthalpy KJ/h

$Q_{heating}$ = heat supplied KJ/h

$$Q = n \Delta H$$

$$\Delta H = \int_{T_{ref}}^T C_p dt$$

$$Q = n \int_{T_{ref}}^T C_p dt$$

n = Molar flow rate Kmole/h

C_p = gas heat capacity

dT = temperature difference

$$Q = \text{Kmol/h} * \text{KJ/Kmol} * k = \text{KJ/h}$$

$$C_p = a + bT + cT^2 + dT^3$$

$$\int_{T_{ref}}^T C_p dt = \int_{T_{ref}}^T (a + bT + cT^2 + dT^3) dT$$

$$= [a(T - T_{ref}) + b/2(T - T_{ref})^2 + c/3(T - T_{ref})^3 + d/4(T - T_{ref})^4]$$

$$T_{ref} = 25 \text{ } ^\circ\text{C} = 298$$

$$Q_{in} = n \int_{298}^{298} C_p dT + n \int_{298}^{516} C_p dT$$

For CH₄

$$\int_{298}^{516} C_p dt = [32.243 (516 - 298) + 1.923 * 10^{-3} * (516^2 - 298^2) / 2 + 1.055 * 10^{-5} * (516^3 - 298^3) / 3 - 3.596 * 10^{-3} * (516^4 - 298^4) / 4]$$

$$= 7531 \text{ KJ/Kmol}$$

$$n = M/M_w$$

$$= 2247.18 / 18 = 124.8433 \text{ Kmole/h}$$

$$n = \int_{298}^{516} C_p dT = (124.8433) * (7531) \\ = 940195 \text{ KJ/h}$$

$$Q_{in} = 0 + 940195$$

$$Q_{in} = 940195 \text{ KJ/h}$$

$$Q_{\text{out}} = \sum n_i \int_{298}^{1063} C_{pi} dT$$

Component	M Kg/h	n Km ³ /h	$\int_{298}^{1063} C_{pi} dT$	n $\int_{298}^{1063} C_{pi} dT$
CH ₄	665.83	41.6144	42959	1787713
H ₂ O	2247.18	124.8433	28657	3577634

$$Q_{\text{out}} = \sum n_i \int_{298}^{1063} C_{pi} dT$$

$$= 5365347 \text{ KJ/h}$$

$$940195 + Q_{\text{heating}} = 5365347$$

$$Q_{\text{heating}} = 4425152 \text{ KJ/h}$$

This heat is supplied by burning of natural gas

$$\text{NHV} = 1005 \text{ Btu/ft}^3$$

$$\rho = 0.0471 \text{ lb/ft}^3$$

NHV = Net heating value

$$\text{NHV} = 1005 \text{ Btu/ft}^3 * 1.055 \text{ KJ/Btu} * \text{ft}^3 / 0.0471 \text{ lb} * 2.2 \text{ lb/Kg}$$

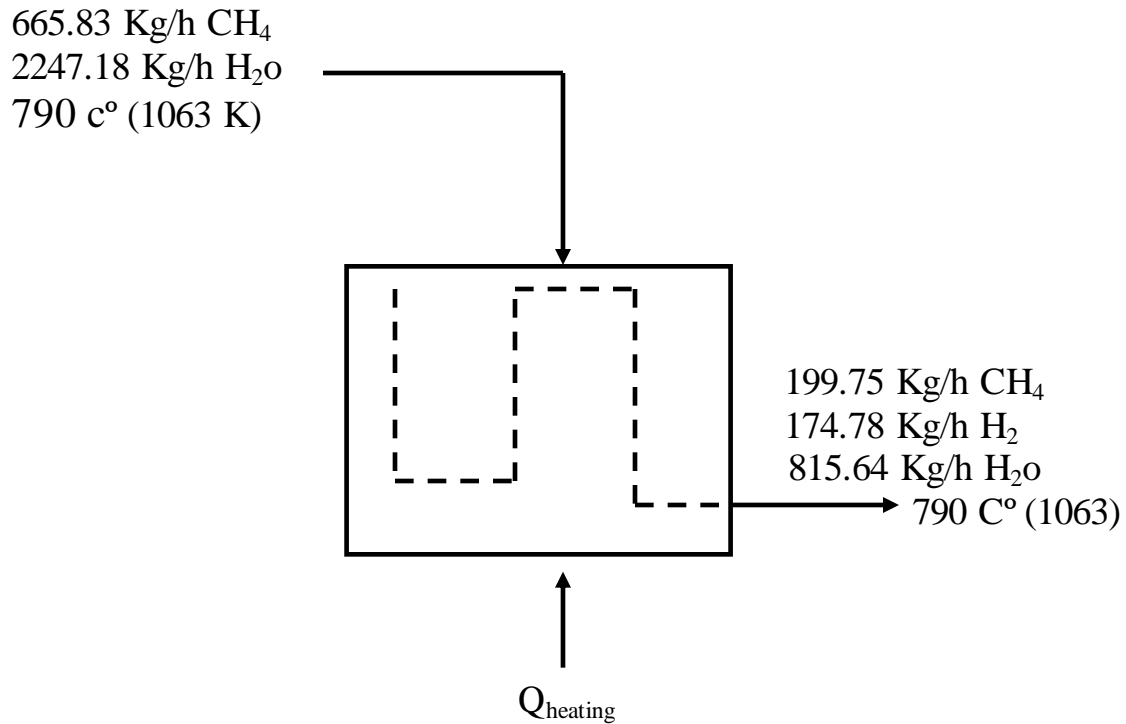
$$\text{NHV} = 49525 \text{ KJ/kg}$$

$$Q_{\text{heating}} = m (\text{NHV})$$

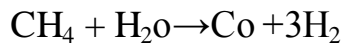
$$M = Q_{\text{heating}} / \text{NHV} = 4425152 / 49525$$

$$= 89.35 \text{ Kg/h}$$

Energy balance on primary reformer :



Reaction



Energy balance

$$Q_{\text{in}} + Q_{\text{heating}} = Q_{\text{out}} + Q_{\text{reaction}}$$

$$T_{\text{raf}} = 25\text{C}^\circ (298\text{K})$$

$$Q_{\text{in}} = \sum n_i \int_{298}^{1063} C_{p_i} dT$$

$$Q_{\text{out}} = \sum n_i \int_{298}^{1063} C_{p_i} dT$$

$$\sum n_i \int_{298}^{1063} C_{p_i} dT = 6138415 \text{ KJ/h}$$

$$Q_{\text{out}} = 6138415 \text{ KJ/h}$$

$$\Delta H_{r_{298}} = \sum \Delta H_{f_{298}} - \sum \Delta H_{f_{298}}$$

$$\Delta H_{r_{298}} = 0 + (-110.62) - (-242) - (-74.86)$$

$$= 206.24 \text{ KJ/mol} = 206240 \text{ KJ/Kmol}$$

CH₄ reaction n= 29.1301 Kmol/h

$$Q_{\text{reaction}} = n \Delta H_r$$

$$= (29.1301) * (206240) = 6007792 \text{ KJ/h}$$

$$5365347 + Q_{\text{heating}} = 6138415 + 6007792$$

$$Q_{\text{heating}} = 6780860 \text{ KJ/h}$$

$$Q_{\text{heat}} = m (\text{NHV})$$

$$m = Q/\text{NHV} = 6780860/49525 = 137 \text{ Kg/h}$$

Energy balance on secondary reformer:

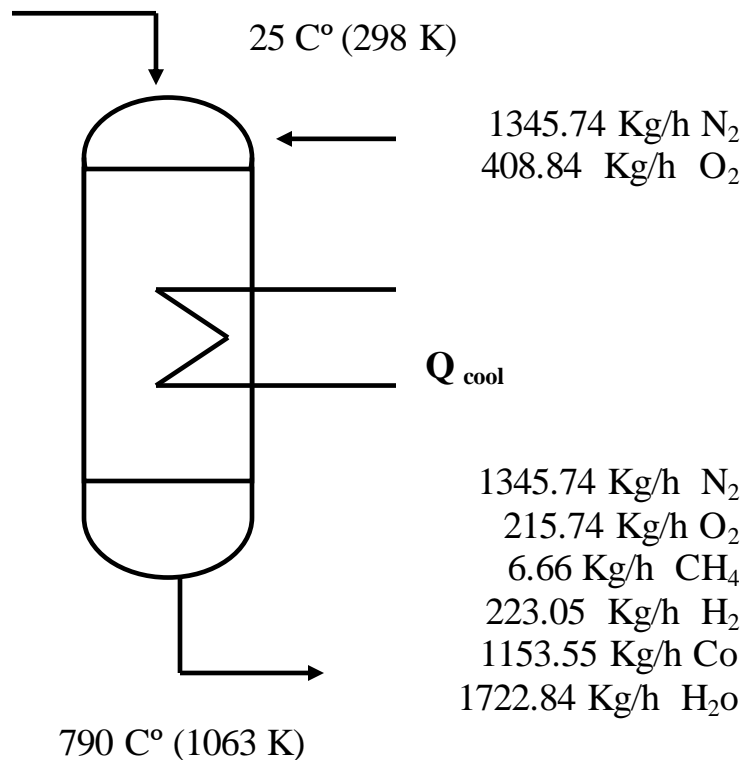
790 C° (1063 K)

199.75 Kg/h CH₄

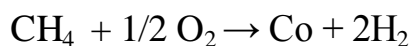
172.78 Kg/h H₂

815.64 Kg/h Co

1722.84 Kg/h H₂O



Reaction



Energy balance

$$Q_{\text{in}} = Q_{\text{out}} + Q_{\text{reaction}} + Q_{\text{cool}}$$

$$T_{\text{ref}} = 25 \text{ C}^\circ = 298\text{K}$$

$$Q_{\text{in}} = \sum n_i \int_{298}^{298} C_{p,i} dT + \sum n_i \int_{298}^{1063} C_{p,i} dT$$

$$\sum n_i \int_{298}^{298} C_{p,i} dT = 0$$

$$\sum n_i \int_{298}^{1063} C_{p,i} dT = 6138415 \text{ KJ/h}$$

$$Q_{\text{in}} = 0 + 6138415$$

$$= 6138415 \text{ KJ/h}$$

$$Q_{\text{out}} = \sum n_i \int_{298}^{1063} C_{p,i} dT$$

$$= 6259144 \text{ KJ/h}$$

$$\Delta H_{r,298} = \sum \Delta H_{f,298} - \sum \Delta H_{f,298}$$

$$= 0 + (-110.62) - 0 - (-74.86)$$

$$= -35.76 \text{ KJ/mol CH}_4$$

$$= -35760 \text{ KJ/Kmol CH}_4$$

$$\text{CH}_4 \text{ reaction } n = 12.0682 \text{ Kmol/h}$$

$$Q_{\text{reaction}} = n \Delta H_r$$

$$= (12.0682) * (-35760) = -431559 \text{ KJ/h}$$

$$6138415 = 6259144 - 431559 + Q_{\text{cool}}$$

$$Q_{\text{cool}} = 310830 \text{ KJ/h}$$

This heat is removed by cooling

Water inlet at 25 C° and 20 bar and exit saturated steam at 20 bar for steam tables

$$T_{\text{sat}} = 212 \text{ C}^\circ$$

$$\lambda = 1890.2 \text{ KJ/Kg}$$

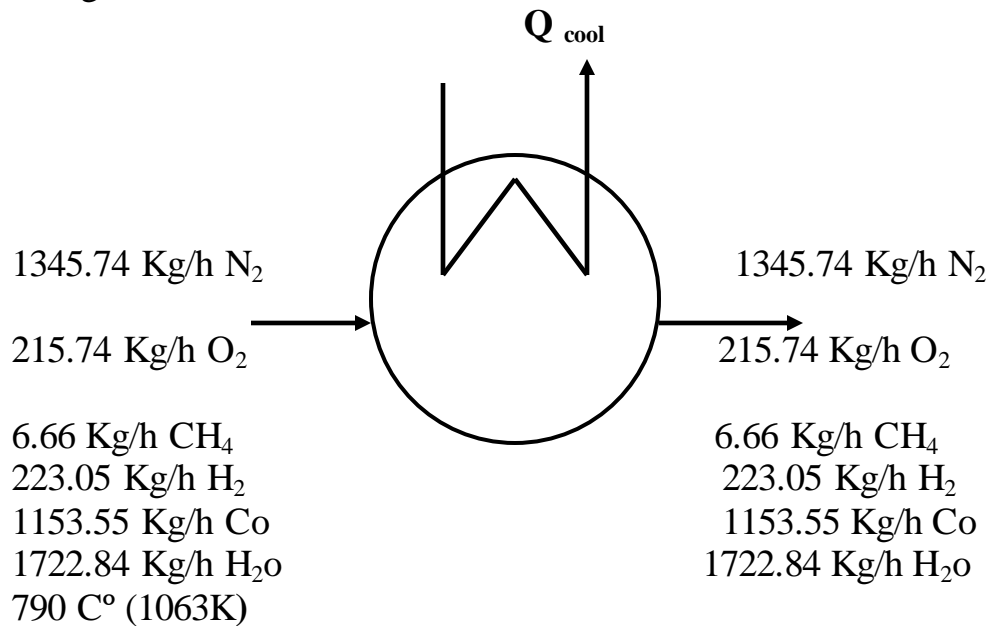
$$Q_{\text{cool}} = m C_p \Delta T + m \lambda$$

$$310830 = m * 412 * (214 - 25) + 1890.2m$$

$$m = 116 \text{ Kg/h H}_2\text{O}$$

Energy balance on water heat boiler

The gases are cooled from 790 to 232 C°



Energy balance

$$Q_{\text{in}} = Q_{\text{out}} + Q_{\text{cool}}$$

$$T_{\text{ref}} = 25 \text{ C}^\circ = 298 \text{ K}$$

$$Q_{\text{in}} = \sum n_i \int_{298}^{1063} C_{p_i} dT$$

$$= 6259144 \text{ KJ/h}$$

$$Q_{\text{out}} = \sum n_i \int_{298}^{505} C_{p_i} dT$$

$$= 1886185 \text{ KJ/h}$$

$$6259144 = 1886185 + Q_{\text{cool}}$$

$$Q_{\text{cool}} = 4372959 \text{ KJ/h}$$

This heat is removal by cooling water inlet at 25 C° and 105 Kg/Cm² (103 bar) and exit

Saturated steam at 103 bar from steam tables

$$\lambda = 1294.1 \text{ KJ/h} , \quad T_{\text{sat}} = 314 \text{ C}^\circ$$

$$Q_{\text{cool}} = m C_p \Delta T + m \lambda$$

$$4372959 = m * 4.2 * (314-25) + 1 * 294.1m$$

$$m = 1744 \text{ Kg/h} , \quad = 0.48 \text{ Kg/h H}_2\text{O}$$

Energy balance on shift converter:

1345.74 Kg/h N₂

215.74 Kg/h O₂

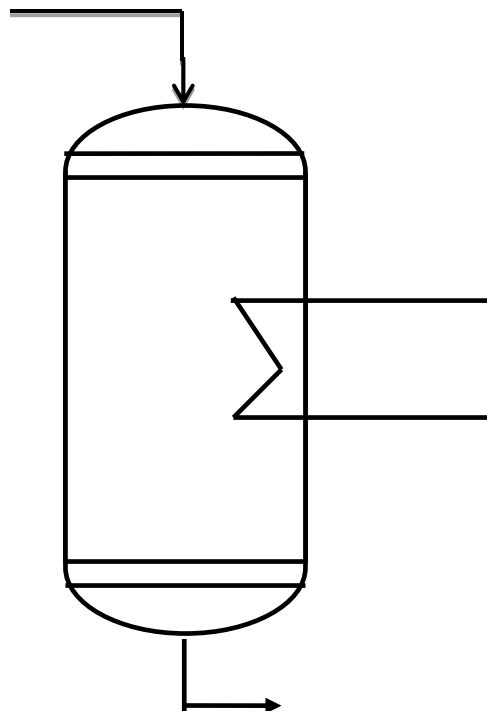
6.66 Kg/h CH₄

223.05 kg/h H₂

1153.55 Kg/h Co

1722.84 Kg/h H₂O

232 C° (505K)



1345.74 Kg/h N₂

215.74 Kg/h O₂

6.66 Kg/h CH₄

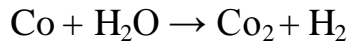
301.33 Kg/h H₂

57.68 Kg/h Co

1722.09 kg/h Co₂

1018.34 Kg/h H₂O

Reaction



Energy balance

$$Q_{\text{in}} = Q_{\text{out}} + Q_{\text{reaction}} + Q_{\text{cool}}$$
$$T_{\text{ref}} = 25 \text{ C}^\circ = 298 \text{ K}$$

$$Q_{\text{in}} = \sum n_i \int_{298}^{505} C_{p_i} dT$$
$$= 1886185 \text{ KJ/h}$$

$$Q_{\text{out}} = \sum n_i \int_{298}^{505} C_{p_i} dT$$
$$= 1894616 \text{ KJ/h}$$

$$\Delta H_{r_{298}} = 0 + (-393.77) - (-242) - (-110.62)$$
$$= -41.15 \text{ KJ/mol}$$
$$= -41150 \text{ KJ/Kmol}$$

Co reaction $n = 39.1384 \text{ Kmol/h}$

$$Q_{\text{reaction}} = n \Delta H_r$$
$$= (39.1384) * (-41150)$$
$$= -1610545 \text{ KJ/h}$$

$$1886185 = 1895616 - 1610545 + Q_{\text{cool}}$$

$$Q_{\text{cool}} = 1601114 \text{ KJ/h}$$

This heat is removed by cooling water at 25 C° and 10 bar and exit saturated steam at 10 bar , $T_{\text{sat}} = 180 \text{ C}^\circ$

$$\lambda = 2113.2 \text{ KJ/h}$$

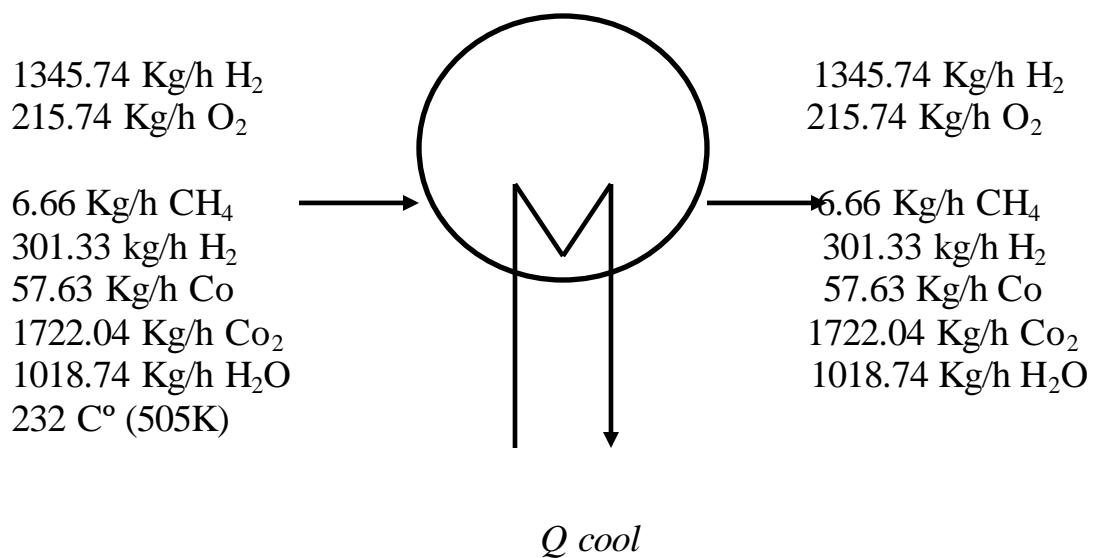
$$Q_{\text{cool}} = m C_p \Delta T + m \lambda$$

$$1601114 = m \cdot 2.4 \cdot (180 - 25) + 2113.2 m$$

$$M = 579 \text{ Kg/h H}_2\text{O}$$

Energy balance on waste heat boiler:

The gases are cooled from 232 to 150 C°



Energy balance

$$Q_{in} = Q_{out} + Q_{cool}$$

$$T_{ref} = 25 \text{ C}^\circ = 298 \text{ K}$$

$$Q_{in} = \sum n_i \int_{298}^{505} C_{pi} dT$$

$$= 1895616 \text{ KJ/h}$$

$$Q_{out} = \sum n_i \int_{298}^{423} C_{pi} dT$$

$$Q_{out} = 1080501 \text{ KJ/h}$$

$$1895616 = 1080501 + Q_{cool}$$

$$Q_{\text{cool}} = 815115 \text{ KJ/h}$$

This heat is removed by cooling water at 25 C° and 10 bar and exit saturated steam at

10 bar

$$T_{\text{sat}} = 180 \text{ C}^\circ$$

$$\lambda = 2113.2 \text{ KJ/h}$$

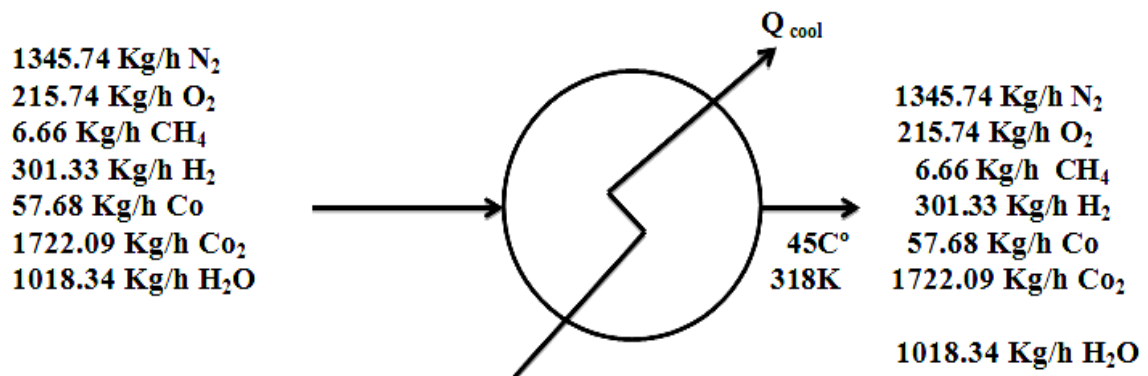
$$Q_{\text{cool}} = m C_p \Delta T + m \lambda$$

$$815115 = m \cdot 4.2 \cdot (180 - 25) + 2113.2 m$$

$$m = 295 \text{ Kg/h}$$

Energy balance on partial condenser:

The gases are cooled from 150 to 45 C° where water is condenser



Energy balance

$$Q_{\text{in}} = Q_{\text{out}} + Q_{\text{cool}}$$

$$T_{\text{ref}} = 25 \text{ C}^\circ = 298 \text{ K}$$

$$Q_{\text{in}} = \sum n_i \int_{298}^{423} C_{p_i} dT + m \lambda$$

$$\sum n_i \int_{298}^{423} C_{p_i} dT = 1080501 \text{ KJ/h}$$

$$M \lambda = (1018.34) * (2256.9)$$

$$= 2298292 \text{ KJ/h}$$

$$Q_{in} = 1080501 + 2298292$$

$$= 3378793 \text{ KJ/h}$$

$$Q_{out} = \sum n_i \int_{298}^{318} C_{pi} dT + m C_p \Delta T$$

$$\sum n_i \int_{298}^{318} C_{pi} dT = 131514 \text{ KJ/h}$$

$$m C_p \Delta T = (1018.34) * (4.20) * (318 - 298)$$
$$= 85541 \text{ KJ/h}$$

$$Q_{out} = 131514 + 85541$$
$$= 217055 \text{ KJ/h}$$

$$3378793 = 217055 + Q_{cool}$$

$$Q_{cool} = 3161738 \text{ KJ/h}$$

$$Q_{cool} = m C_p \Delta T$$

For cooling water

$$T_{in} = 25 \text{ C}^\circ$$

$$T_{out} = 40 \text{ C}^\circ$$

$$m = Q_{cool} / C_p \Delta T$$

$$m = 3161738 / 4.2 * (40.25)$$

$$m = 50186 \text{ Kg/h}$$

$$= 13.9 \text{ Kg/s}$$

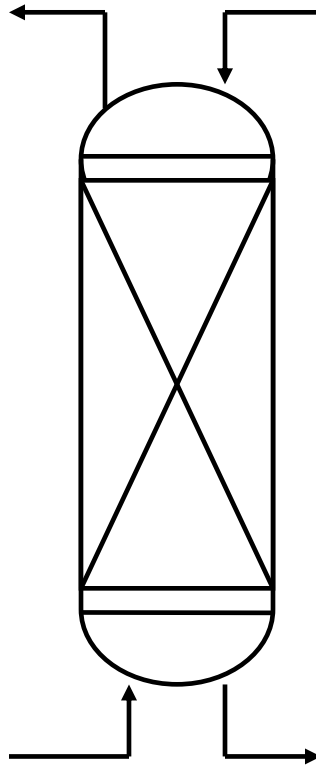
Energy balance on absorber

5968.61 Kg/h MEA
 23874.44 Kg/h H₂
 72.02 Kg/h (C₂H₅ONH₃)₂CO₃

1345.74 Kg/h N₂
 215.74 Kg/h O₂
 6.66kg/hr CH₄
 301.33 Kg/h H₂
 57.68 Kg/h Co
 17.22 Kg/h Co₂

P = 30 atm

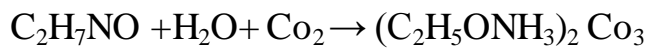
1345.74 Kg/h N₂
 215.74 Kg/h O₂
 6.66 Kg/h CH₄
 301.33 Kg/h H₂
 57.68 Kg/h Co
 17.22 Kg/h Co₂
 45 C° (318K)



23176.99 Kg/h H₂O
 1241.47 Kg/h MEA

7201.47 Kg/h (C₂H₅ONH₃)₂ Co₃

Reaction



Component	Cp _L KJ/Kg .k
H ₂ O	4.20
ME A	2.67
(C ₂ H ₅ ONH ₃) ₂ Co ₃	2.93

Energy balance

$$Q_{in} = Q_{out} + Q_{reaction}$$

$$T_{ref} = 25 \text{ C}^\circ = 298 \text{ K}$$

$$Q_{in} = \sum n_i \int_{298}^{318} C_{p,i} dT + \sum m_i C_{p,i} \Delta T$$

$$\sum n_i \int_{298}^{318} C_{p,i} dT = 131514 \text{ KJ/h}$$

$$\sum m_i C_{p,i} \Delta T = \{(23874.44 * 4.20) + (5968.61 * 2.67) + (72.02 * 2.93)\} * \{318 - 298\}$$

$$= 2328397 \text{ KJ/h}$$

$$Q_{in} = 131514 + 2328397 = 2459911 \text{ KJ/h}$$

By trial and error assume exit temperature = 70 C° (333 K)

Mean heat Capacity of gases between 25.7

Component	Cp KJ/Kg .k
N₂	1.04
O₂	0.92
CH₄	2.31
H₂	14.39
Co	1.04
Co₂	0.86

$$Q_{out} = \sum m_i C_{p,i} \Delta T + \sum n_i \int_{298}^{333} C_{p,i} dT$$

$$= [(1345.74 * 1.04) + (215.74 * 0.92) + (6.66 * 2.31) + (301.33 * 14.39) + (57.68 * 1.04) + (17.22 * 0.86)]$$

$$* [T - 298] + [(23176.99 * 4.20) + (1241.47 * 2.67) + (7201.47 * 2.093)] * [T - 298]$$

$$Q_{out} = 6024.3701 (T - 298) + 121758.39 (T - 298)$$

$$Q_{\text{out}} = 127782.76 (T-298)\text{KJ/h}$$

$$\Delta H_r = -825 \text{ BTU/lb } \text{CO}_2$$

$$\Delta H_r = (-825 \text{ BTU/lb } \text{CO}_2) * (1.055 \text{ KJ/BTU}) * (2.2 \text{ lb/Kg}) * (44 \text{ Kg/Kmol})$$

$$\Delta H_r = -84252 \text{ KJ/Kmol } \text{CO}_2 \text{ reaction}$$

$$\text{CO}_2 \text{ reacted } n = 38.7470 \text{ Kmol/h}$$

$$Q_{\text{reaction}} = n \Delta H_r$$

$$= (38.7470) * (-84252) = -3264512 \text{ KJ/h}$$

$$2459911 = 127782.79 (T-298) - 3264512$$

$$5724423 = 127776.6 * (T-298)$$

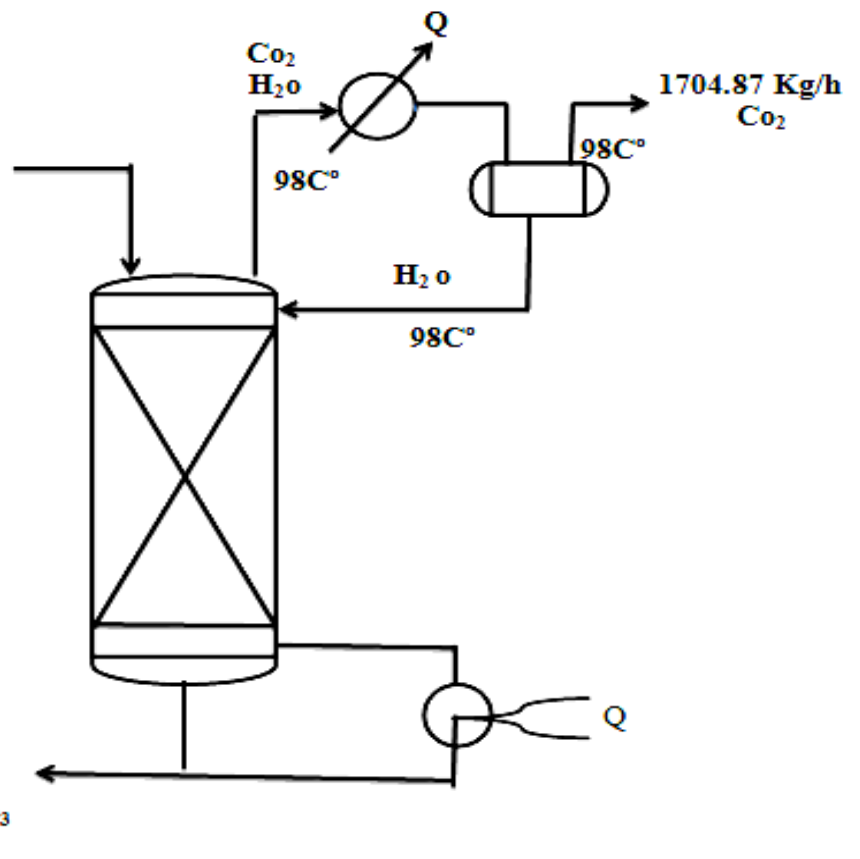
$$T = 342.8 \text{ K} = 69.8 \text{ C}^\circ$$

$$T \text{ assume} = 70 \text{ C}^\circ$$

Energy balance on stripper operating conditions:

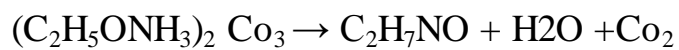
Top	Bottom
T=208 F° = 98 C°	T= 240 F° =116 C°
P =20 psia = 137.3 Kpa	P= 24 psia =165.4Kpa

23176.99 Kg/h H₂O
 1241.47 Kg/h MEA
 90 C° (363K)



23874.44 Kg/h H₂o
 5968.61 Kg/h MEA
 72.02 Kg/h (C₂H₅ONH₃)₂ Co₃
 116 C°(389K)

Reaction



Energy balance

$$Q_{in} + Q_{req} = Q_{out} + Q$$

$$T_{ref} = 25 C^\circ$$

$$Q_{in} = \sum m_i C_{p,i} \Delta T$$

$$= [(23176.99 * 4.20) + (1241.47 * 2.67) + (7201.47 * 2.93)] * [363 - 298]$$

$$Q_{in} = 7914295 \text{ KJ/h}$$

$$Q_{\text{out}} = n \int_{298}^{371} C_p dT + \sum m_i C_{p,i} \Delta T$$

$$n \int_{298}^{371} C_p dT = 112010 \text{ KJ/h}$$

$$\sum m_i C_{p,i} \Delta T = [(23874.44 \cdot 4.2) + (5968.61 \cdot 2.67) + (72.02 + 2.93)] \cdot [389 - 298]$$

$$= 10594207 \text{ KJ/h}$$

$$Q_{\text{out}} = 112010 + 10594207$$

$$= 10706212 \text{ KJ/h}$$

Calculation of Q_c

$$Y_w = P_w / P$$

Y_w = mol fraction of water

P_w = partial pressure of water

P = total pressure Kpa

Assume exit CO_2 is saturated with water vapor at to top conditions

$$T = 98 \text{ C}^\circ$$

$$P = 137.8 \text{ Kpa}$$

At saturation

Partial pressure of water vapor = vapor pressure of water vapor

$$P_w = P_w^*$$

$$Y_w = P_w^* / P$$

Vapour pressure of water at 98 C°

$$P_w^* = 94.3 \text{ Kpa}$$

$$Y_w = 94.3 / 137.8 = 0.6843$$

$$Y_w = \text{Kmol H}_2\text{O}/\text{total Kmol}$$

$$\text{H}_2\text{O} = w \text{ Kmol/h}$$

$$\text{CO}_2 = 1704.87/44 = 38.7470 \text{ Kmol/h}$$

$$W = 83.99 \text{ Kmol H}_2\text{O} \quad *18$$

$$= 1511.76 \text{ Kg/h}$$

$$Q_c = m \lambda$$

$$\text{At } 98 \text{ C}^\circ \quad \lambda = 2262.2 \text{ KJ/Kg}$$

$$Q_c = (1511.76) * (2262.2) = 3419904 \text{ KJ/h}$$

$$7914295 + Q_{\text{reb}} = 10706217 + 3419904$$

$$Q_{\text{reb}} = 6211826 \text{ KJ/h}$$

$$Q_c = m C_p \Delta T$$

For cooling water

$$T_{\text{in}} = 25 \text{ C}^\circ$$

$$T_{\text{out}} = 40 \text{ C}^\circ$$

$$m = Q_c / C_p \Delta T = 3419904 / 4.2 * (40.25)$$

$$= 54284 \text{ Kg/h}$$

$$= 1511 \text{ Kg/s}$$

$$Q_{\text{reb}} = m \lambda$$

Using saturated steam at 180 C°

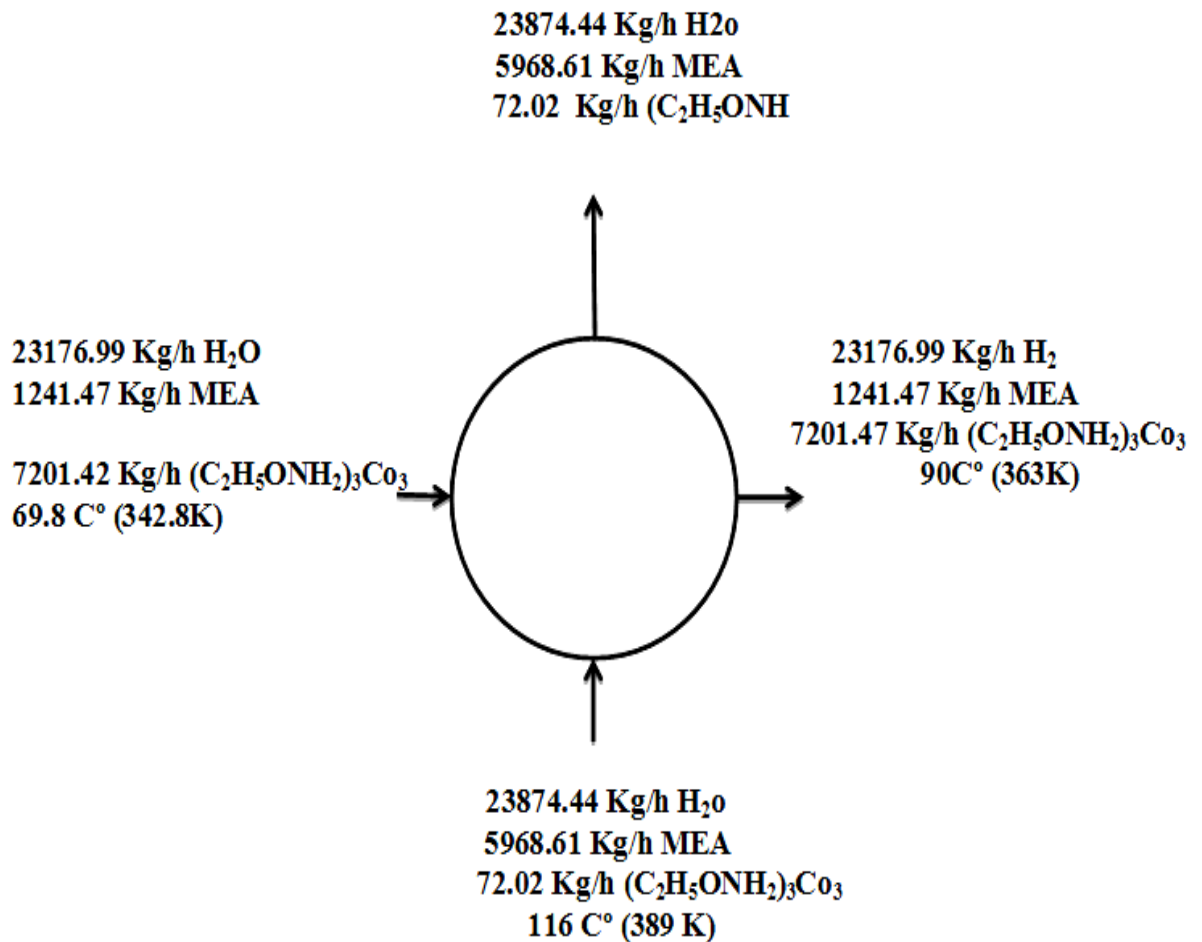
$$\lambda = 2113.2 \text{ KJ/Kg}$$

$$m = Q_{\text{reb}} / \lambda = 6211826 / 2113.2$$

$$= 2940 \text{ Kg/h}$$

$$= 0.82 \text{ Kg/s}$$

**Energy balance on heat exchanger
bottom of Absorber**



Energy balance

$$Q_{in} = Q_{out}$$

$$T_{ref} = 25C^{\circ} = 298K$$

$$Q_{in} = \sum m_i C_{p,i} \Delta T + \sum m_i C_{p,i} \Delta T$$

$$\sum m_i C_{p,i} \Delta T = 5724423 \text{ KJ/h}$$

$$\sum m_i C_{p,i} \Delta T = 10594207 \text{ KJ/h}$$

$$Q_{in} = 5724423 + 10594207$$

$$= 16318632 \text{ KJ/h}$$

$$Q_{\text{out}} = \sum m_i C_{pi} \Delta T + \sum m_i C_{pi} \Delta T$$

$$\sum m_i C_{pi} \Delta T = 7914295 \text{ KJ/h}$$

$$\sum m_i C_{pi} \Delta T = [(2384.44 \cdot 4.2) + (5968.61 \cdot 2.67) + (72.02 \cdot 2.93)] \cdot [T - 298]$$

$$= 116419.86 (T - 298) \text{ KJ/h}$$

$$16318632 = 7914295 + 116419.86(T - 298)$$

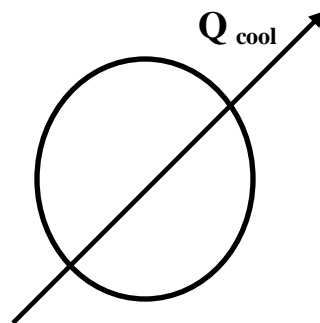
$$8404337 = 116419.86(T - 298)$$

$$T = 370 \text{ K}$$

= 97°C to absorber

Energy balance on cooler :

23874.44 Kg/h H₂O
 5968.61 Kg/h MEA
 72.02 Kg/h (C₂H₅ONH₃)₂ CO₃
 45 C°



23874.44 Kg/h H₂O
 5968.61 Kg/h MEA
 72.02 Kg/h (C₂H₅ONH₃)₂ CO₃
 97 C°

Energy balance

$$Q_{\text{in}} = Q_{\text{out}} + Q_{\text{cool}}$$

$$T_{\text{ref}} = 25 \text{ C}^\circ (298)$$

$$Q_{in} = \sum m_i C_{pi} \Delta T$$
$$= 8404337 \text{ KJ /h}$$

$$Q_{out} = \sum m_i C_{pi} \Delta T$$
$$= 2328397 \text{ KJ/h}$$

$$8404337 = 2328397 + Q_{cool}$$

$$Q_{cool} = 6075940 \text{ KJ/h}$$

$$Q_{cool} = m C_p \Delta T$$

For cooling water

$$T_{in} = 25 \text{ C}^\circ$$

$$T_{out} = 40 \text{ C}^\circ$$

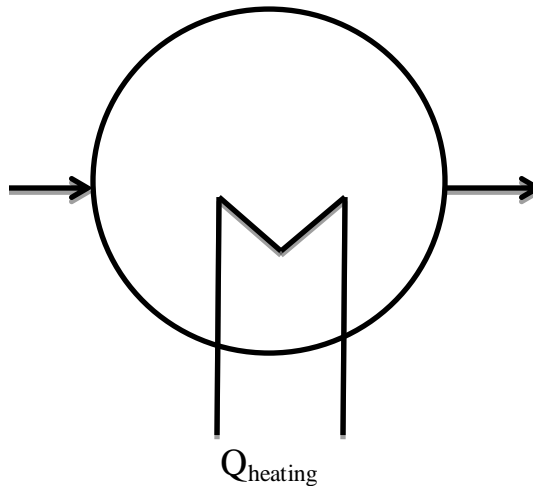
$$m = Q_{cool} / C_p \Delta T$$

$$= 6075940 / 4.2 * (40.25) = 96443 \text{ Kg/h}$$

$$= 26.8 \text{ Kg/s}$$

Energy balance on heat exchanger

1345.74 Kg/h N₂
 215.74 Kg/h O₂
 6.66 Kg/h CH₄
 301.33 Kg/h H₂
 57.68 Kg/h Co
 17.22 Kg/h Co₂
 70 C° (343K)



1345.74 Kg/h N₂
 215.74 Kg/h O₂
 6.66 Kg/h CH₄
 301.33Kg/h H₂
 57.68 Kg/h Co
 17.22 Kg/h Co₂
 T= 300 C° (573K)

Energy balance

$$Q_{in} + Q_{heating} = Q_{out}$$

$$T_{ref} = 25C^{\circ} = 298k$$

$$Q_{in} = \sum n_i \int_{298}^{343} C_p dt$$

$$Q_{in} = 5724423 \text{ KJ/h}$$

$$Q_{out} = 38480843 \text{ KJ/h}$$

$$5724433 + Q_{heat} = 38480843$$

$$Q_{heat} = 32756420 \text{ KJ/h}$$

$$Q_{heat} = m\lambda$$

Using saturated steam at 350 C°

$\lambda = 895.6 \text{ KJ/kg}$ from steam table

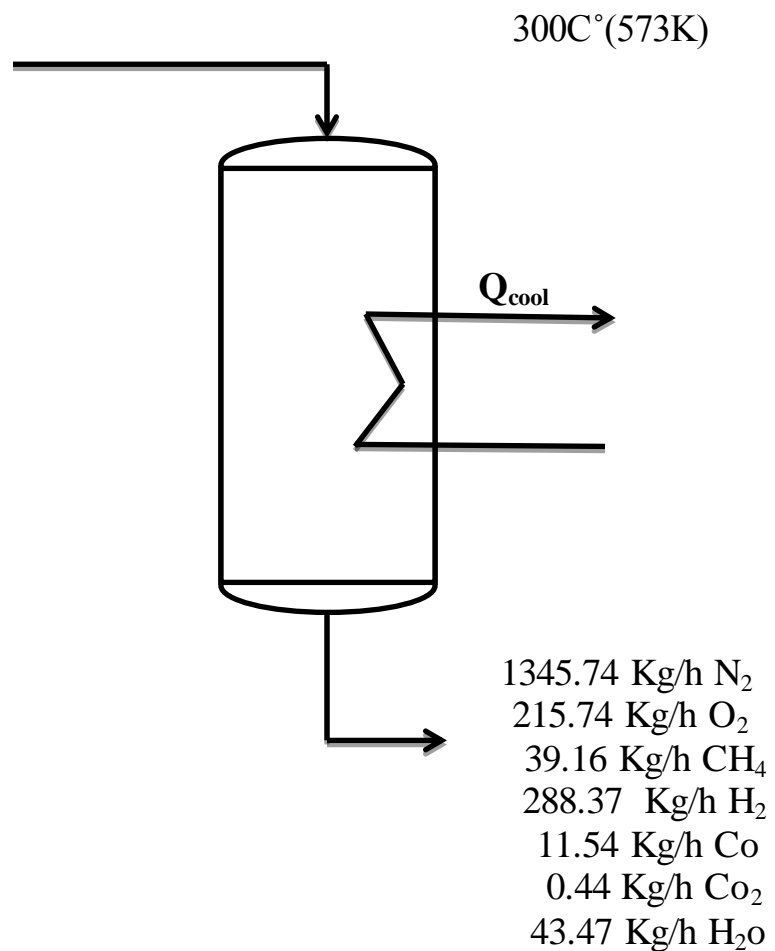
$$m = Q_{\text{heat}} / \lambda$$

$$m = 32756420 / 895.6 = 36575 \text{ Kmol/h}$$

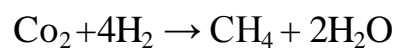
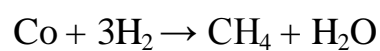
$$= 10.16 \text{ Kg/s}$$

Energy balance on methanomator

1345.74 Kg/h N₂
 215.74 Kg/h O₂
 6.66 Kg/h CH₄
 301.33 Kg/h H₂
 52.68 Kg/h Co
 17.22 Kg/h Co₂



Reaction



Energy balance

$$Q_{in} = Q_{out} + Q_{cool} + Q_{reaction}$$

$$T_{ref} = 25C^{\circ} = 298K$$

$$Q_{in} = \sum n_i \int_{298}^{573} C_{pi} dT$$

$$Q_{in} = 38480843 \text{ KJ/h}$$

$$Q_{out} = \sum n_i \int_{298}^{573} C_{pi} dT$$

$$= 37711226 \text{ KJ/h}$$

$$\Delta H_{r298} = (-242) + (-74.86) - 0 - (-110.62)$$

$$= -206.24 \text{ KJ/mol}$$

$$= -206240 \text{ KJ/Kmol}$$

$$C_o \text{ reacted } n_i = 106479 \text{ Kmol/h}$$

$$n \Delta H_{r1} = (1.6479) * (-206240) = -339863 \text{ KJ/h}$$

$$\Delta H_{r2} = 2 * (-242) + (-74.86) - 0 - (-393.77)$$

$$\Delta H_{r2} = -165.09 \text{ KJ/mol}$$

$$= -165090 \text{ KJ/Kmol}$$

$$CO_2 \text{ reacted } n_2 = 0.3836 \text{ Kmol/h}$$

$$n_2 \Delta H_{r2} = 0.3836 * (-165090)$$

$$= -63329 \text{ KJ/h}$$

$$Q_{reaction} = -339863 - 63329$$

$$= -403192 \text{ KJ/h}$$

$$38480843 = 37711226 - 403192 + Q_{cool}$$

$$Q_{\text{cool}} = 1172809 \text{ KJ/h}$$

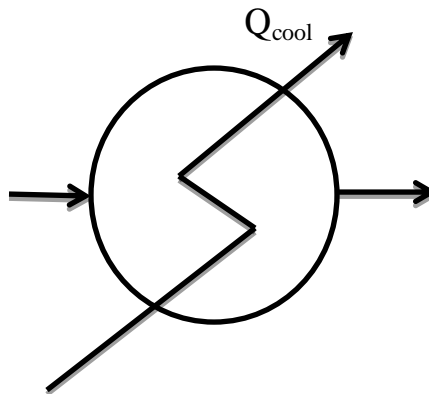
$$Q = m C_p \Delta T + m\lambda$$

$$1172809 = m \cdot (100 - 25) + 2256.9 m$$

$$M = 456 \text{ Kg/h H}_2\text{O}$$

Energy balance on waste e that boiler

1345.74 Kg/h N
 215.74 Kg/h O₂
 39.16 Kg/h CH₄
 288.33 Kg/h H₂
 11.54 Kg/h Co
 0.44 Kg/h Co₂
 43.47 H₂O
 300 C°



1345.74 Kg/h N₂
 215.74 Kg/h O₂
 39.16 Kg/h CH₄
 288.33 Kg/h H₂
 11.54 Kg/h Co
 0.44 Kg/h Co₂
 43.47 Kg/h H₂O
 100 C°

Energy balance

$$Q_{\text{in}} = Q_{\text{out}} + Q_{\text{cool}}$$

$$T_{\text{ref}} = 25 \text{ C} = 298 \text{ K}$$

$$Q_{\text{i}} = \sum n_i \int_{298}^{573} c_p dt$$

$$= 37711226 \text{ KJ/h}$$

$$Q_{\text{out}} = \sum n_i \int_{298}^{573} c_p dt$$

$$= 10071982 \text{ KJ/h}$$

$$37711226 = 10079182 + Q_{\text{cool}}$$

$$Q_{\text{cool}} = 27632044 \text{ KJ/h}$$

$$Q_{\text{cool}} = mcp\Delta T + m\lambda$$

$$27632044 = m + 4.2 * (100 - 25) + 2256.7m$$

$$M = 10744 \text{ Kg/h H}_2\text{O}$$

$$= 2.98 \text{ Kg/s}$$

Energy balance on Compressor t

The gases are compressed from 29 to 200 bar

201.859 Kmol/h

Gases

$$T_1 = 45^\circ\text{C}$$

$$P_1 = 29 \text{ bar}$$



201.859 Kmol/h

gases

$$T_2 = ?$$

$$P = 200 \text{ bar}$$

$$(T_2/T_1) = (P_2/P_1)^m$$

$$m = (\gamma - 1) / \epsilon \gamma$$

$$\gamma = C_p / C_v \cong 1.4$$

$$E_p = \text{efficiency of compressor} = 80\%$$

$$M = (1.4 - 1) / (0.8 * 1.4) = 0.35$$

$$(T_2/318) = (200/29)^{0.35}$$

$$T_2 = 625 \text{ K}^\circ$$

$$= 352 \text{ C}^\circ$$

Energy balance on NH₃ conversion

5719.37 Kg/h N₂

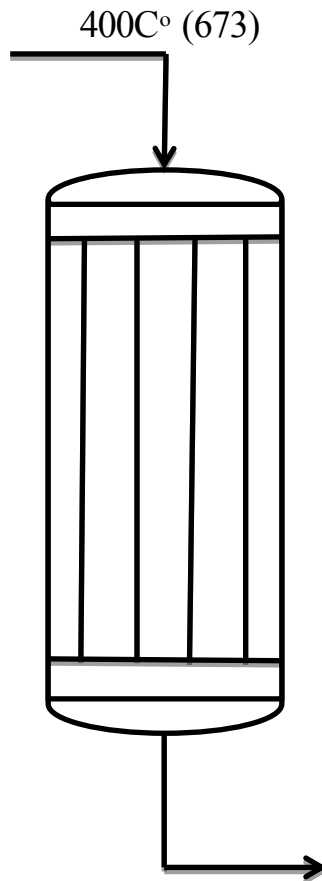
4890.18 Kg/h O₂

887.66 Kg/h CH₄

1225.59 Kg/h H₂

261.48 Kg/h Co

9.92 Kg/h Co₂



457.49 Kg/h N₂

4890.18 Kg/h O₂

887.66 Kg/h CH₄

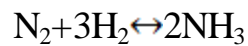
980.47 Kg/h H₂

261.48 Kg/h Co

9.97 Kg/h Co₂

1389 Kg/h NH₃

Reaction



Energy balance

$$Q_{\text{in}} = Q_{\text{out}} + Q_{\text{reaction}} + Q_{\text{cool}}$$

$$T_{\text{ref}} = 25^\circ\text{C} = 298 \text{ K}$$

$$Q_{\text{in}} = \sum n_i \int_{298}^{673} c_p dt$$

$$Q_{\text{in}} = 11866100 \text{ KJ/h}$$

$$Q_{\text{out}} = \sum n_i \int_{298}^{698} c_p dt$$

$$Q_{\text{out}} = 12783745 \text{ KJ/h}$$

$$\Delta H_r = (-45.72) - 0 - 0$$

$$= -45.72 \text{ KJ/mol}$$

$$= -45720 \text{ KJ/mol}$$

$$\text{N}_2 \text{ reacted } n = 40.8530 \text{ Kmole/h}$$

$$Q_{\text{reaction}} = (40.8530) \cdot (-45720)$$

$$= -1867799 \text{ KJ/h}$$

$$11866100 = 12783745 - 1867799 + Q_{\text{cool}}$$

$$Q_{\text{cool}} = 950154 \text{ KJ/h}$$

$$Q_{\text{cool}} = m C_p \Delta T + m \lambda$$

$$950154 = m \cdot 4.2 \cdot (100 - 25) + 2256.9 m$$

$$M = 369.4 \text{ Kg/h } \text{H}_2\text{O}$$

3.1 Introduction to *Equipment design*

Chemical Industries involve problems in process design, unit operations, equipment design and overall plant design. In design of a chemical plant these problems cannot be segregated. However, these problems may be advantageously segregated for study and development because of different principles involved in it. The course goes deeper into the various aspects of mechanical design in the chemical process plant. It stresses upon the design and analysis of the basic process equipment viz. vessels, distillation column, absorption column, heat exchanger, driers and evaporators etc.

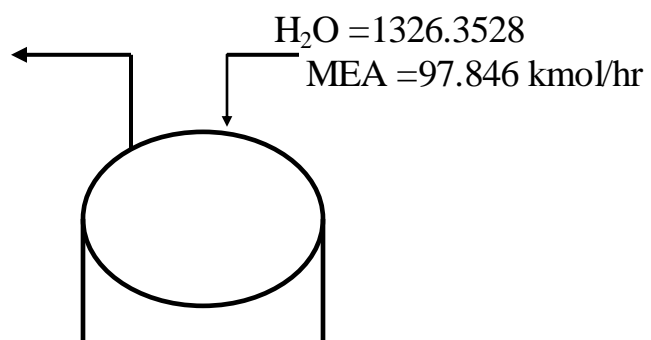
The course emphasizes on the development of design skills among the students to take design related decisions. Whatever be the earlier conception, today a chemical engineer is expected to be able to make complete design of a piece of chemical equipment. The course will be very useful to undergraduate students and practitioners. A number of problems will be solved to illustrate the concepts clearly.

Chapter Three

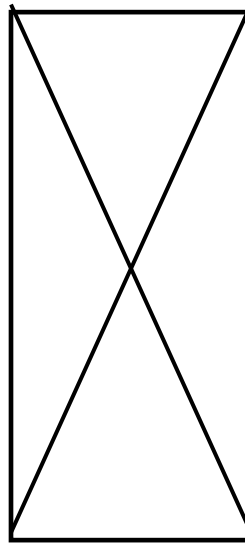
Equipment design

3.2 Absorber design:

48.0623Kmol/h N₂
0.4161Kmol/h CH₄
16.2420 Kmol/h O₂



150.6651Kmol/h H₂
 2.0599 Kmol/h CO
 CO₂=0.3914 Kmol/h



48.0623Kmol/hr N₂
 6.7420Kmol/hr O₂
 0.4161Kmol/hr CH₄

150.6651Kmol/hr H₂
 39.1384Kmol/hr CO₂
 2.0599Kmol/hr CO



H₂O=1287.6108 Kmol/hr

MEA=20.3520 Kmol/hr

Input : let A= CO₂= 39.1384 Kmol/hr

B=207.9 Kmol/hr

$y_1 = A/B = 39.1384/207.9 = 0.188$

Output :A= CO₂=0.3914 Kmol/hr

B=207.9 Kmol/hr

$y_2 = A/B = 0.3914 /207.9 = 0.0018$

G_s= B=207.9 Kmol/hr

$G_s y_1 + L_s x_2 = G_s y_2 + L_s x_1$

$x_2 = 0$

$207.9*0.188 = 207.9*0.0018 + x_1 *1347.1$

$x_1 = 0.03$

from (x_1, y_1) & (x_2, y_2)

$m = 4.3$

$$\Phi = m G_s / L_s = 4.3 * 207.9 / 1347.1 = 0.66$$

$$y_1 / y_2 = 0.188 / 0.0018 = 104.4$$

from fig.(3.1) (appendix A)

$$\text{NOG} = 10$$

$$\text{HTU} = 0.5 \text{ m (table 3.1)(appendix B)}$$

Where :

HTU :the height of a transfer unit

$$Z = 0.5 * 10 = 5 \text{ m}$$

$$G_s = 207.9 \text{ kmol /hr} = 0.057 \text{ kmol/s}$$

$$L_s = 1347.1 \text{ kmol /hr} = 0.374 \text{ kmol/s}$$

Where :

G_s : is the mole rate of the inlet gas , kmol/s

L_s : is the mole rate of the liquid solvent, kmol/s

$$\rho_L = 998 \text{ kg/m}^3$$

$$\mu = 10^{-3} \text{ NS/m}^2$$

$$F_p = 170 \text{ m}^{-1}$$

$$\rho_v = 1.094 \text{ kg/m}^3$$

$$F_{LV} = L_s / G_s \sqrt{(\rho_v / \rho_L)}$$

$$G_s = 1.653 \text{ kg/s}$$

$$L_s = 0.618 \text{ kg/s}$$

$$0.618 / 1.653 \sqrt{(1.094 / 998)} = 0.012$$

Design for pressure drop of 20mm H₂O

From fig.(3.2) (appendix A)

$$K_4 = 1.51$$

At flooding $K_4 = 1.64$

$$\text{Percent flooding} = \sqrt{(1.51 / 1.64)} * 100 = 95\%$$

$$V_w^* = \left[K_4 \rho_v (\rho_L - \rho_v) / 13.1 F_p (\mu / \rho_L) \right]$$

Where:

V_w^* : gas mass flow-rate per unit column cross-sectional area, kg/m²s

F_p : packing factor, characteristic of the size and type of packing,

see Table (3.1), m⁻¹.

μ : liquid viscosity, Ns/m²

ρ_L & ρ_v : liquid and vapor densities, kg/m³

$$= 1.7$$

$$\text{Col. area required} = 1.013 / 1.7 = 0.59$$

$$\text{Diameter} = \sqrt{4/\pi * 0.59} = 0.87\text{m}$$

$$\text{Column area} = \pi/4 d^2 = \pi/4(0.87)^2 = 0.56$$

$$\text{Packing size to column diameter ratio} = 0.87/(38*10^{-3}) = 22.8$$

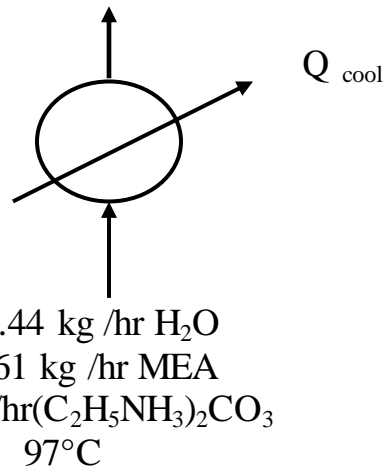
3.3 Heat exchanger design:

45°C (318k)

23874.44 kg /hr H₂O

5968.61 kg /hr MEA

72.02 kg /hr(C₂H₅NH₃)₂CO₃



$$Q_{\text{cool}} = 6075940 \text{ kJ/hr}$$

$$t_{\text{in}} = 25^{\circ}\text{C}$$

$$t_{\text{out}} = 40^{\circ}\text{C}$$

$$Q_{\text{in}} = 8404337 \text{ kJ/hr}$$

$$Q_{\text{out}} = 2328397 \text{ kJ/hr}$$

$$M = 96443 \text{ kg /hr}$$

$$\text{Mean tube temperature } t_m = t_1 + t_2 / 2$$

$$t_m = 97 + 45 / 2 = 71^{\circ}\text{C}$$

physical properties at 71°C of 15% MEA

$$\text{sp. gr} = 0.986$$

$$\rho = 986 \text{ kg /m}^3$$

$$\mu = 4.9 * 10^{-4} \text{ kg/m .s}$$

$$\text{cp mix} = \text{cp} / \text{m } \Delta T$$

$$\text{cp mix} = 6075940 / 29915.07 (97-45) = 3906 \text{ J/kg .c}$$

$$= 3.906 \text{ kJ/ kg .c}$$

Thermal conductivity :

$$\text{H}_2\text{O } k_1 = 156.2 * 10^{-5} \text{ cal /cm .s . } ^{\circ}\text{C}$$

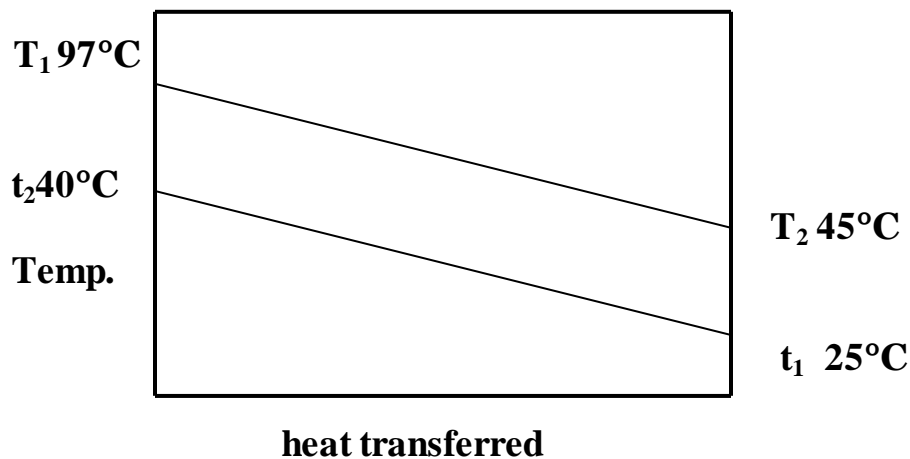
$$= 0.654 \text{ w / m. } ^{\circ}\text{C}$$

$$\text{MEA } k_2 = 0.297 \text{ w/m. } ^{\circ}\text{C}$$

$$K_{\text{mix}} = k_1 x_1 + k_2 x_2$$

$$= 0.654 * 0.85 + 0.277 * 0.15$$

$$= 0.597 \text{ w/m. } ^{\circ}\text{C}$$



$$\Delta T_{1m} = (97 - 40) - (45 - 25) / \ln (97 - 40) / (45 - 25)$$

$$= 35.3^\circ\text{C}$$

$$R = (T_1 - T_2) / (t_2 - t_1) = 97 - 45 / 40 - 25$$

$$= 3.46$$

Use One shell pass & two tube passes

$$S = (t_2 - t_1) / (T_1 - t_1) = 40 - 25 / 97 - 25 = 0.21$$

R: is equal to the shell-side fluid flow-rate times the fluid mean specific heat; divided

by the tube-side fluid flow-rate times the tube-side fluid specific heat.

S: is a measure of the temperature efficiency of the exchanger.

From fig. (3.3) Ft = 0.85

Where : Ft = the temperature correction factor.

ΔT_m = true temperature difference, the mean temperature difference

$$\Delta T_m = \Delta T_{1m} * Ft$$

$$\Delta T_m = 0.85 * 35.3 = 30^\circ\text{C}$$

$$U = 860 \text{ w/m}^2.\text{c}$$

$$\text{Heat load} = 96443 / 3600 * 3.906(97 - 45)$$

$$= 5441.3 \text{ k w}$$

$$A = 5441.3 * 10^3 / (30 * 860)$$

$$= 210 \text{ m}^2$$

Choose 20mm o.d ,16mm i.d ,4.88m long

Allowing for tube –sheet thickness ,take L=4.83m

$$\text{Area of one tube} = 4.83 * 20 * 10^{-3} \Pi = 0.303 \text{ m}^2$$

$$\text{Number of tube} = 210 / 0.303 = 693$$

As the shell –side fluid is relatively clean use 1.25 triangular pitch

$$\text{Tube bundle diameter } D_b = 20 (693 / 0.249)^{1/2.207}$$

$$=727 \text{ mm}$$

Use a split-ring floating head type from fig.(3.4)(appendix A)

$$\text{bundle diametrical clearance} = 69 \text{ mm}$$

$$\text{Shell diameter } D_s = 727 + 69 = 796 \text{ mm}$$

Tube –side coefficient :

$$\text{Tube –cross –sectional area} = \pi/4 * 16^2 = 201 \text{ mm}^2$$

$$\text{Tube per pass} = 693/2 = 347$$

$$\text{Total flow area} = 347 * 201 * 10^{-6} = 0.069 \text{ m}^2$$

$$\text{Cooling flow} = 5441.3 / 3.906 (40-25)$$

$$= 5441.3 / 585.9 = 92.8$$

$$\text{Linear velocity} = 1344 / 995 = 1.35 \text{ m/s}$$

$$\text{Mean mass velocity} = 92.8 / 0.069 = 1344 \text{ kg/s.m}^2$$

The coefficient can also be calculated using equ. 12.15, this is done to illustrate use of this method.

$$h_i d_i / k_f = j_f \text{ Re Pr}^{0.33}$$

$$k \text{ (thermal conductivity)} = 0.59 \text{ w/m} \cdot \text{c}$$

$$\text{Re} = \rho u d / \mu = 993.6 * 1.35 * 16 * 10^{-3} / 7.53 * 10^{-4} = 28501$$

$$\text{Pr} = c_p \mu / k_f = 3.906 * 10^3 * 7.53 * 10^{-4} = 4.98$$

$$L/d_i = 4.83 * 10^3 / 16 = 302$$

$$\text{From fig (3.6) } j_h = 3.5 * 10^{-3}$$

$$h_i = 0.59 / 16 * 10^{-3} * 3.5 * 10^{-3} * 28501 * 4.98^{0.33}$$

$$= 6248 \text{ w/m}^2 \cdot \text{c}$$

Shell – side coefficient:

$$\text{Choose baffle spacing} = D_s / 5 = 796 / 5 = 159.2 \text{ mm}$$

$$\text{Tube pitch } P_t = 1.25 * 20 = 25 \text{ mm}$$

$$\text{Cross –flow area } A_s = (P_t - d_o) D_s l_B / P_t$$

Where:

P_t = tube pitch,

d_o = tube outside diameter,

D_s = shell inside diameter, m,

l_B = baffle spacing, m.

$$A_s = 25 \cdot 20 / 25 \cdot 796 \cdot 159.2 \cdot 10^{-6} = 0.025 \text{ m}^2$$

$$\text{Mass velocity } G_s = W_s / A_s$$

Where : W_s = fluid flow-rate on the shell - side, kg/s

$$= 96443 / 3600 \cdot (1 / 0.025)$$

$$= 1071 \text{ kg/s.m}^2$$

$$\text{Equivalent diameter} = 1.1 / 20 (25^2 - 0.917 \cdot 20^2) = 14.4 \text{ mm}$$

$$\text{Mean shell side temp.} = 71^\circ\text{C}$$

$$Re = G_s d_e / \mu = 1071 \cdot 14.4 \cdot 10^{-3} / 7.53 \cdot 10^{-4}$$

$$= 20481$$

$$Pr = 4.98$$

Choose 25 percent baffle cut

From fig (3.5)(appendix A)

$$j_h = 4.6 \cdot 10^{-3}$$

$$h_s = 0.59 / 14.4 \cdot 10^{-3} \cdot 20481 \cdot 4.98^{1/3} \cdot 4.6 \cdot 10^{-3}$$

$$= 6591 \text{ w/m}^2 \cdot \text{c}$$

Overall coefficient :

$$1/U_o = (1/h_o) + (1/h_{od}) + d_o \ln(d_o/d_i) / 2kw + (d_o/d_i) \cdot (1/h_{id}) + (d_o/d_i) \cdot (1/h_i)$$

Where:

U_o = the overall coefficient based on the outside area of the tube, $\text{w/m}^2 \cdot \text{c}$

h_o = outside fluid film coefficient, $\text{w/m}^2 \cdot \text{c}$

h_i = inside fluid film coefficient, $\text{w/m}^2 \cdot \text{c}$

h_{od} = outside dirt coefficient (fouling factor), $\text{w/m}^2 \cdot \text{c}$

h_{id} = inside dirt coefficient, $\text{w/m}^2 \cdot \text{c}$

kw = thermal conductivity of the tube wall material, $\text{w/m} \cdot \text{c}$

d_i = tube inside diameter, m,

d_o = tube outside diameter, m.

$$1/U_o = (1/6591) + (1/5000) + 20 \cdot 10^{-3} \ln(20/16) / 2 \cdot 50$$

$$+ (20/16) \cdot (1/3000) + (20/16) \cdot (1/4022)$$

$$= 891 \text{ w/m}^2 \cdot \text{c}$$

well above assumed value of $860 \text{ w/m}^2 \cdot \text{c}$

pressure drop:

tube – side :

from fig 12.24 for $Re = 28501$

$$j_f = 3.5 \cdot 10^{-3}$$

$$\Delta P_t = N_p [8 j_f (L/d_i) + 2.5] \rho u_t^2 / 2$$

Where:

ΔP_t = tube-side pressure drop, N/m² (Pa),

N_p = number of tube-side passes,

u_t = tube-side velocity, m/s,

L = length of one tube.

$$\Delta P_t = 2(8 \cdot 3.5 \cdot 10^{-3} \cdot (4.83 \cdot 10^3 / 16) + 2.5) \cdot 993.6 \cdot 1.35^2 / 2$$
$$= 17569 \text{ N/ m}^2$$

Shell –side:

Linear velocity = $G_s / \rho = 1071 / 993.6 = 1.077 \text{ m/s}$

from fig(12.30) (appendix A) for Re = 20481

$$j_f = 4.6 \cdot 10^{-3}$$

$$\Delta P_s = 8 j_f \cdot (D_s / d_e) (L / l_B) \rho u_s^2 / 2$$

where :

L = tube length,

l_B = baffle spacing

$$\Delta P_s = 8 \cdot 4.6 \cdot 10^{-3} (796 / 14.4) \cdot (4.83 \cdot 10^3 / 159.2) \cdot (993.6 \cdot 1.077^2) / 2$$
$$= 35564 \text{ N/ m}^2$$

3.4 Reactor design :

Catalyst:

A triply promoted ($K_2O-CaO-AL_2O_3$) iron oxide catalyst will be used. The iron oxide (Fe_2O_3-FeO) is in the form of nonstoichiometric magnetite. It is made by fusing the magnetite with the promoters.

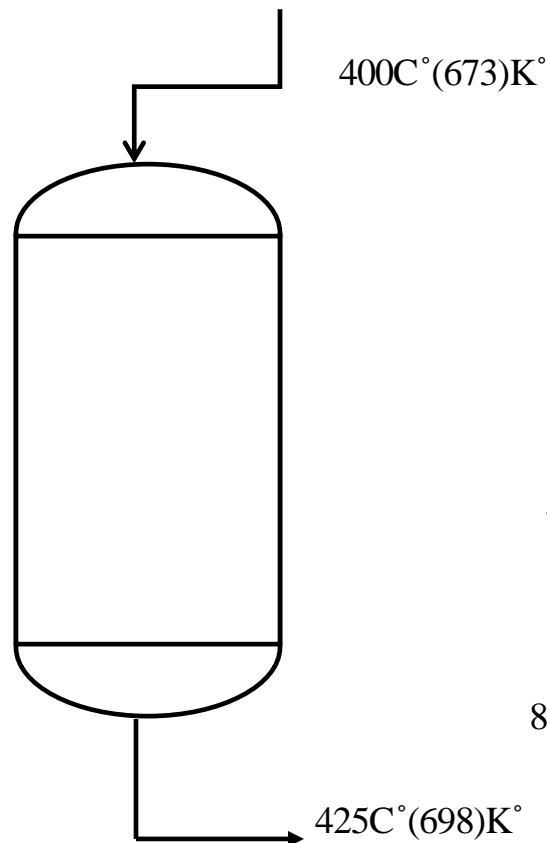
The iron oxide is catalyst

Properties:

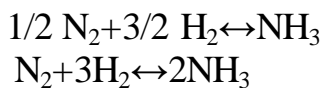
1-particle size = 6 - 10 mm 2- Bulk density = 1200 kg/m^3 3-particle density = 305 lb/ft^3 (4.9 g/cm^3)

204.2633 kmol/h N_2
 152.8182 kmol/h O_2
 55.4788 kmol/h CH_4
 612.7945 kmol/h H_2
 9.3386 kmol/h CO
 0.2267 kmol/h CO_2

P=200bar



163.4103 kmol/h N_2
 152.8182 kmol/h O_2
 55.4788 kmol/h CH_4
 490.2356 kmol/h H_2
 9.3386 kmol/h CO
 0.2267 kmol/h CO_2
 81.7059 kmol/h NH_3



The volume can be evaluated using method as SIMPSONS one -third Rule :

$$V = \int_0^X \frac{FA^\circ}{-rA} dx = h/3 \{f(x)_0 + 4f(x)_1 + f(x)_2\}$$

$$H = X_2 - X_0$$

$$-RA = K_c a$$

$$K_1 = k_1^\circ \exp -E/RT$$

$$K_1^\circ = 1.78954 \cdot 10^{-4} \text{ kmol/m}^3 \cdot \text{hr}$$

$$K_2 = K_2^\circ \exp -E/RT$$

$$K_2^\circ = 2.5714 \cdot 10^6 \text{ kmol/m}^3 \cdot \text{hr}$$

$$E_1 = 20,800 \text{ Kcal /Kg mol}$$

$$E_2 = 47,400 \text{ kcal/kg mol}$$

$$K_1 = 1.78959 \cdot 10^4 \exp -20800/8.314 \cdot 673$$

$$= 0.0434 \text{ kmol/m}^3 \cdot \text{h}$$

$$K_2 = 2.5714 \cdot 10^6 \exp -47400/8.314 \cdot 673$$

$$= 538.398 \text{ kmol/m}^3 \cdot \text{h}$$

$$\xi = yA^\circ \delta$$

$$\xi = 1 \cdot (2 - 4/2)$$

$$= -0.5$$

$$CA^\circ = yA^\circ p / RT^\circ$$

$$CA^\circ = 1 \cdot 197.2 \text{ atm} / 8.314 \cdot 673$$

$$= 0.0352 \text{ kmol/m}^3$$

$$FA^\circ = 204.2633 \text{ kmol/h}$$

Simpsons rule:

X	(1-x)/(1+ξx)	(1+x)/(1-x)=f(x)
----------	---------------------	-------------------------

0	1	1
0.4	0.75	2.33
0.8	0.33	9.1

$$H=0.4 - 0= 0.4$$

$$F(x)=0.4/3 \{1+4*2.33+9.1\}$$

$$=2.58$$

$$V= 204.2633 / (538.398*0.0352) * 2.58$$

$$V= 27.81 \text{ m}^3$$

$$V=\pi /4 * D^2 *L$$

$$27.81= \pi /4 *D^2*(4D)$$

$$D^3=8.852$$

$$D=2.06 \text{ m}$$

$$\text{Hence length (L)}= 4*2.06 = 8.24 \text{ m}$$

Catalyst is divided in 2 beds in ratio 1:2.5

$$\text{Volume of catalyst in 1}^{\text{st}} \text{ bed} =27.81/3.5 =7.94$$

$$\text{Volume of catalyst in 2}^{\text{nd}} \text{ bed} = 27.81- 7.94 =19.87$$

$$\text{Bulk density of catalyst} = 1200 \text{ Kg/m}^3$$

$$\text{Weight of catalyst in 2}^{\text{nd}} \text{ bed} =19.87*1200$$

$$= 23844 \text{ tonne}$$

$$\text{Weight of catalyst in 1}^{\text{st}} \text{ bed} = 7.94*1200 = 9528 \text{ tonne}$$

Now giving allowance for space for gas movement

Upward and downward and insulation be 0.5 m

Hence diameter of reactor becomes $=2.06-0.5 = 1.56$ m

Pressure at which reactor works $= 200 \text{ Kg/m}^2$

Let factor of safety $= 20\%$

Design pressure $= 1.2 * 200$

$$= 240 \text{ Kg/m}^2$$

We use carbon steel with internal lining of titanium

Allowable stress $= 66,000$ psi {hesse and ruston}

$$= 4494 \text{ Kgf/cm}^2$$

NOW allowable stress $= p_D (1/k_2 - 1)$ from (Hesse)⁽⁹⁾

Where $p_D =$ design pressure

$$4494 = 240 (1/k_2 - 1)$$

$$K = 1.0337$$

$$D^o = K D_i$$

$$D^o = 1.0337 * 2.06$$

$$D^o = 2.129 \text{ m}$$

Thickness of shell $= (2.129 - 2.06) / 2$

$$= 32 \text{ mm}$$

Operating pressure: 200 Kg/m^2

Design pressure: 240 Kg/m^2

Temperature : $673 - 698 \text{ k}^\circ$

Allowable stress: 4494 Kgf/cm^2

Diameter inside of converter : 2.06 m

Bulk density of catalyst: 1200 Kg/m^3

Volume : 27.81 m^3

Thickness of shell: 34mm

Volume of 1st catalyst bed : 7.94 m³

Volume of 2nd catalyst bed : 19.87 m³

Chapter four
Process control

4.1 Introduction

Ammonia is one of the top 10 chemicals produced today. Most ammonia is produced by steam methane reforming of a natural gas feedstock. 85% of the ammonia produced is used for fertilizer in one form or another.

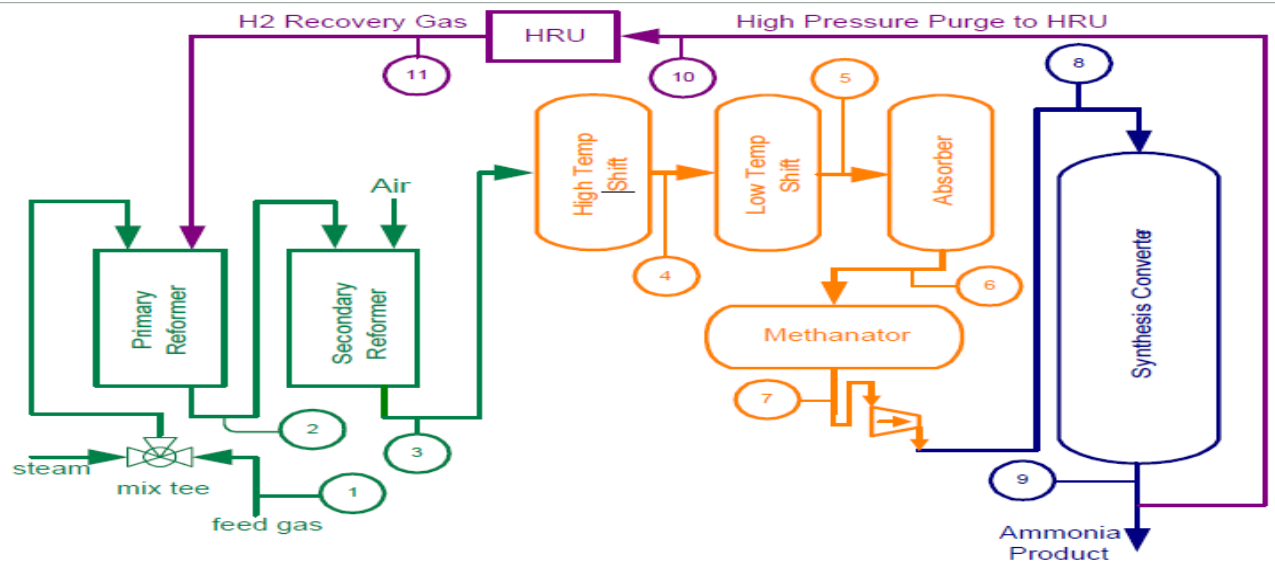
Other industrial uses for ammonia include fibers, plastics, coatings and resins. In looking to improve process efficiency and cut costs ammonia manufacturers have adopted a closed loop control strategy.

The key to the success of this strategy is the Extrel CMS MAX300-IG Process Mass Spectrometer. The MAX300-IG is a rugged Quadrupole mass spectrometer utilizing sophisticated electronics and offering fully automatic operation.

The MAX300-IG is a universal analyzer offering stable real time data acquisition and a dynamic range of 10 ppb to 100%. The exact concentrations of the stream components are immediately available for integration into the control scheme.

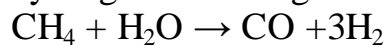
Due to its reliability, speed and accuracy the MAX300-IG provides unsurpassed performance in ammonia production. The MAX300_IG allows for tight control of four important process variables: the steam to carbon ratio, the H₂:N₂ ratio, the methane slippage and the process inert gases.

Maintaining tight control of these variables results in three significant economic benefits to ammonia producers: lower fuel costs, higher yields and an overall optimized efficiency. Figure 1 is a simple diagram of the ammonia process indicating the typical streams analyzed by the MAX300-IG.



Feed Gas

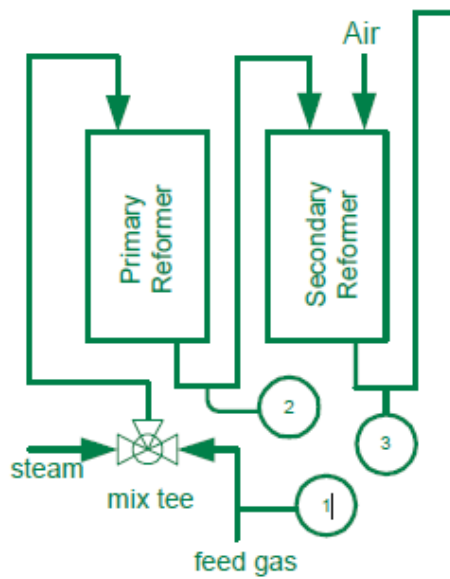
In the Primary and Secondary Reformers, a hydrocarbon feed stream and steam are mixed and passed over a reformer containing a nickel-based catalyst at 1500F. This results in the formation of carbon monoxide and hydrogen according to the following equation.



Controlling Steam Generation Costs through Control of the Steam to Carbon Ratio to a tolerance of $\pm 0.02\%$

The formation of carbon, known as coking, is detrimental to the catalyst. To prevent coking, an excess amount of steam is mixed with the feed gas than what the reaction requires. A minimum ratio of 3.0 lb. moles of steam to 1.0 lb. moles of carbon, i.e., 3:1 should be maintained. However, to avoid dropping below this ratio and degrading the catalyst, plants normally operate with generously elevated steam to carbon ratios.

The generation of steam is one of the most expensive processes in ammonia production. Typically, steam is produced at a level 4 times that of the ammonia product. Tight control of the steam to carbon ratio can significantly decrease production costs.



The MAX300-IG performs rapid analysis of the feed stream for hydrocarbon concentration and calculates the BTU value.

This information is used to control the steam flow; allowing rapid adjustments and avoiding potential converter problems. The MAX300-IG typically analyzes hydrocarbons up to and including hexane in a Natural Gas Feed Stream.

The instrument has the ability to analyze the butane and pentane isomers separately. This results in better accuracy for the calculated BTU value. Using the MAX300-IG as part of a closed loop strategy, Steam to Carbon ratio tolerances of ± 0.02 have been demonstrated.

Using the electron multiplier dual detector option, the MAX300-IG can also analyze the Feed Gas stream for the presence of sulfur containing components such as hydrogen sulfide and mercaptans. Sulfur components are poisons and will de-activate the catalyst.

Feed Gas	
Component	% Conc.
Nitrogen	0.30
Carbon Dioxide	2.30
Methane	81.00
Ethane	9.60
Propane	4.10
n-Butane	1.10
i-Butane	0.40
n-Pentane	0.35
i-Pentane	0.15
Hexane	0.05
Hydrogen Sulfide	5 ppm

4.2 Primary Reformer Effluent and Secondary Reformer Effluent:

The amount of un-reacted methane in the Primary Reformer Effluent and Reformer Effluent is referred to as methane slippage. Methane slippage, also called leakage, is an indication of the reformer efficiency in the conversion of methane to hydrogen and carbon dioxide.

The control objective is to maintain the methane slippage out of the secondary reformer at a very low and constant value, typically 0.3% methane or less. The analysis of the Primary and Secondary Reformers requires both a highly sensitive and very stable sensor because of the complex chemical makeup.

A wide dynamic range is also required as the methane conversion starts at 90% methane and ends at 0.3% methane. The sensitivity and the stability of the MAX300-IG enables control tolerances of 0.05% to be achieved.

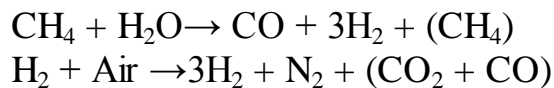
These tight tolerances allow the operator to minimize the changes in methane slippage thereby maximizing yield and lengthening equipment life. Increases in run time and equipment life of up to 20% have been observed in plants utilizing closed loop control with an Extrel CMS MAX300-IG.

Maximizing Yield and Equipment Life through Unsurpassed Control of Methane Slippage

Air is used as the source of the nitrogen required for the ammonia reaction. Air is introduced into the process at the secondary reformer **in**

an amount to yield an exit gas that has a 3:1 hydrogen to nitrogen ratio. The oxygen in the air reacts with the hydrocarbon feedstock in combustion and supplies the energy to reform the remainder of the gas. The secondary reformer reaction increases the hydrogen content and oxidizes most of the CO to CO₂.

CO₂ is preferred over CO as it is much more soluble in water and can easily be removed further along in the process. The Secondary Reformer Outlet also contains a small amount of methane and argon which enter the synthesis loop as inert gases.



Primary Reformer Effluent	
Component	% Conc.
Hydrogen	67.00
Nitrogen	1.50
Carbon Dioxide	11.50
Carbon Monoxide	8.00
Methane	12.00
Argon	0.10

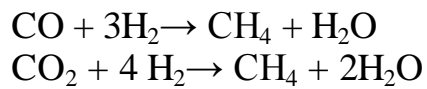
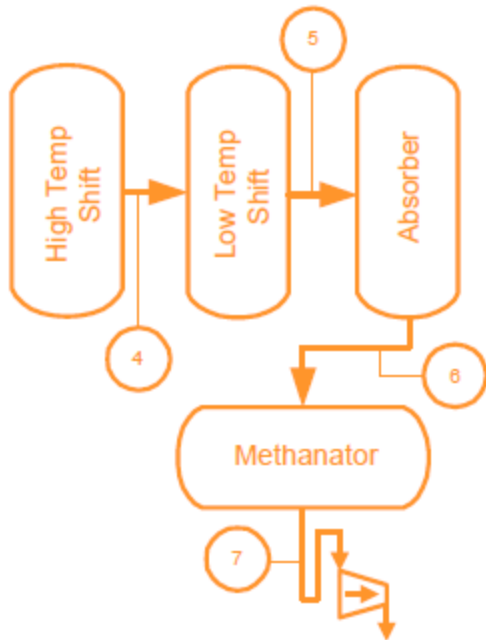
Secondary Reformer Effluent	
Component	% Conc.
Hydrogen	57.50
Nitrogen	22.50
Carbon Dioxide	8.50
Carbon Monoxide	12.00
Methane	0.30
Argon	0.30

4.3 CO₂ Absorber and Methanator:

The CO₂ is generally removed by absorber-regenerator units containing many types of absorbent including monoethanolamine solution (MEA), sulfinol, propylene carbonate and others. After this process, CO in the product gas is reduced to less than 100 ppm. CO₂ and CO collectively known as carbon oxides, are poisons for many types of catalysts.

The trace amounts of CO and CO₂ remaining after the absorber must be removed from the synthesis gas. This is done by converting them to methane over a nickel or ruthenium catalyst in the presence of hydrogen

in the Methanator. The amount of methane produced by this process is insignificant compared to the level of the un-reacted methane slip. After this process, the residual levels of the carbon oxides in the process gas will be less than 5 ppm.



Absorber Outlet	
Component	% Conc.
Hydrogen	65.33
Nitrogen	31.80
Carbon Dioxide	0.08
Carbon Monoxide	0.48
Methane	1.81
Argon	0.41

Methanator Outlet	
Component	% Conc.
Hydrogen	69.80
Nitrogen	28.00
Carbon Dioxide	< 5 ppm
Carbon Monoxide	< 5 ppm
Methane	1.70
Argon	0.30

Chapter five

Location ,Economic &Safety Consedration

5.1 Location and site considerations:

The site chosen for the plant chemical directly affect the success or failure of the manufacturing process. The basic principle is to have the factory site so that the cost of production and distribution of the product at the minimum , and this can only be achieved if it is taking a number of important factors into consideration during the design and selection process plant site.

The most important considerations to be followed in the selection of the plant site are:

1. Sources of raw materials
2. Consuming markets
3. energy resources
4. the climate
5. transport services
6. Water Resources
7. Waste disposal
8. Labor
9. Financial considerations

10. Prevention From Fire and floods

11. Proximity From Positions Industrial

5.1.1. Sourcing of raw materials

That one raw material processing of the most important factors that go into determining the plant site especially when large quantities required. The factory near the site of the raw materials are not allowed to reduce transportation expenses only, but to reduce the capital invested in storage services as well.

5.1.2. Common Markets

The viability of the delivery of the final product to the consumer quickly and the lowest possible cost no less important than site - built near the source of raw materials. So From the necessary that Be there is Comparison between Proximity From Items the source Crude and ease receipt Product to me The consumer. The choice Will depend On the quantity Required And the cost Transfer Articles Crude And the product. When Taking Marketing For the product Final The eyes consideration

Must that Calculates an account Sale And distribution Outputs Occasional Often What Be Factor at Profit Work.

5.1.3. Energy Sources

In most chemical factories need large fuel and energy be when coal and oil shale amounts required large site near the source of the processing is necessary , therefore the process is economical. When large amounts of electrical power is available , it is one of the very important factors to be taken into consideration, it is noted that in the case of the need for a large ownership of the energy it is better to be near the site of the hydroelectric generators.

5.1.4. Climate

It is possible that a large climate impact of the economic process. When Aldharov be harsh climate, the cost of personal protective buildings and services must be part of the total cost. In hot weather it may be dismissed in cooling towers, air - conditioning and possibly freeze coefficient of important considerations.

5.1.5. Transportation services

The choice of modes of transportation for both raw materials and the final product depends no doubt on the quality and quantity of the materials used , if possible, provide the three modes of transportation of any roads and population as rail and maritime transport. It is important that the plant site near either the railway station or docks.

5.1.6. Water Resources

Most industrial processes require large amounts of water not only for industrial process but also used for cooling processes, washing and steam generation processes. To make the process economical, the plant site or plant near a river or lake is an important and necessary.

5.1.7. Waste disposal

The water supply is linked to a large extent with the waste disposal problem, whether solid, liquid or gaseous where there are strict limitations on the issue of waste disposal and it changes according to industrial unit site.

It is necessary to study the stamina to be allowed for water, earth, air metastatic sites and in several cases the level of a test as a result of the center. The additional cost due to treatment outcomes emitted must be balanced with other appropriate site selection factors.

5.1.8. Employment

Although the trend is the increased use of automated equipment in the lab , a chemical addition to the angel of the assistant engineers , but there are several processes still need a large workforce especially when using a rotation system , it is important to choose the industrial unit site , taking into account the payment of wages prevailing rate specific region and industry competition as well as skilled and intelligent workers. In remote spokesman it should be calculated to provide housing and causes of social amenities for the workers expense.

5.1.9. Financial considerations

When trying to create a sophisticated industry in a particular area should provide the necessary and which have a significant impact in the selection of the plant site and financial incentives. It is also necessary to take the applicable tax system into consideration . When choosing the location of his family in the area of future expansion should his account , especially in the busy industrial areas.

5.1.10. Prevention of fires, floods

as such We mentioned Previously, for the purpose of Large amounts of cooling water for the chemical factories processing location is often near rivers and lakes .wallhakiqh that this site will be vulnerable to flooding so it is necessary to know the historic flood that has happened in that region. It is important to provide adequate services to prevent n fire in the region, as well as knowledge of its impact on neighboring sites. The Earth and its terrain with a price tag and buildings should take into consideration.

5.1.11. Proximity to industrial centers

The plant site near the important industrial centers be useful in terms of getting large numbers of manpower of different means of transport. Perhaps there are other industries adjacent to bid for materials and services at low prices , and there are several cases where materials processing manufacturers have directly.

- Due to this reason we find that the desired position is basra because

production of urea is used as raw material .

5.2 Safety and environmental consequences:

Causing ammonia in the incidence of infections and irritabilities flogging and nose and eyes and throat part upper from the device respiratory and where that Ammonia the source the main element Nitrogen necessary growth the plants aqueous it possible that contribute Ammonia at a plus enrichment food surface Aqueous stagnant or a slow syriacs especially those spaces of the specific content of nitrogen addition to that is the average of ammonia toxin for objects water even right now did not prove scientifically that ammonia from articles casing cancer.

Returns Problems Environmental Which Cause In which Ammonia to me Feature Solubility water. The tendency Leak at Case Gaseous The Cause Ammonia at Happening Damage seriously life wilderness when Absorbed at Currents Water it causes Damage Fish To Fish water cold As well As for at Air Vtaathd Ammonia With Ions Sulfates And melt at waters Rain To return Quickly to me the soil The flats Aqueous

Safety Precautions for Ammonia:

- CORROSIVE, COMPRESSED GAS. May also be an EXPLOSION HAZARD, especially in confined spaces. Engineering controls, proper training, protective equipment requirements and personal hygiene measures are essential.
- Never work alone with this chemical.
- Unprotected persons should avoid all contact with this chemical, including contaminated equipment.
- Do not use with incompatible materials such as oxidizing agents (e.g., nitrogen oxide), halogens (e.g., chlorine, fluorine) and heavy metals (e.g., mercury, silver).
- Shut flow off at cylinder valve and not just at the regulator after use. Replace outlet caps or plugs and cylinder caps as soon as cylinder is disconnected from equipment.
- Make sure cylinders are labeled clearly. Avoid damaging cylinders. Move cylinders by hand truck or cart designed for that purpose.

- Keep empty cylinders under slightly positive pressure. Do not use cylinders as rollers or for any other purpose than to contain the gas as supplied.
- Follow handling precautions on Material Safety Data Sheet. Have suitable emergency equipment for fires, spills and leaks readily available. Practice good housekeeping. Maintain handling equipment. Comply with applicable regulations.
-

5.3 Economic and feasibility considerations:

5.3.1. cost estimation of Absorption column:

Height = 5 m

Diameter = 0.87 m

Pressure = 30 bar

Material of construction : Carbon steel

Cost = 8000\$

Pressure factor = 1.6

Material factor = 1

Cost = $8000 \times 1.6 \times 1 = 12800$ \$ in 2004

Cost in 2016 = (cost in 2004) \times (cost index in 2015/cost index in 2004)

Cost index in 2004 = 480 \$

Cost index in 2015 = 1081 \$

Cost in 2016 = $12800 \times (1081/480)$

Cost in 2016 = 28826.66667 \$
= 37186400 I.D

5.3.2 Cost estimation of Heat exchanger :

Area = 210 m²

Cost = 7500 \$

Pressure factor = 1

Material factor = 1

Cost = $7500 \times 1 \times 1 = 7500$ \$ in 2004

Cost in 2016 = cost in 2004 \times (cost index in 2015/cost index in 2004)

Cost index in 2004 = 480\$

Cost index in 2015 = 1081\$

Cost in 2016 = $7500 \times (1081/480)$

$$\begin{aligned}\text{Cost in 2016} &= 16890.625 \$ \\ &= 21788906.25 \text{ I.D}\end{aligned}$$

5.3.3 cost of estimation of reactor

$$D = 2.04 \text{ m}$$

$$H = 8.24 \text{ m}$$

$$P = 200 \text{ bar}$$

$$\text{Material factor} = 1$$

$$\text{Pressure factor} = 7.8$$

$$\text{Cost} = 12200 \$$$

$$\text{Cost} = 12200 * 7.8 * 1 = 95160 \$ \text{ in 2004}$$

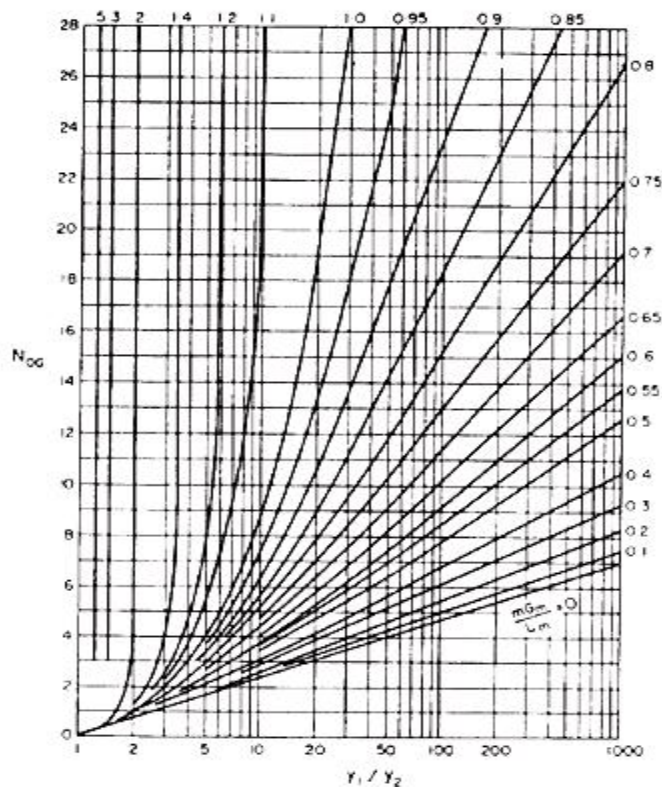
$$\text{Cost in 2016} = \text{cost in 2004} * (\text{cost index in 2015} / \text{cost index in 2004})$$

$$\text{Cost index in 2004} = 480 \$$$

$$\text{Cost index in 2015} = 1081 \$$$

$$\begin{aligned}\text{Cost in 2016} &= 95160 * (1081 / 480) \\ &= 214308.25 \$ \\ &= 276457642.5 \text{ I.D}\end{aligned}$$

Appendix A:



Figure(3.1) number of transfer unit NOG as a function y_1/y_2 with mG_m/L_m as parameter

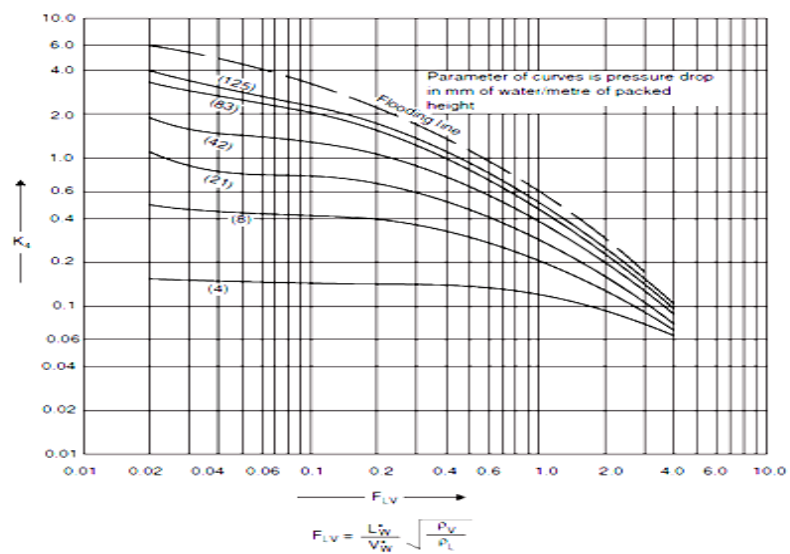


Figure (3.2) Generalized pressure drop correlation ,adapted from a figure by the Norton CO. with permission

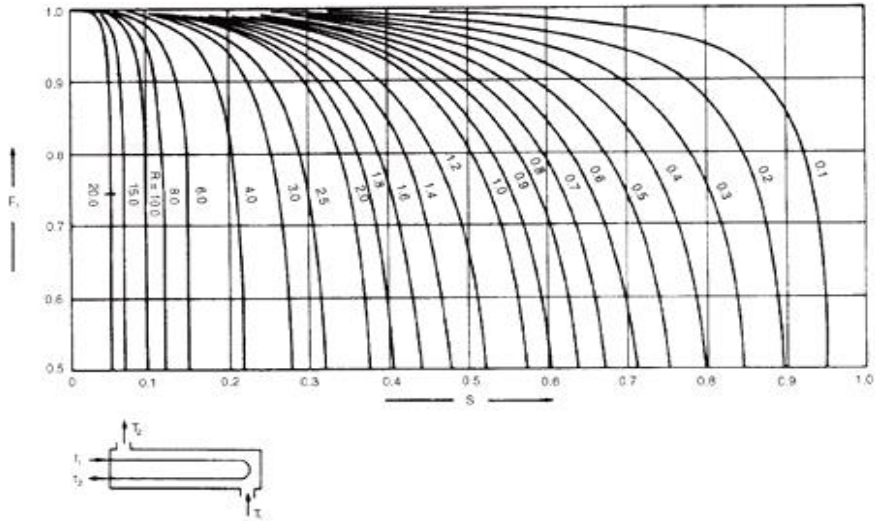


Figure (3.3) Temperature correction factor ,one shell pass ,two or more even tube passes

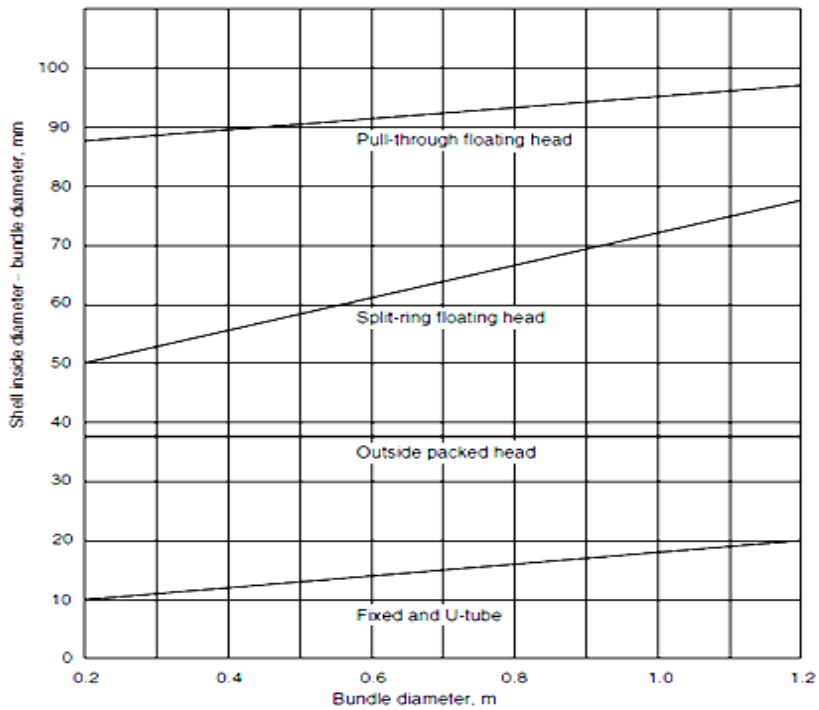


Figure.(3.4) shell bundle clearance

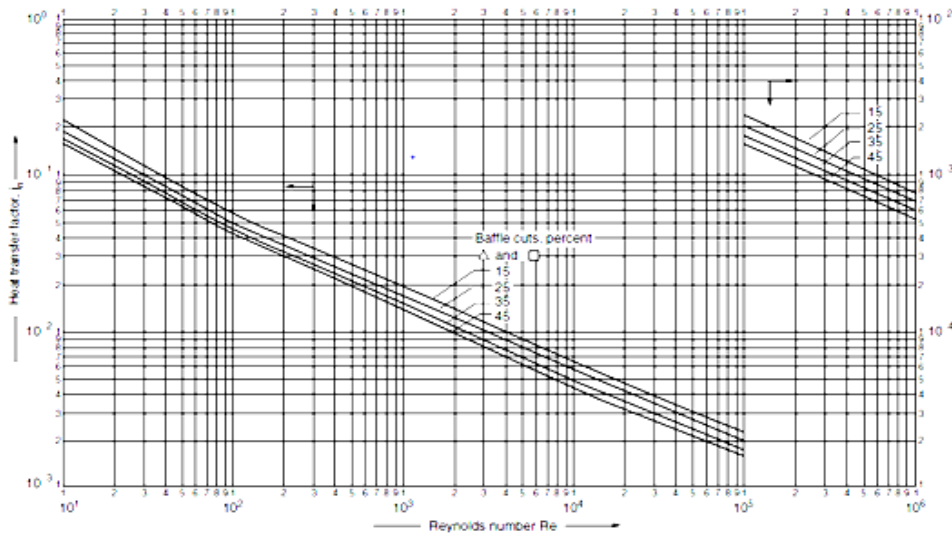


figure (3.5) Shell side heat transfer factor ,segmental baffles

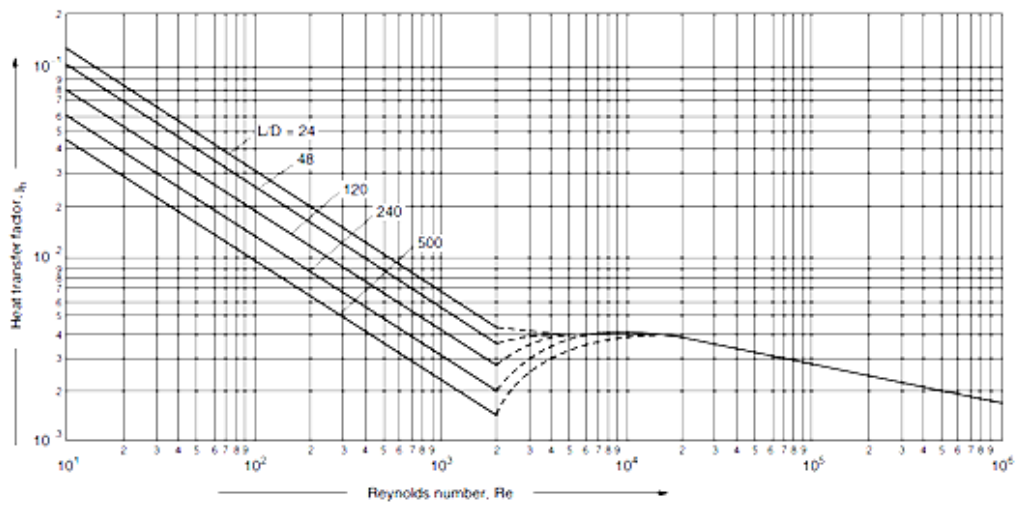


Figure (3.6) Tube side heat transfer factor

Appendix B:

System	Pressure kPa	Column dia, m	Packing type	Packing size, mm	HTU m	HETP m
<i>Absorption</i>						
Hydrocarbons	6000	0.9	Pall	50		0.85
NH ₃ -Air-H ₂ O	101	—	Berl	50	0.50	
Air-water	101	—	Berl	50	0.50	
Acetone-water	101	0.6	Pall	50		0.75
<i>Distillation</i>						
Pentane-propane	101	0.46	Pall	25		0.46
IPA-water	101	0.46	Int.	25	0.75	0.50
Methanol-water	101	0.41	Pall	25	0.52	
	101	0.20	Int.	25		0.46
Acetone-water	101	0.46	Pall	25		0.37
	101	0.36	Int.	25		0.46
Formic acid-water	101	0.91	Pall	50		0.45
Acetone-water	101	0.38	Pall	38	0.55	0.45
	101	0.38	Int.	50	0.50	0.45
	101	1.07	Int.	38		1.22
MEK-toluene	101	0.38	Pall	25	0.29	0.35
	101	0.38	Int.	25	0.27	0.23
	101	0.38	Berl	25	0.31	0.31

Pall = Pall rings, Berl = Berl saddles, Int. = Intalox saddles

Table (3.1) typical packing efficiencies

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