# DESIGNING OF DEPROPANIZER TRAY COLUMN USING ASPEN HYSYS AS A SIMULATOR

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# "DESIGNING OF DEPROPANIZER TRAY COLUMN USING ASPEN HYSYS AS A SIMULATOR"

A dissertation submitted in partial fulfillment of the requirements for the Degree of Bachelor of Technology

(Applied Petroleum Engineering)

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### **CERTIFICATE**

This is to certify that the work contained in this thesis titled "DESIGNING OF A DEPROPANIZER TRAY COLUMN USING ASPEN HYSYS AS A SIMULATOR" has been carried out by JAI DUBEY & MAYANK KANODIA under my supervision and has not been submitted elsewhere for a degree.

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#### **ABSTRACT**

Distillation utilizes vapor and liquid phases at essentially the same temperature and pressure for the coexisting zones. Plates or trays are used to bring the two phases into intimate contact. Trays are stacked one above the other and enclosed in a cylindrical shell to form a column. The overall flow pattern in a distillation column provides countercurrent contacting of vapor and liquid streams on all the trays through the column. These phases on a given tray approach thermal, pressure, and composition equilibriums to an extent dependent upon the efficiency of the contacting tray. Column process design specifies the separation, and sets column and utility pressure, reflux, stages, and feed point. These in turn yield internal flows and reboiler and condenser duties.

Designing of Depropanizer column deals in determining various parameters like number of equilibrium stages, column internals, thermodynamic properties of fluid, etc. Initial data for designing column is taken form a refinery. Column is designed using Fenske Underwood Gilliland theoretical method. Consecutively, calculation of column internals is also done. A simulation for depropanizer column is done using Aspen Hysys as simulating software. Thus, a correlating study is done between the theoretical and simulation method. A complete understanding of software is done under the able guidance of mentor before using it. A layout model of column is also prepared based upon the data obtained. This project provides the designing of the depropanizer column and use of simulation software to validate the calculated data.

3.0



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# **ABBREVIATIONS**

EOS Equation of state

RK Redlich Kwang

VLE Vapor Liquid Equilibrium

PR Peng Robinson

PFD Process flow diagram

SB Smith Brinkley

FUG Fenske Underwood Gilliland



# **NOMENCLATURE**

$[x_i/x_r]$	The ratio of the concentration of any component i to the concentration of a
	reference component r
$x_{LK}$	Light key concentrations
$x_{HK}$	Heavy key concentrations
$X_{f,HK}$	Concentration of the heavy key in the feed
$X_{f,LK}$	Concentration of the light key in the feed
$X_{\text{d, HK}}$	Concentration of the heavy key in the top product
$X_{b, LK}$	Concentration of the light key if in the bottom product
$N_{\text{m}}$	Minimum number of stages at total reflux, including the reboiler
$N_{r}$	Number of stages above the feed, including any partial condenser
$N_s$	Number of stages below the feed, including the reboiler
$N_{\text{min}}$	Minimum number of stages
$\alpha_i$ .	Average relative volatility of the component / with respect to the reference
	component
$\alpha_{LK}$	Average relative volatility of the light key with respect to the heavy key
q	Thermal condition of the feed
$R_{\text{m}}$	Minimum reflux
В	Molar flow bottom product
D	Molar flow top product
$\mathbf{u}_{\mathbf{v}}$	Maximum allowable vapor velocity, based on the gross (total) column cross -
	sectional area, m/s
$\mathbf{l_t}$	Plate spacing, m
$D_c$	Column diameter, m
$V_{\rm w}$	Maximum vapor rate, kg/s
Ac	Total column cross-sectional area, m <sup>2</sup>
$A_{d}$	Cross-sectional area of downcomer, m <sup>2</sup>
$A_n$	Net area available for vapor-liquid disengagement, m <sup>2</sup>
A <sub>a</sub>	Active or bubbling, area, m <sup>2</sup>
$A_h$	Hole area, the total area of all the active holes, m <sup>2</sup>

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A<sub>p</sub> Perforated area (including blanked areas), m<sup>2</sup>

 $A_{ap}$  The clearance area under the downcomer apron,  $m^2$ 

u<sub>f</sub> Flooding vapor velocity, m/s

F<sub>LV</sub> Liquid-vapor flow factor

Lw Liquid mass flow-rate, kg/s

Liquid flow rate in downcomer, kg/s

V<sub>w</sub> Vapor mass flow-rate, kg/s

u<sub>h</sub> Minimum vapor velocity through the holes(based on the hole area), m/s

d<sub>h</sub> Hole diameter, mm

ρ<sub>V</sub> Density of vapor, Kg/m<sup>3</sup>

ρ<sub>L</sub> Density of liquid, Kg/m<sup>3</sup>

h<sub>w</sub> Height of the weir, mm

h<sub>ow</sub> Depth of the crest of liquid over the weir, mm

h<sub>d</sub> The dry plate drop, mm liquid

h<sub>r</sub> Residual head loss, mm liquid

h<sub>t</sub> Total plate pressure drop, mm liquid

h<sub>b</sub> Downcomer back-up, measured from plate surface, mm

h<sub>dc</sub> Head loss in the downcomer, mm

h<sub>ap</sub> Height of the bottom edge of the apron above the plate, mm

h<sub>bc</sub> Clear liquid back-up, mm

t<sub>r</sub> Residence time, sec

C<sub>o</sub> Orifice coefficient

 $\Delta P_t$  Total plate pressure drop, Pa (N/m<sup>2</sup>)

l<sub>w</sub> Weir length, mm

l<sub>p</sub> Hole pitch, mm

T<sub>C</sub> Critical temperature, °C

T<sub>R</sub> Residual temperature, °C

P<sub>C</sub> Critical pressure, psi

H<sup>IG</sup> Ideal gas enthalpy calculated at temperature, T

S<sup>IG</sup> Ideal gas entropy calculated at temperature, T

C<sub>P</sub><sup>IG</sup> Ideal gas heat capacity calculated at temperature, T

Φ<sub>i</sub> Fugacity coefficient



# CHAPTER 1 INTRODUCTION



## 1. INTRODUCTION

#### 1.1 OVERVIEW OF DISTILLATION

Distillation is simply defined as a process in which a liquid or vapor mixture of two or more substances is separated into its component fractions of desired purity, by the application and removal of heat. The process is based on the fact that the vapor of a boiling mixture will be richer in the components that have lower boiling points. Hence, when this vapor is cooled and condensed, the condensate will contain more volatile components. At the same time, the original mixture will contain more of the less volatile material. The primary equipment employed in the process of distillation are distillation columns, which are designed to achieve this separation efficiently.

The best way to reduce operating costs of existing units is to improve their efficiency and operation via process optimization and control. To achieve this improvement, a thorough understanding of distillation principles and how distillation systems are designed is essential. As stated, distillation is the process of heating a liquid until some of its ingredients pass into the vapor phase, and then cooling the vapor to recover it in liquid form by condensation. The main purpose of distillation is to separate a mixture. If the difference in boiling points between two substances is great, complete separation may be easily accomplished by a single-stage distillation. If the boiling points differ only slightly, many redistillations may be required. In the simplest mixture of two mutually soluble liquids with similar chemical structures, the readiness to vaporize of each is undisturbed by the presence of the other. For example, would be halfway between the boiling points of the pure substances, and the degree of separation achieved by a single distillation would depend only on each substance's readiness to vaporize at this temperature. This simple law was first stated by 19th- century by the French chemist Frangois Marie Raoult (known as Raoult's law).

The term "still" is applied only to the vessel in which liquids are boiled during distillation, but the term is sometimes applied to the entire apparatus, including the fractionating column, the condenser, and the receiver in which the distillate is collected. If a water and alcohol distillate is returned from the condenser and made to drip down through a long column a series of plates, and if the vapor, as it rises to the condenser, is made to bubble through this liquid at each plate,



the vapor and liquid will interact so that some of the water in the vapor condenses and some of the alcohol in the liquid vaporizes. The interaction at each plate is equivalent to a redistillation. This process is referred to by several names in the industry; namely rectification, fractionation, or fractional distillation.

If two insoluble liquids are heated, each is unaffected by the presence of the other and vaporizes to an extent determined only by its own nature. Such a mixture always boils at a temperature lower than is true for either substance alone. This effect may be applied to substances that would be damaged by overheating if distilled in the usual fashion. Substances can also be distilled at temperatures below their normal boiling points by partially evacuating the still. The greater the vacuum, the lower is the distillation temperature.

#### **1.2 BASIC COMPONENTS OF DISTILLATION COLUMN**

There are a variety of configurations for distillation columns, each designed to perform specific types of separations. The two major types are batch and continuous columns. In a batch operation, the feed to the column is introduced batch-wise. That is, the column is charged with 'batch' and then the distillation process is conducted. When the desired separation is achieved, a next batch of feed is introduced. In contrast, continuous columns process a continuous feed stream. They are capable of handling high throughputs and are more common of the two types. Continuous columns can be further classified according to:

- The nature of the feed that they are processing (binary column feed contains only two components, and multi-component column feed contains more than two components);
- The number of product streams they have (multiproduct column column has more than two product streams);
- Where the extra feed exits when it is used to help with the separation (extractive distillation

   where the extra feed appears in the bottom product stream , and azeotropic distillation where the extra feed appears at the top product stream );
- The type of column internals (tray column where trays of various designs are used to hold up the liquid to provide better contact between vapor and liquid, and hence achieve better separation, and the packed column where instead of trays, packings are employed to effect contact between vapor and liquid).



There are several important components in a distillation column, each of which is used either to transfer heat energy or enhance mass transfer. The major components in a typical distillation are:

- Vertical shell where the separation of liquid components is carried out,
- Column internals such as tray, plates and/or packings which are used to enhance component separations,
- Reboiler to provide the necessary vaporization for the distillation process
- Condenser to cool and condense the vapor leaving the top of the column
- Reflux drum to hold the condensed vapor from the top of the column
- Liquid (reflux) is recycled back to the column.

The column internals are housed within a vertical shell, and together with the condenser and reboiler, constitute a distillation column. The liquid mixture that is to be processed is called the feed. The feed introduced usually somewhere near the middle of the column to a tray known as the feed tray. The feed tray divides the column into a top (enriching or rectification) section and a bottom (stripping) section. The feed flows down the column where it is collected at the bottom in the reboiler. Heat is supplied to the reboiler to generate vapor. The source of heat input can be any suitable fluid, although in most chemical plants this is normally steam. In refineries, the heating source may be the output streams of other columns. The vapor raised in the reboiler is re-introduced into the unit at the bottom of the column. The liquid removed from the reboiler is known as the bottoms product or simply, the bottoms. The vapor travels up the column, and as it exits the top of the unit, it is cooled by a condenser. The condensed liquid is stored in a holding vessel known as the reflux drum. Some of this liquid is recycled back to the top of the column and this is called the reflux. The condensed liquid that is removed from the system is known as the distillate or top product.

#### 1.3 CONTINUOUS DISTILLATION: PROCESS DESCRIPTION

The separation of liquid mixtures by distillation depends on differences in volatility between the components. The greater the relative volatilities, the easier are the separation. The basic equipment required for continuous distillation is shown in Figure 1. Vapor flows up the column and liquid counter-currently down the column. The vapor and liquid are brought into contact on



plates, or packing. Part of the condensate from the condenser is returned to the top of the column to provide liquid flow above the feed point (reflux), and part of the liquid from the base of the column is vaporized in the reboiler and returned to provide the vapor flow. In the section below the feed, the more volatile components are stripped from the liquid and this is known as the stripping section. Above the feed, the concentration of the more volatile components is increased and this is called the enrichment, or more commonly, the rectifying section.

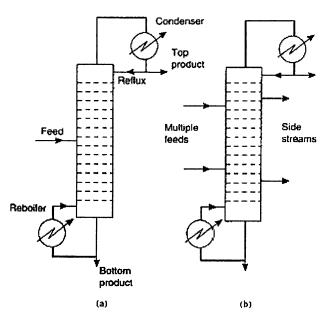


Figure 1 Distillation column (a) Basic column (b) Multiple feeds and side streams

Figure 1(a) shows a column producing two product streams, referred to as tops and bottoms, from a single feed. Columns are occasionally used with more than one feed and with side streams withdrawn at points up the column, Figure 1(b). This does not alter the basic operation, but complicates the analysis of the process, to some extent. If the process requirement is to strip a volatile component from a relatively non-volatile solvent, the rectifying section may be omitted, and the column would then be called as stripping column. In some operations, where the top product is required as a vapor, only sufficient liquid is condensed to provide the reflux flow to the column, and the condenser is referred to as a partial condenser. When the liquid is totally condensed, the liquid returned to the column will have the same composition as the top product. In a partial condenser the reflux will be in equilibrium with the vapor leaving the condenser. Virtually pure top and bottom products can be obtained in a single column from a binary feed, but where the feed contains more than two components; only a single "pure" product can be produced, either from the top or bottom of the column. Several columns will be



needed to separate a multicomponent feed into its constituent parts. Continuous distillation is of many types but here we are considering only multi component distillation.

#### **1.4 MULTICOMPONENT DISTILLATION**

A tower comprised of rectifying (above the feed) and stripping (below the feed) sections is capable of making a more or less sharp separation between two products or pure components of the mixture, that is, between the light and heavy key components. Key components are the two components in the feed mixture whose separation is specified. The more volatile of these components is the light key (Propane), and the less volatile is the heavy key (Butane). Other components are termed nonkeys. The key components appear to a significant extent in both overhead and bottom products. Light nonkeys end up almost exclusively in the overhead product, and heavy nonkeys end up almost exclusively in the bottom product in many separations, components are present whose relative volatilities are intermediate between the light key and the heavy key. These components are termed intermediate keys or distributed keys. Intermediate keys are split between the top and bottom products. We are considering a depropanizer column over here.

#### 1.5 DEPROPANIZER COLUMN

The bottoms from the de-ethanizer serves as the principal feed to the depropanizer. A second feed stream to this column consists of the bottoms liquid from the condensate stripper for the fourth and fifth compression stages. The primary function of this column is propylene and propane recovery from these two feeds. The column operating pressure typically will be 240 to 340 psia, which is sufficient to condense the overhead vapor with cooling water or ambient air. The overhead from this column contains all of the propane present in the feed, as well as some butane that are stripped to provide complete recovery of propane. Ethane content in the distillate is negligible due to the over stripping commonly carried out in the preceding de-ethanizer.

A depropanizer normally contains 35 to 55 actual trays and has a uniform diameter. In these systems, the surface tension of the liquid phase is below 6 dyne/cm, the liquid density is near 30 lb/ft<sup>3</sup>, and the vapor density is about 8% of the liquid density. Under such conditions, the



downcomer residence time required can be a significant factor in the specification of the tower diameter for a tray column. The downcomer normally is designed so that the froth height is no more than 70% of the tray spacing; therefore, downcomer area must be large to avoid the need for excessively tall columns.

#### **1.6 ADVANTAGES OF TRAY COLUMN**

#### 1.6.1 Factors Favoring Tray Columns

The following factors generally favor trays compared to either random or structured packings:

#### • Solids:

Trays can handle solids a lot easier than packed columns. Both gas and liquid velocities are often an order of magnitude higher on a tray than through packings. These high liquid and gas velocities provide a sweeping action that keeps tray openings and perforations clear. Solids tend to accumulate in the void of packed column. There are fewer locations where solids can be deposited in a tray column.

Further, packed towers need liquid distributors and plugging in these has been a common trouble spot. Cleaning trays is easier than cleaning random packings, while cleaning structured packings is practically impossible.

#### • High Liquid rates:

Multipass trays effectively lower the liquid load "seen" by each part of the tray. The capacity of packings, especially structured, tends to rapidly fall off at high liquid rates. It is often more economical to handle high liquid rates in tray columns.

#### • Large diameter:

Packings are prone to maldistribution problems in large diameter columns. These problems are far less in plate columns.



#### • Complex columns:

Inter-reboilers, inter-condensers, cooling coils, and side drawoffs are more easily incorporated in tray than in packed columns. In packed columns, every complexity requires additional distribution and/or liquid collection equipment.

#### • Feed composition variation:

One way of allowing for design uncertainties and feedstock variation is by installing alternate feed points. In packed columns, every alternate feed point requires expensive distribution equipment.

#### • Performance prediction:

There is greater uncertainty in predicting packed column performance. Greater overdesign is often required.

#### • Chemical reaction/absorption:

By using high weirs, trays are capable of providing greater residence time for absorption or chemical reaction than packing.

#### • Weight:

Tray columns usually weigh less than packed columns. This saves on the cost of foundations, supports, and column shell.

#### • Intermittent operation:

When temperature is either lower or higher than atmospheric, intermittent operation repeatedly expands and contracts the shell. This may crush the packings or damage the shell in a packed column, but is easy to accommodate for in tray columns.



# CHAPTER 2 LITERATURE REVIEW



# 2. <u>LITERATURE REVIEW</u>

#### 2.1 MULTICOMPONENT DISTILLATION METHOD

Some approximate calculation methods for the solution of multicomponent, multistage separation problems continue to serve useful purposes even though computers are available to provide more rigorous solutions. The available phase equilibrium and enthalpy data may not be accurate enough to justify the longer rigorous methods. In design and optimization studies, a large number of cases can be worked quickly and cheaply by an approximate method to define roughly the optimum specifications, which can then be investigated more exactly with a rigorous method.

Two approximate multicomponent shortcut methods for simple distillation are the Smith-Brinkley (SB) method, which is based on an analytical solution of the finite-difference equations that can be written for staged separation processes when stages and interstage flow rates are known or assumed and the Fenske-Underwood-Gilliland (FUG) method, which combines Fenske's total-reflux equation and Underwood's minimum-reflux equation with a graphical correlation by Gilliland that relates actual column performance to total- and minimum- reflux conditions for a specified separation between two key components. Both methods work best when mixtures are nearly ideal.

#### 2.1.1 Fenske-Underwood-Gilliland (FUG) Method

The first step in the design of distillation equipment is specification of light and heavy key components. Then the specific operating conditions and equipment size are established, ultimately on the basis of an economic balance or simply by exercise of judgment derived from experience. The design parameters that need to be determined include intermediate ones such as limiting reflux and trays that are needed for establishing a working design. These design parameters are the following:

- Minimum number of theoretical trays
- Distribution of nonkeys between the overhead and bottoms products
- Minimum reflux
- Operating reflux



- Number of theoretical trays
- Location of the feed tray
- Tray efficiencies

These shortcut methods assume constant molar overflow in the rectifying and stripping zones and constant relative volatilities, which may be taken at the conditions of the feed tray or as a geometric mean of the values at the top and bottom of the column. Since the top conditions are not known completely in advance, evaluation of a mean relative volatility is an iterative process that can be started with the value at the feed tray or at the feed condition. Particular modes of variation of  $\alpha$  sometimes are assumed.

#### 2.1.1.1 Minimum Number of Stages

The Fenske equation (Fenske, 1932) can be used to estimate the minimum stages required at total reflux. The equation applies equally to multicomponent systems and can be written as:

$$\left[\frac{x_i}{x_r}\right]_d = \alpha_i^{N_m} \left[\frac{x_i}{x_r}\right]_b$$

Where,

 $[x_i/x_r]$  = the ratio of the concentration of any component i to the concentration of a reference component r, and the suffixes d and b denote the distillate (d) and the bottoms (b),

 $N_m$  = minimum number of stages at total reflux, including the reboiler

 $\alpha_i$  = average relative volatility of the component / with respect to the reference component.

Normally the separation required will be specified in terms of the key components, and the above equation can be rearranged to give an estimate of the number of stages:

$$N_{m} = \frac{\log \left[\frac{x_{LK}}{x_{HK}}\right]_{d} \left[\frac{x_{HK}}{x_{LK}}\right]_{b}}{\log \alpha_{LK}}$$



Where,  $\alpha_{LK}$  is the average relative volatility of the light key with respect to the heavy key, and  $x_{LK}$  and  $x_{HK}$  are the light and heavy key concentrations. The relative volatility is taken as the geometric mean of the values at the column top and bottom temperatures. To calculate these temperatures initial estimates of the compositions must be made, so the calculation of the minimum number of stages by the Fenske equation is a trialand – error procedure. If there is a wide difference between the relative volatilities at the top and bottom of the column the use of the average value in the Fenske equation will underestimate the number of stages. In these circumstances, a better estimate can be made by calculating the number of stages in the rectifying and stripping sections separately; taking the concentration as the base concentration for the rectifying section and as the top concentration for the stripping section, and estimating the average relative volatilities separately for each section. This procedure will also give an estimate of the feed point location.

#### 2.1.1.2 Minimum Reflux

The method of Underwood employs auxiliary parameters derived from the equation:

$$\sum_{i=1}^{C} \frac{\alpha_i x_{Fi}}{\alpha_i - \theta} = 1 - q$$

Where, q is the thermal condition of the feed and the summation extends over all the components in the feed. The only roots required are those in numerical value between the relative volatilities of the light and heavy keys. For instance, if there is one distributed component, subscript dk, the required roots  $\theta_1$  &  $\theta_2$ 

$$\alpha_{1k} > \theta_1 > \alpha_{dk},$$
  

$$\alpha_{dk} > \theta_2 > \alpha_{hk}.$$

Then the minimum reflux and the distribution of the intermediate component are found from the two equations that result from substitution of the two values of  $\theta$  into Underwood's second equation:



$$R_m + 1 = \frac{1}{D} \sum \frac{\alpha_i d_i}{\alpha_i - \theta}.$$

The number of values of  $\theta=1$  plus the number of components with relative volatilities between those of the light and heavy keys. When there is no distributed component, the equation below may be used in terms of mole fractions and only a single form is needed for finding the minimum reflux,

$$R_m + 1 = \sum \frac{\alpha_i x_{iD}}{\alpha_i - \theta}.$$

Occasionally the minimum reflux calculated by this method comes out a negative number. That, of course, is a signal that some other method should be tried, or it may mean that the separation between feed and overhead can be accomplished in less than one equilibrium stage.

#### 2.1.1.3 Operating Reflux

The operating reflux is an amount in excess of the minimum that ultimately should be established by an economic balance between operating and capital costs for the operation. In many cases, however, the assumptions  $R = 1.5R_m$ , often is close to the optimum and is used without further study unless the installation is quite a large one.

#### 2.1.1.4 Actual Number of Theoretical Plates

An early observation by of the plate-reflux relation was:

$$(R - R_m)(N - N_m) = \text{const},$$



But no general value for the constant was possible. Several correlations of calculated data between these same variables have since been made. A graphical correlation made by Gilliland has found wide acceptance because of its fair accuracy and simplicity of use. Of the several representations of the plot by equations that of Molokanov et al. is accurate and easy to use:

$$Y = \frac{N - N_{\min}}{N + 1} = 1 - \exp\left[\left(\frac{1 + 54.4X}{11 + 117.2X}\right)\left(\frac{X - 1}{X^{0.5}}\right)\right],$$

Where,

$$X = \frac{R - R_{\min}}{R + 1},$$

From which the number of theoretical trays is

$$N = \frac{N_m + Y}{1 - Y}$$

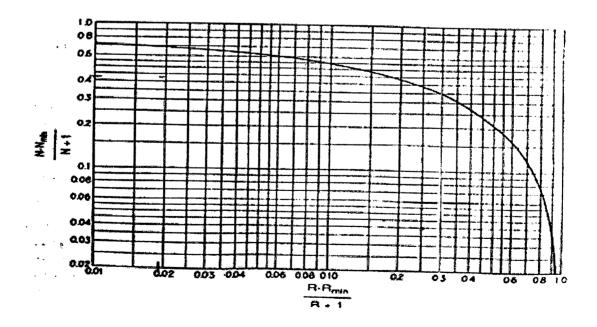


Figure 2: The Gilliland Correlation



The Gilliland correlation appears to be conservative for feeds with values of  $\mathbf{q}$  (the thermal condition of the feed), and can be in error when there is a large difference in tray requirements above and below the feed. The principal value of the correlation appears to be for preliminary exploration of design variables which can be refined by computer calculations. Although it is often used for final design, that should be done with caution.

#### 2.1.1.5 Feed Tray Location

An estimate can be made by using the Fenske equation to calculate the number of stages in the rectifying and stripping sections separately, but this requires an estimate of the feed-point temperature. An alternative approach is to use the empirical equation given by Kirkbride:

$$\log \left[ \frac{N_r}{N_s} \right] = 0.206 \log \left[ \left( \frac{B}{D} \right) \left( \frac{x_{f,HK}}{x_{f,LK}} \right) \left( \frac{x_{b,LK}}{x_{d,HK}} \right)^2 \right]$$

Where,

 $N_r$  = number of stages above the feed, including any partial condenser,

N<sub>s</sub> = number of stages below the feed, including the reboiler,

B = molar flow bottom product,

D = molar flow top product,

 $X_{f, HK}$  = concentration of the heavy key in the feed,

 $X_{f, LK}$  = concentration of the light key in the feed,

 $X_{d, HK}$  = concentration of the heavy key in the top product,

 $X_{b, LK}$  = concentration of the light key if in the bottom product.

# **2.2 DESIGNING PARAMETERS FOR DEPROPANIZER COLUMN**

The following data needs to be calculated to design a depropanizer column:

- Distillates flow rate
- Bottoms flow rate
- Composition of the various components obtained from distillates and bottoms.
- Minimum number of stages



- Minimum reflux
- Number of Theoretical stages
- Actual Number of stages
- Stripping and Rectifying feed tray location
- Optimum Reflux
- Column Internals
- Reboiler and Condenser Loads

However, the following specifications are needed to be given in order to design a distillation column:

# **Separation Specification:**

$$F = B + D \qquad -----(A)$$

$$Fx_F = Bx_B + Dx_D - (B)$$

At a given feed flow rate and feed composition, there are two equations [Eqs. (A) & (B)] and four unknowns: B, D,  $x_B$  and  $x_D$ . There fore, only two variables can be specified for the separation. Further, at least one of the two specified variables must be a composition.

#### **Composition specification**

If one product flow is specified, the concentration of one component either in the distillate or in the bottom (but not both) can be specified. If neither a recovery nor a product rate is specified, the concentration of one component in the distillate and one component in the bottom can be specified.

The above applies to both binary and multicomponent distillation. In multicomponent distillation, once the above are specified, other components will distribute according to the equilibrium relationship. Frequently, a product specification sets the maximum concentration of impurities that can be tolerated in the product. The one impurity which is dependent on the column separation and is most difficult to achieve sets the composition specification in the column.



### **Physical Property Specification**

A product composition can often be specified in terms of a physical property that is a direct function of composition. For instance, the vapor pressure of a bottom product is often a good measure of the concentration of lights in the bottoms, and may be specified instead. Other physical properties include the Reid vapor pressure (RVP), viscosity, refractive index, freezing point, molecular weight & others. A physical property specification is often preferred either when it is easy to monitor (e.g., refractive index), or when it provides a good functional specification of product purity.

#### **Heat Duty Specifications**

A composition or product rate specification may be substituted by a heat duty or internal flow (e.g., reflux) specification. This is done either to improve convergence in a computer simulation (especially if compositions are in the part per million levels), or in a revamp when the column or its exchangers are at a capacity limits. The mass, component, and energy balance equations translate this specification into a composition or product rate specification.

#### Side product

For each side product, one additional specification is required. This specification is either a product rate (e.g., the side product rate) or a product composition.

#### Heat addition or removal

For each point of heat addition or removal, an additional specification is required. This specification is usually a heat duty or an internal product flow.



# 2.3 IMPORTANT CONSIDERATIONS IN OPTIMIZATION OF DISTILLATION COLUMN

#### 2.3.1 Pressure Considerations

As the pressure of a column is raised:

- Separation becomes more difficult since the relative volatility decreases more plates and reflux are required to achieve the separation.
- The latent heat of vaporization decreases, reducing the duties of the reboiler and condenser.
- The vapor density increases, resulting in a smaller column diameter.
- The reboiler temperature increases. This is usually limited by the decomposition temperature of the material being vaporized.
- Condenser temperature increases.

As the pressure is lowered, these effects are reversed. A lower pressure limit is usually encountered by a desire to avoid vacuum operation and / or refrigeration in the condenser. For an initial design, it is adequate to set the distillation pressure above ambient and less as allowed by cooling water or air cooling in the condenser. An initial starting value might be selected so that the bubble point of the overhead product is 10°C above the summer cooling water temperature or to atmospheric pressure if vacuum operation is suggested.

#### 2.3.2 Reflux Ratio Considerations

We have several trade-offs in the selection of a reflux ratio. As the reflux ratio is increased:

- The purity of the product is increased.
- The capital costs decrease since the number of trays is decreased.
- The energy costs increase as more reboiling and condensing are required.

If the optimal reflux ratio is less than 1.1 times the minimum reflux, select 1.1 times the minimum reflux since a small error in design data or operating conditions might lead to a column that does not work.



#### 2.3.3 Feed Considerations

The feed consideration is more of an afterthought rather than a critical design parameter. The question is whether the feed is at the bubble point, sub cooled, partial vapor, or all vapor. In general, a sub cooled feed:

- Decreases the number of tray in the rectifying section but increases the trays in the stripping section.
- Increases the size of the reboiler but decreases the size of the condenser.

Partially vaporized feed reverses this.

#### 2.4 PROCESS DESIGN AND ADOPTED PROCEDURE

Process design proceeds in the following steps:

- 1. Specify separation. If product composition or product flow requirements are not defined, determine them by material and energy balance optimization.
- 2. Set column pressure.
- 3. Determine the minimum reflux and minimum number of stages.
- 4. Find the optimum feed stage.
- 5. Select three ratios of actual to minimum reflux. For each, calculate the number of stages and size the column and auxiliaries. Determine which is the most economical. This optimization procedure can be bypassed by selecting a single ratio of reflux to minimum reflux.
- 6. The calculations so far can be shortcut or rigorous.
- 7. Re examine steps 3 and 4, refining earlier estimates as necessary. If the refinements are large, steps 5 and 6 may need repeating.
- 8. Analyze the design graphically to ensure optimum design and absence of pinched regions.

# 2.4.1 Approximate Column Sizing

An approximate estimate of the overall column size can be made once the number of real stages required for the separation is known. This is often needed to make a rough estimate of the capital cost for project evaluation.



#### 2.4.1.1 Plate Spacing

The overall height of the column will depend on the plate spacing. Plate spacing's from 0.15 m (6 in.) to 1 m (36 in.) are normally used. The spacing chosen will depend on the column diameter and operating conditions. Close spacing is used with small-diameter columns, and where head room is restricted; as it will be when a column is installed in a building. For columns above 1 m diameter, plate spacing's of 0.3 to 0.6 m will normally be used, and 0.5 m (18 in.) can be taken as an initial estimate. This would be revised, as necessary, when the detailed plate design is made. A larger spacing will be needed between certain plates to accommodate feed and side streams arrangements, and for manways.

#### 2.4.1.2 Column Diameter

The principal factor that determines the column diameter is the vapor flow-rate. The vapor velocity must be below that which would cause excessive liquid entrainment or a high-pressure drop. The equation given below, which is based on the well-known Souders and Brown equation, Lowenstein (1961), can be used to estimate the maximum allowable superficial vapor velocity, and hence the column area and diameter:

$$\hat{\boldsymbol{u}}_v = (-0.171l_t^2 + 0.27l_t - 0.047) \left[ \frac{(\rho_L - \rho_v)}{\rho_v} \right]^{1/2}$$

Where.

 $u_v$  = maximum allowable vapor velocity, based on the gross (total) column crosssectional area, m/s,

 $l_t$  = plate spacing, m, (range 0.5-1.5).

The column diameter, D<sub>c</sub>, can then be calculated:

$$D_{c} = \sqrt{\frac{4\hat{V}_{w}}{\pi \rho_{v} \hat{u}_{v}}}$$

Where, Vw is the maximum vapor rate, kg/s.



This approximate estimate of the diameter would be revised when the detailed plate design is undertaken.

#### 2.4.2 Tray Design

Once the process design is completed, the equipment design begins. This phase of the design translates the process requirements (i.e., the vapor and liquid loads in each section of the column) into actual hardware. The hardware design proceeds in two phases: primary (basic) and secondary (detailed layout). The primary phase sets column diameter, type of tray, and split of tray area into bubbling and downcomer areas.

This phase also provides a preliminary (and usually close) estimate of tray spacing, number of passes, and other features of tray and downcomer layout such as weir height, fractional hole area, hole diameter, and clearance under the downcomer. These estimates are later firmed up in the secondary phase. Functionally, the primary phase sets the major equipment requirements, while the secondary phase engineers the finer details. The primary phase has a major impact on column costs, but a relatively small influence on achieving trouble-free operation. These roles are reversed in the secondary phase: it has a relatively small impact on column costs, but a major impact on achieving trouble-free operation.

#### 2.4.2.1 Common Types of Tray

- Bubble cap tray
- Sieve tray
- Dual flow tray
- Valve tray

#### **Bubble Cap Tray:**

The bubble cap tray was the workhorse of distillation before the 1960s. Presently, bubble-cap trays are specified only for special applications, while sieve and valve trays are the most popular types. The bubble-cap tray is a flat perforated plate with risers (chimney like pipes) around the holes, and caps in the form of inverted cups over the risers. The caps are usually (but not always) equipped with slots or holes through which the vapor comes out. Liquid and froth are



trapped on the tray to a depth at least equal to the weir height or riser height, giving the bubble-cap tray a unique ability to operate at low vapor and liquid rates.

They offer the distinct advantage of being able to handle very wide ranges of liquid and gas flow rates satisfactorily. They have now been abandoned for new installations because of their cost, which is roughly, double that for sieve, counter flow, and valve tray.

#### Sieve Tray:

The sieve tray is a flat perforated plate. Vapor issues from the holes to give a multi-orifice effect. The gas dispersed by the perforations, expands the liquid into a turbulent froth, characterized by a very large interfacial surface for mass transfer. The vapor velocity keeps the liquid from flowing down through the holes (weeping). At low velocities, liquid weeps through the holes, bypassing some of the tray and reducing efficiency, giving sieve trays relatively poor turndown. Sieve trays are simple and easy to fabricate, and are therefore relatively inexpensive.

#### **Dual Flow Tray:**

A dual-flow tray is a sieve tray with no downcomers. This tray operates with liquid continuously weeping through the holes, hence its low efficiency. Tray froth height diminishes rapidly when vapor velocity is reduced, causing further efficiency deterioration upon turndown.

Turndown of a dual-flow tray is even poorer than that of a sieve tray with downcomers. Large-diameter (>8 ft) dual flow trays are known to sometimes experience instability. Dual flow trays are prone to channeling, and are therefore sensitive to out of levelness and to liquid distribution. Due to the absence of downcomers, dual flow trays give more tray area, and therefore have a greater capacity than any of the common tray types. The absence of downcomers, and the larger open areas, renders dual flow trays the most suitable to handle highly fouling services, slurries, and corrosive services. Dual-flow trays are also the least expensive to make, and easiest to install and maintain.

#### Valve Tray:

These are the sieve trays with large (roughly 35 to 40 mm diameter) variable openings for gas flow. The perforations are covered with movable caps which rise as the flow rate of gas



increases. Valves can be round or rectangular, with or without caging. The upper limit of opening is controlled by a caging structure or by restrictive legs at the bottom of the valve unit. As vapor rate falls, the disk openings are reduced, or they may settle intermittently over the holes, this stops the liquid from weeping and gives the valve tray its main advantage-good operation at low flow rates, and therefore, a high turndown.

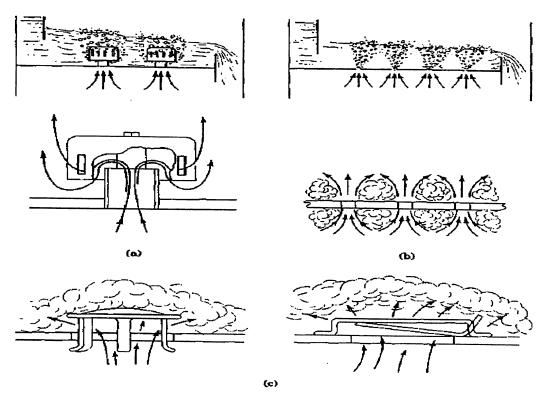


Figure 3. Flow Through Tray Vapor Passages (a) Bubble Cap (b) Sieve (c) Valve

# 2.4.2.2 Comparison of Common Types of Trays

The principal factors to consider when comparing the performance of bubble-cap, sieve and valve plates are: cost, capacity, operating range, efficiency and pressure drop.

#### Cost:

Bubble-cap plates are appreciably more expensive than sieve or valve plates. The relative cost will depend on the material of construction used; for mild steel the ratios, bubble-cap: valve: sieve, are approximately 3.0: 1.5: 1.0.



#### Capacity:

There is little difference in the capacity rating of the three types (the diameter of the column required for a given flow-rate); the ranking is sieve, valve and bubble-cap.

#### **Operating range:**

This is the most significant factor. By operating range is meant the range of vapor and liquid rates over which the plate will operate satisfactorily (the stable operating range). Some flexibility will always be required in an operating plant to allow for changes in production rate, and to cover start-up and shut-down conditions. The ratio of the highest to the lowest flow rates is often referred to as the "turn-down" ratio. Bubble-cap plates have a positive liquid seal and can therefore operate efficiently at very low vapor rates. Sieve plates rely on the flow of vapor through the holes to hold the liquid on the plate, and cannot operate at very low vapor rates. But, with good design, sieve plates can be designed to give a satisfactory operating range; typically, from 50 per cent to 120 per cent of design capacity. Valve plates are intended to give greater flexibility than sieve plates at a lower cost than bubble-caps.

#### Efficiency:

The Murphree efficiency of the three types of plate will be virtually the same when operating over their design flow range and no real distinction can be made between them

#### Pressure drop:

The pressure drop over the plates can be an important design consideration, particularly for vacuum columns. The plate pressure drop will depend on the detailed design of the plate but, in general, sieve plates give the lowest pressure drop, followed by valves, with bubble-caps giving the highest.

#### **Summary:**

Sieve plates are the cheapest and are satisfactory for most applications. Valve plates should be considered if the specified turn-down ratio cannot be met with sieve plates. Bubble-caps should only be used where very low vapor (gas) rates have to be handled and a positive liquid seal is essential at all flow-rates.



Table 1. Comparison of Common Type of Trays

Type	Sieve Tray	Valve Tray	Bubble Cap Tray	Dual Flow Trays
Maintenance	Low	Low to moderate	Relatively high	Low
Fouling	Low	Low to moderate	High, tends to	Extremely low.
tendency			collect solids	Suitable where
				fouling is
				extensive and for
				slurry handling.
Effect of	Low	Low to moderate	High	Very low
corrosion				
Availability of	Well known	Proprietary but	Well known	Some information
design		information		available
information		readily available		
Capacity	High	High to very high	Moderately high	Very high
Efficiency	High	High	Moderately high	Lower than other
				types
Turndown	About 2:1, not	About 4-5:1, some	Excellent, better	Low, even lower
	suitable for	special designs	than valve trays.	than sieve trays.
	operation	achieve 10:1 or	Good at extremely	Unsuitable for
	under variable	more	low liquid rates	variable load
	loads			operations
Entrainment	Moderate	Moderate	High, about 3	Low to moderate
			times higher tan	
			sieve trays	
Pressure drop	Moderate	Moderate, early	High	Low to moderate
		designs some what		
		higher, recent		
		designs same as		
		sieve trays		
Cost	Low	About 20% higher	High, about 2 to 3	Low
		than sieve trays	times the cost of	
			sieve trays	



Main	Most columns	Most columns,	Extremely low	Capacity revamps
applications	when	services where	flow conditions,	where
	turndown is	turndown is	where leakages	efficiency and
	not critical	important	must be minimized	turndown can
				be sacrificed,
				Highly fouling and
				corrosive services
Share of	25%	70%	5%	No information
market				

### 2.4.3 Plate Hydraulic Design

The basic requirements of a plate contacting stage are that it should:

- Provide good vapor-liquid contact.
- Provide sufficient liquid hold-up for good mass transfer (high efficiency).
- Have sufficient area and spacing to keep the entrainment and pressure drop within acceptable limits.
- Have sufficient downcomer area for the liquid to flow freely from plate to plate.

The plate design methods use semi-empirical correlations derived from fundamental research work combined with practical experience obtained from the operation of commercial columns. Proven layouts are used, and the plate dimensions are kept within the range of values known to give satisfactory performance.

### 2.4.3.1 Operating Range

Satisfactory operation will only be achieved over a limited range of vapor and liquid flow rates. A typical performance diagram for a sieve plate is shown in Figure given below:



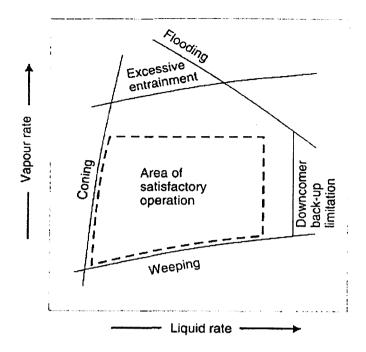


Figure 4. Sieve Tray Performance Diagram

The upper limit to vapor flow is set by the condition of flooding. At flooding there is a sharp drop in plate efficiency and increase in pressure drop. Flooding is caused by either the excessive carryover of liquid to the next plate by entrainment, or by liquid backing-up in the downcomers. The lower limit of the vapor flow is set by the condition of weeping. Weeping occurs when the vapor flow is insufficient to maintain a level of liquid on the plate. "Coning" occurs at low liquid rates, and is the term given to the condition where the vapor pushes the liquid back from the holes and jets upward, with poor liquid contact. In the following sections gas can be taken as synonymous with vapor when applying the method to the design of plates for absorption columns.

### 2.4.3.2 Plate-Design Procedure

A trial-and-error approach is necessary in plate design: starting with a rough plate layout, checking key performance factors and revising the design, as necessary, until a satisfactory design is achieved, a typical design procedure is set out below and discussed in the following sections. The normal range of each design variable is given in the discussion, together with recommended values which can be used to start the design.



- 1. Calculate the maximum and minimum vapor and liquid flow-rates, for the turn down ratio required.
- 2. Collect, or estimate, the system physical properties.
- 3. Select trial plate spacing.
- 4. Estimate the column diameter, based on flooding considerations.
- 5. Decide the liquid flow arrangement.
- 6. Make a trial plate layout: downcomer area, active area, hole area, hole size, weir height.
- 7. Check the weeping rate, if unsatisfactory return to step 6.
- 8. Check the plate pressure drop, if too high return to step 6.
- 9. Check downcomer back-up, if too high return to step 6 or 3.
- 10. Decide plate layout details: calming zones, unperforated areas. Check hole pitch, if unsatisfactory return to step 6.
- 11. Recalculate the percentage flooding based on chosen column diameter.
- 12. Check entrainment, if too high return to step 4.
- 13. Optimize design: repeat steps 3 to 12 to find smallest diameter and plate spacing acceptable (lowest cost).
- 14. Finalize design: draw up the plate specification and sketch the layout.

### 2.4.3.3 Plate Areas

The following areas terms are used in the plate design procedure:

 $A_c$  = total column cross-sectional area,

 $A_d$  = cross-sectional area of downcomer,

 $A_n$  = net area available for vapor-liquid disengagement, normally equal to  $A_c$  - $A_d$  for a single pass plate,

 $A_a$  = active, or bubbling, area, equal to  $A_c$  —  $2A_d$  for single-pass plates,

 $A_h$  = hole area, the total area of all the active holes,

 $A_p$  = perforated area (including blanked areas),

 $A_{ap}$  = the clearance area under the downcomer apron.



### **2.4.3.4 Diameter**

The flooding condition fixes the upper limit of vapor velocity. A high vapor velocity is needed for high plate efficiencies, and the velocity will normally be between 70 to 90 per cent of that which would cause flooding. For design, a value of 80 to 85 per cent of the flooding velocity should be used.

The flooding velocity can be estimated from the correlation given by Fair (1961):

$$u_f = K_1 \sqrt{\frac{\rho_L - \rho_v}{\rho_v}}$$

Where,

 $u_f$  = flooding vapor velocity, m/s, based on the net column cross-sectional area An  $K_1$ = a constant obtained from Figure given below:

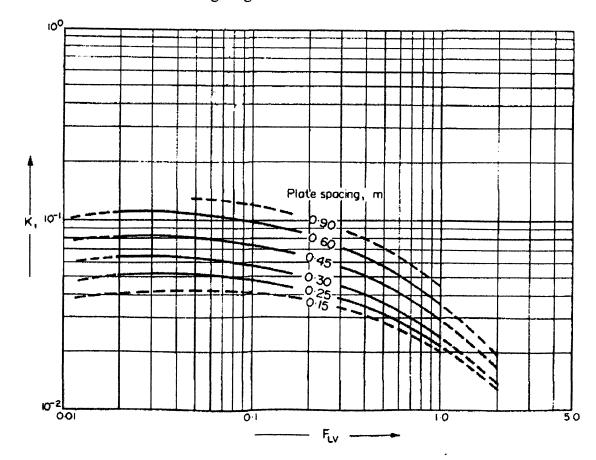


Figure 5: Flooding Velocity for Sieve Trays



The liquid-vapor flow factor F<sub>LV</sub> in Figure 5 is given by:

$$F_{LV} = \frac{L_w}{V_w} \sqrt{\frac{\rho_v}{\rho_L}}$$

Where,

 $L_w$  = liquid mass flow-rate, kg/s,

 $V_w$  = vapor mass flow-rate, kg/s.

The following restrictions apply to the use of Figure 5:

- 1. Hole size less than 6.5 mm. Entrainment may be greater with larger hole sizes.
- 2. Weir height less than 15 per cent of the plate spacing.
- 3. Non-foaming systems.
- 4. Hole: active area ratio greater than 0.10; for other ratios apply the following corrections:

hole: active area	multiply K <sub>1</sub> by
0.10	1.0
0.08	0.9
0.06	0.8

5. Liquid surface tension 0.02 N/m, for other surface tensions  $\rho$  multiply the value of K1 by  $[\rho/0.02]^{0.2}$ .

To calculate the column diameter an estimate of the net area  $A_n$  is required. As a first trial take the downcomer area as 12 per cent of the total, and assume that the hole-active area is 10 per cent. Where the vapor and liquid flow-rates, or physical properties, vary significantly throughout the column a plate design should be made for several points up the column. For distillation it will usually be sufficient to design for the conditions above and below the feed points, Changes in the vapor flow-rate will normally be accommodated by adjusting the hole area; often by blanking off some rows of holes. Different column diameters would only be used where there is a considerable change in flow-rate. Changes in liquid rate can be allowed for by adjusting the liquid downcomer areas.



### 2.4.3.5 Liquid-Flow Arrangement

The choice of plate type (reverse, single pass or multiple pass) will depend on the liquid flow-rate and column diameter. An initial selection can be made using Figure-6, which has been adapted from a similar figure given by Huang and Hodson (1958). The selection of plate can be done by using the figure five.

### 2.4.3.6 Entrainment

Entrainment can be estimated from the correlation given by Fair (1961), Figure 7, which gives the fractional entrainment  $\psi$  (kg/kg gross liquid flow) as a function of the liquid-vapor factor  $F_{LV}$  with the percentage approach to flooding as a parameter. The percentage flooding is given by:

percentage flooding = 
$$\frac{u_n \text{ actual velocity (based on net area)}}{u_f \text{ (from equation 11.81)}}$$

As a rough guide the upper limit of  $\psi$  can be taken as 0.1; below this figure the effect on efficiency will be small. The optimum design value may be above this figure.

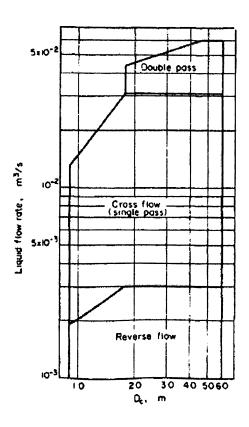


Figure 6. Selection of Liquid Flow Arrangement



### **2.4.3.7 Weep Point**

The lower limit of the operating range occurs when liquid leakage through the plate holes becomes excessive. This is known as the weep point. The vapor velocity at the weep point is the minimum value for stable operation. The hole area must be chosen so that at the lowest operating rate the vapor flow velocity is still well above the weep point. Several correlations have been proposed for predicting the vapor velocity at the weep point; see Chase (1967). That given by Eduljee (1959) is one of the simplest to use, and has been shown to be reliable. The minimum design vapor velocity is given by:

$$\check{u}_h = \frac{[K_2 - 0.90(25.4 - d_h)]}{(\rho_v)^{1/2}}$$

Where,

 $u_h$  = minimum vapor velocity through the holes(based on the hole area), m/s,

 $d_h$  = hole diameter, mm,

 $K_2$  = a constant, dependent on the depth of clear liquid on the plate, obtained from Figure 8.

The clear liquid depth is equal to the height of the weir  $h_w$  plus the depth of the crest of liquid over the weir  $h_{ow}$ .



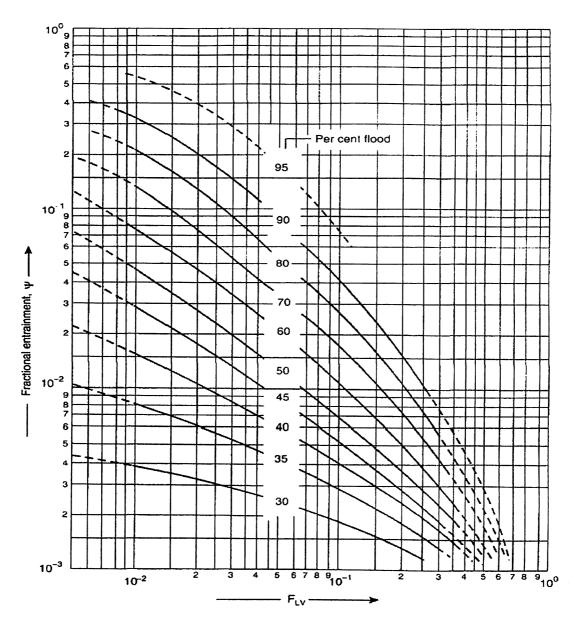


Figure 7. Entrainment Correlation for Sieve Tray

### 2.4.3.8 Weir Liquid Crest

The height of the liquid crest over the weir can be estimated using the Francis weir formula. For a segmental downcomer this can be written as:

$$h_{ow} = 750 \left[ \frac{L_w}{\rho_L l_w} \right]^{2/3}$$



Where,

lw =weir length, m,

 $h_{ow}$  = weir crest, mm liquid,

 $L_w$  = liquid flow-rate, kg/s.

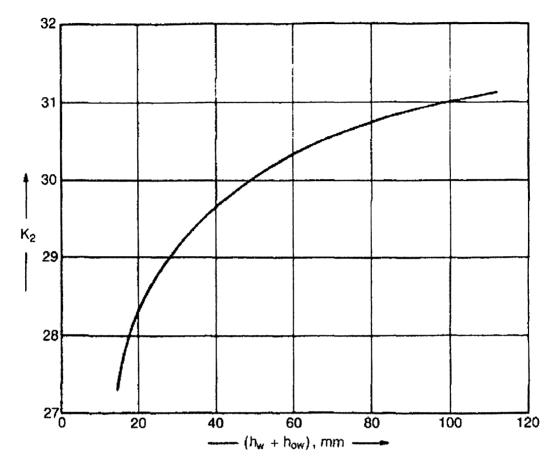


Figure 8. Weep Point Correlation

With segmental downcomers the column wall constricts the liquid flow, and the weir crest will be higher than that predicted by the Francis formula for flow over an open weir. To ensure an even flow of liquid along the weir, the crest should be at least 10 mm at the lowest liquid rate. Serrated weirs are sometimes used for very low liquid rates.

### 2.4.3.9 Weir Dimensions

### Weir Height

The height of the weir determines the volume of liquid on the plate and is an important factor in determining the plate efficiency. A high weir will increase the plate efficiency but at the expense



of a higher plate pressure drop. For columns operating above atmospheric pressure the weir heights will normally be between 40 mm to 90 mm (1.5 to 3.5 in.); 40 to 50 mm is recommended. For vacuum operation lower weir heights are used to reduce the pressure drop; 6 to 12 mm (0.25 to 0.5 in.) is recommended.

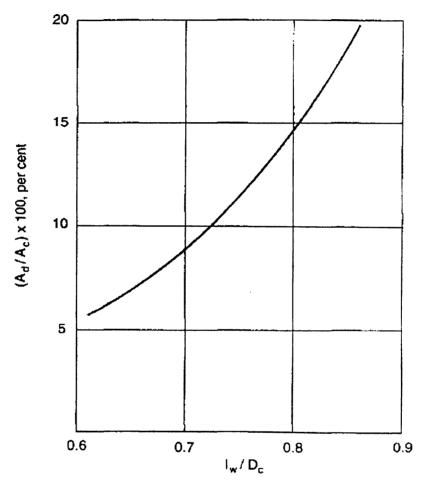


Figure 9. Relation between Downcomer Area & Weir Height

#### **Inlet Weirs**

Inlet weirs, or recessed pans, are sometimes used to improve the distribution of liquid across the plate; but are seldom needed with segmental downcomers.

#### Weir Length

With segmental downcomers the length of the weir fixes the area of the downcomer. The chord length will normally be between 0.6 to 0.85 of the column diameter. A good initial value to use is 0.77, equivalent to a downcomer area of 12 per cent. The relationship between weir length and



downcomer area is given in Figure 8. For double-pass plates the width of the central downcomer is normally 200-250 mm (8-10 in.).

### 2.4.3.10 Perforated Area

The area available for perforation will be reduced by the obstruction caused by structural members (the support rings and beams), and by the use of calming zones. Calming zones are unperforated strips of plate at the inlet and outlet sides of the plate. The width of each zone is usually made the same; recommended values are: below 1.5 m diameter, 75 mm; above, 100 mm. The width of the support ring for sectional plates will normally be 50 to 75 mm; the support ring should not extend into the downcomer area. A strip of unperforated plate will be left round the edge of cartridge-type trays to stiffen the plate.

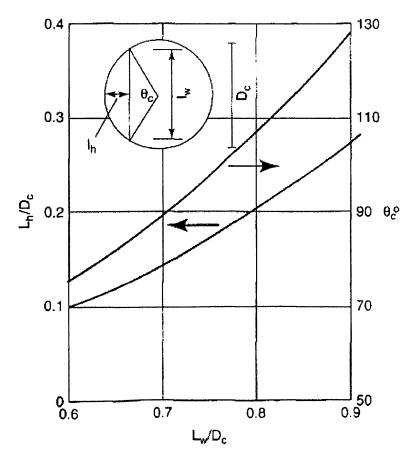


Figure 10. Relation between Angle Subtended by Chord, Chord Height and Chord Length



The unperforated area can be calculated from the plate geometry. The relationship between the weir chord length, chord height and the angle subtended by the chord is given in Figure 10.

### 2.4.3.11 Hole Size

The hole sizes used vary from 2.5 to 12 mm; 5 mm is the preferred size. Larger holes are occasionally used for fouling systems. The holes are drilled or punched. Punching is cheaper, but the minimum size of hole that can be punched will depend on the plate thickness, For carbon steel, hole sizes approximately equal to the plate thickness can be punched, but for stainless steel the minimum hole size that can be punched is about twice the plate thickness. Typical plate thicknesses used are: 5 mm (3/16 in.) for carbon steel, and 3 mm (12 gauges) for stainless steel. When punched plates are used they should be installed with the direction of punching upward. Punching forms a slight nozzle, and reversing the plate will increase the pressure drop.

### 2.4.3.12 Hole Pitch

The hole pitch (distance between the hole centers)  $l_p$  should not be less than 2.0 hole diameters, and the normal range will be 2.5 to 4.0 diameters. Within this range the pitch can be selected to give the number of active holes required for the total hole area specified. Square and equilateral triangular patterns are used; triangular is preferred. The total hole area as a fraction of the perforated area  $A_p$  is given by the following expression, for an equilateral triangular pitch:

$$\frac{A_h}{A_p} = 0.9 \left[ \frac{d_h}{l_p} \right]^2$$

This equation is plotted in Figure 11.



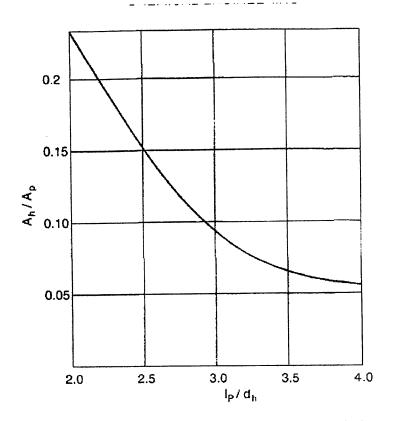


Figure 11. Relation between Hole Area & Pitch

### 2.4..3.13 Hydraulic Gradient

The hydraulic gradient is the difference in liquid level needed to drive the liquid flow across the plate. On sieve plates, unlike bubble-cap plates, the resistance to liquid flow will be small, and the hydraulic gradient is usually ignored in sieve-plate design. It can be significant in vacuum operation, as with the low weir heights used the hydraulic gradient can be a significant fraction of the total liquid depth. Methods for estimating the hydraulic gradient are given by Fair (1963).

### 2.4.3.14 Liquid Throw

The liquid throw is the horizontal distance travelled by the liquid stream flowing over the downcomer weir. It is only an important consideration in the design of multiple-pass plates. Bolles (1963) gives a method for estimating the liquid throw.



### 2.4.3.15 Rate Pressure Drop

The pressure drop over the plates is an important design consideration. There are two main sources of pressure loss: that due to vapor flow through the holes (an orifice loss), and that due to the static head of liquid on the plate.

A simple additive model is normally used to predict the total pressure drop. The total is taken as the sum of the pressure drop calculated for the flow of vapor through the dry plate (the dry plate drop  $h_d$ ); the head of clear liquid on the plate  $(h_w + h_{gw})$  and a term to account for other, minor, sources of pressure loss, the so-called residual loss  $h_r$ . The residual loss is the difference between the observed experimental pressure drop and the simple sum of the dry-plate drop and the clear-liquid height. It accounts for the two effects: the energy to form the vapor bubbles and the fact that on an operating plate the liquid head will not be clear liquid but a head of "aerated" liquid froth, and the froth density and height will be different from that of the clear liquid. It is convenient to express the pressure drops in terms of millimeters of liquid. In pressure units:

$$\Delta P_t = 9.81 \times 10^{-3} h_t \rho_L$$

Where,

 $\Delta P_t$  = total plate pressure drop, Pa (N/m<sup>2</sup>),

 $h_t$  = total plate pressure drop, mm liquid.

### 2.4.3.16 Dry Plate Drop

The pressure drop through the dry plate can be estimated using expressions derived for flow through orifices.

$$h_d = 51 \left[ \frac{u_h}{C_0} \right]^2 \frac{\rho_v}{\rho_L}$$

Where the orifice coefficient  $C_0$  is a function of the plate thickness, hole diameter, and the hole to perforated area ratio.  $C_0$  can be obtained from Figure 12; which has been adapted from a similar figure by Liebson et al. (1957).  $U_f$ , is the velocity through the holes, m/s.



### 2.4.3.17 Residual Head

Methods have been proposed for estimating the residual head as a function of liquid surface tension, froth density and froth height. However, as this correction term is small the use of an elaborate method for its estimation is not justified, and the simple equation proposed by Hunt et al. (1955) can be used:

$$h_r = \frac{12.5 \times 10^3}{\rho_L}$$

Above equation is equivalent to taking the residual drop as a fixed value of 12.5 mm of water (0.5 in.).

### **2.4.3.18 Total Drop**

The total plate drop is given by:

$$h_t = h_d + (h_w + h_{ow}) + h_r$$

If the hydraulic gradient is significant, half its value is added to the clear liquid height.

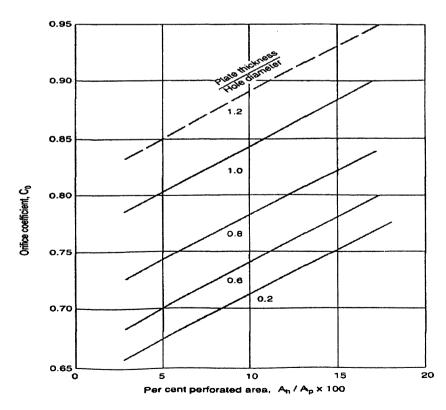


Figure 12. Discharge Coefficient, Sieve Plate



### 2.4.3.19 Downcomer Design [Back-Up]

The downcomer area and plate spacing must be such that the level of the liquid and froth in the downcomer is well below the top of the outlet weir on the plate above. If the level rises above the outlet weir the column will flood. The back-up of liquid in the downcomer is caused by the pressure drop over the plate (the downcomer in effect forms one leg of a U-tube) and the resistance to flow in the downcomer itself; see Figure 13. In terms of clear liquid the downcomer back-up is given by:

$$h_b = (h_w + h_{ow}) + h_t + h_{dc}$$

Where,

 $h_b$  = downcomer back-up, measured from plate surface, mm,

 $h_{dc}$  = head loss in the downcomer, mm.

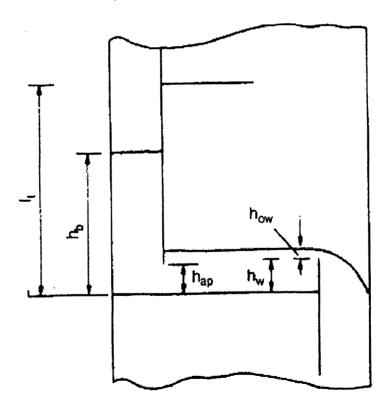


Figure 13. Downcomer Backup

The main resistance to flow will be caused by the constriction at the downcomer outlet, and the head loss in the downcomer can be estimated using the equation given by Cicalese et al. (1947):



$$h_{dc} = 166 \left[ \frac{L_{wd}}{\rho_L A_m} \right]^2$$

Where,

 $L_{wd}$  = liquid flow rate in downcomer, kg/s,

 $A_m$  = either the downcomer area Ad or the clearance area under the downcomer  $A_{sp}$  whichever is the smaller,  $m^2$ .

The clearance area under the downcomer is given by:

$$A_{ap} = h_{ap}l_{w}$$

Where,  $h_{ap}$  is height of the bottom edge of the apron above the plate. This height is normally set at 5 to 10 mm (0.25 to 0.5 in.) below the outlet weir height:

$$h_{ap} = h_w - (5 \text{ to } 10 \text{ mm})$$

### 2.4.3.20 Froth Height

To predict the height of "aerated" liquid on the plate, and the height of froth in the downcomer, some means of estimating the froth density is required. The density of the "aerated" liquid will normally be between 0.4 to 0.7 times that of the clear liquid. A number of correlations have been proposed for estimating froth density as a function of the vapor flow-rate and the liquid physical properties; however, none is particularly reliable, and for design purposes it is usually satisfactory to assume an average value of 0.5 of the liquid density. This value is also taken as the mean density of the fluid in the downcomer; which means that for safe design the clear liquid back-up should not exceed half the plate spacing  $l_t$  to avoid flooding.

Allowing for the weir height:

$$h_b \not> \frac{1}{2}(l_1 + h_w)$$



This criterion is, if anything, oversafe, and where close plate spacing is desired a better estimate of the froth density in the downcomer should be made. The method proposed by Thomas and Shah (1964) is recommended.

### 2.4.3.21 Downcomer Residence Time

Sufficient residence time must be allowed in the downcomer for the entrained vapour to disengage from the liquid stream; to prevent heavily "aerated" liquid being carried under the downcomer. A time of at least 3 seconds is recommended.

The downcomer residence time is given by:

$$t_r = \frac{A_d h_{bc} \rho_L}{L_{wd}}$$

Where,

 $t_r$  = residence time, s,

 $h_{bc}$  = clear liquid back-up, m

### 2.5 USE OF ASPEN HYSYS FOR SIMULATING DEPROPANIZER COLUMN

### **2.5.1 Use of Process Simulators**

Process simulation allows us to predict the behavior of a process by using basic engineering relationships, such as mass and energy balances, and phase and chemical equilibrium. Given reliable thermodynamic data, realistic operating conditions, and rigorous equipment models, we can simulate actual plant behavior. Process simulation enables you to run many cases, conduct "what if" analyses, and perform sensitivity studies and optimization runs. With simulation, we can design better plants and increase profitability in existing plants. Process simulation is useful throughout the entire lifecycle of a process, from research and development through process design to production.



### 2.5.2 Aspen Hysys as a Process Simulator

A process consists of chemical components being mixed, separated, heated, cooled, and converted by unit operations. These components are transferred from unit to unit through process streams.

We can translate a process into an Aspen Tech process simulation model by performing the following steps:

- 1. Define the process flow sheet:
  - Define the unit operations in the process.
- Define the process streams that flow to and from the unit operations.
- Select models from the Aspen Tech Model Library to describe each unit operation and place them on the process flow sheet.
- Place labeled streams on the process flow sheet and connect them to the unit operation models.
- 2. Specify the chemical components in the process. We can take these components from the Aspen Tech databanks, or we can define them.
- 3. Specify thermodynamic models to represent the physical properties of the components and mixtures in the process. These models are built into Aspen Tech.
- 4. Specify the component flow rates and the thermodynamic conditions (for example, temperature and pressure) of feed streams.
- 5. Specify the operating conditions for the unit operation models.

With Aspen Tech we can interactively change specifications such as, flow sheet configuration; operating conditions; and feed compositions, to run new cases and analyze process alternatives. In addition to process simulation, Aspen Tech allows us to perform a wide range of other tasks such as estimating and regressing physical properties, generating custom graphical and tabular output results, fitting plant data to simulation models, optimizing our process, and interfacing results to spreadsheets.

Aspen Hysys version 3.1.3 is used here for simulation purpose. A complete understanding of software is done under the able guidance of mentor before using it. We have used Peng



Robinson fluid package to estimate the various fluid properties. This fluid package is termed as "basis-1" in our hysys file.

### 2.5.3 Peng-Robinson Equation of State

The Peng Robinson (1976) equation of state (EOS) is a modification of the RK equation to better represent VLE calculations. The densities for the liquid phase in the SRK did not accurately represent the experimental values due to a high universal critical compressibility factor of 0.3333. The PR is a modification of the RK equation of state which corresponds to a lower critical compressibility of about 0.307 thus representing the VLE of natural gas systems accurately. The PR equation is represented by:

Property Class Name Applicable Phase

$$P = \frac{RT}{V - b} - \frac{a}{V(V + b) + b(V - b)}$$
 (4.8)

where:

$$a = a_c \alpha$$

$$a_c = 0.45724 \frac{R^2 T_c^2}{P_c}$$

$$b = 0.077480 \frac{RT_c}{P_c}$$
(4.9)

The functional dependency of the "a" term is shown in the following relation.

$$\sqrt{\alpha} = 1 + \kappa (1 - T_r^{0.5})$$

$$\kappa = 0.37464 + 1.5422\omega - 0.26992\omega^2$$
(4.10)



Where: The functional dependency of the "a" term is shown in the following relation.

The accuracy of the PR and SRK equations of state are approximately the same. However, the PR EOS represents the density of the liquid phase more accurately due to the lower critical compressibility factor. These equations were originally developed for pure components. To apply the PR EOS to mixtures, mixing rules are required for the "a" and "b" terms in Equation (4.2).

### **Property Methods**

A quick reference of calculation methods is shown in the table below for the PR EOS.

Calculation Method	Applicable Phase	Property Class Name
Z Factor	Vapour and Liquid	COTHPRZFactor Class
Molar Volume	Vapour and Liquid	COTHPRVolume Class
Enthalpy	Vapour and Liquid	COTHPREnthalpy Class
Entropy	Vapour and Liquid	COTHPREntropy Class
Isobaric heat capacity	Vapour and Liquid	COTHPRCp Class
Fugacity coefficient calculation	Vapour and Liquid	COTHPRLnFugacityCoeff Class
Fugacity calculation	Vapour and Liquid	COTHPRLnFugacity Class
Isochoric heat capacity	Vapour and Liquid	COTHPRCv Class
Mixing Rule 1	Vapour and Liquid	COTHPRab_1 Class
Mixing Rule 2	Vapour and Liquid	COTHPRab_2 Class
Mixing Rule 3	Vapour and Liquid	COTHPRab_3 Class
Mixing Rule 4	Vapour and Liquid	COTHPRab_4 Class
Mixing Rule 5	Vapour and Liquid	COTHPRab_5 Class
Mixing Rule 6	Vapour and Liquid	COTHPRab_6 Class



## PR Z Factor

The compressibility factor, Z, is calculated as the root for the following equation:

$$Z^{3} - (1 - B)Z^{2} + Z(A - 3B^{2} - 2B) - (AB - B^{2} - B^{3}) = 0$$
 (4.11)

$$A = \frac{aP}{R^2 T^2} \tag{4.12}$$

$$B = \frac{bP}{RT} \tag{4.13}$$

There are three roots for the above equation. It is considered that the smallest root is for the liquid phase and the largest root is for the vapour phase. The third root has no physical meaning.

## PR Molar Volume

The following relation calculates the molar volume for a specific phase.

$$V = \frac{ZRT}{P} \tag{4.14}$$

## Property Class Name and Applicable Phases

Property Class Name	Applicable Phase	
COTHPRVolume Class	Vapour and Liquid	

### **Notes**

The compressibility factor, Z, is calculated using PR Z Factor. For consistency, the PR molar volume always calls the PR Z Factor for the calculation of Z.



# PR Enthalpy

The following relation calculates the enthalpy.

$$H - H^{IG} = PV - RT - \left(a - \left(\frac{da}{dT}\right)T\right) \frac{1}{2\sqrt{2}b} \ln \frac{V + b(1 + \sqrt{2})}{V - b(1 - \sqrt{2})}$$
(4.15)

where:  $H^{IG}$  is the ideal gas enthalpy calculated at temperature, T

### Property Class Name and Applicable Phases

Property Class Name	Applicable Phase	
COTHPREnthalpy Class	Vapour and Liquid	

### Notes

The volume, V, is calculated using PR Molar Volume. For consistency, the PR Enthalpy always calls the PR Volume for the calculation of V.

# PR Entropy

The following relation calculates the entropy.

$$S - S^{IG} = R \ln \left( \frac{V - b}{RT} \right) - \frac{1}{2b\sqrt{2}} \ln \left( \frac{V + b(1 + \sqrt{2})}{V + b(1 - \sqrt{2})} \right) \frac{da}{dT}$$
(4.16)

where:  $S^{IG}$  is the ideal gas entropy calculated at temperature, T

## Property Class Name and Applicable Phases

Property Class Name	Applicable Phase	
COTHPREntropy Class	Vapour and Liquid	



### Notes

The volume, V, is calculated using PR Molar Volume. For consistency, the PR Entropy always calls the PR Volume for the calculation of V.

# PR Cp (Heat Capacity)

The following relation calculates the isobaric heat capacity.

$$C_{p} - C_{p}^{IG} = -T \int_{\infty}^{V} \left( \frac{\partial^{2} P}{\partial T^{2}} \right)_{V} dV + R + \frac{T \left( \frac{\partial V}{\partial T} \right)_{P}^{2}}{\left( \frac{\partial V}{\partial P} \right)_{T}}$$

$$(4.17)$$

where:  $Cp^{IG}$  is the ideal gas heat capacity calculated at temperature, T

## Property Class Name and Applicable Phases

Property Class Name	Applicable Phase	
COTHPRCp Class	Vapour and Liquid	

## Notes

The volume, V, is calculated using PR Molar Volume. For consistency, the PR Entropy always calls the PR Volume for the calculation of V.

# PR Fugacity Coefficient

The following relation calculates the fugacity coefficient.

$$\ln \phi_i = -\ln(V - b) + \frac{\overline{b}}{V - b} + \frac{a}{2\sqrt{2}b} \ln\left(\frac{V + b(1 + \sqrt{2})}{V + b(1 - \sqrt{2})}\right) \left(-1 + \frac{\overline{a}}{a} + \frac{\overline{b}}{b}\right) \tag{4.18}$$



$$\overline{a} = \frac{\partial n^2 a}{\partial n} \tag{4.19}$$

$$\overline{b} = \frac{\partial nb}{\partial n} \tag{4.20}$$

## Property Class Name and Applicable Phases

Property Class Name	Applicable Phase
COTHPRLnFugacityCoeff Class	Vapour and Liquid

## **Notes**

The volume, V, is calculated using PR Molar Volume. For consistency, the PR Fugacity Coefficient always calls the PR Volume for the calculation of V. The parameters a and b are calculated from the Mixing Rules.

# PR Fugacity

The following relation calculates the fugacity for a specific phase.

$$f_i = \phi_i y_i P \tag{4.21}$$

## Property Class Name and Applicable Phases

Property Class Name	Applicable Phase	
COTHPRLnFugacity Class	Vapour and Liquid	



# PR Cv (isochoric)

The following relation calculates the isochoric heat capacity.

$$C_{v} = C_{p} + \frac{T\left(\frac{\partial P}{\partial T}\right)^{2}_{V}}{\left(\frac{\partial P}{\partial V}\right)_{T}}$$

$$(4.22)$$

## Property Class Name and Applicable Phases

Property Class Name	Applicable Phase	
COTHPRCv Class	Vapour and Liquid	

### 2.5.4 Steps for Simulating Depropanizer Column on Aspen Hysys

1. In file menu, go to new case (ctrl+ N) to enter into the simulation basis manager. Under the component list, highlight the "master component" list and add the component used in unit operation. Click on the "fluid packages" tab and select the appropriate property package to all required properties. Here, we have used Peng Robinson property package. Now, enter the simulation environment. The window shown below or figure 14 will be seen:



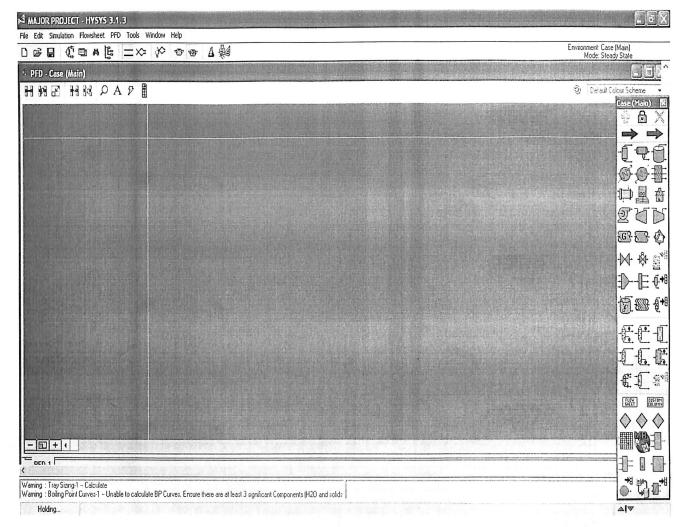


Figure 14

2. In the object palette, click on "shortcut distillation column". In PFD window, a short cut column in the name of T-100-2 will come. On double clicking T – 100-2, a pop up will appear as shown in figure 15. Under "design" tab, in "connections" bar, give the stream name, and fluid package of inlet, condenser duty, distillate, reboiler duty and bottoms. Now in "parameters" bar, put the values of the mole fraction of light key in bottoms and heavy key in distillate. Put the value of the condenser and reboiler pressure. Set external reflux ratio to any value. This will automatically calculate the minimum reflux ratio and hence, the value of external reflux should be made 1.2 – 1.5 times of minimum reflux.



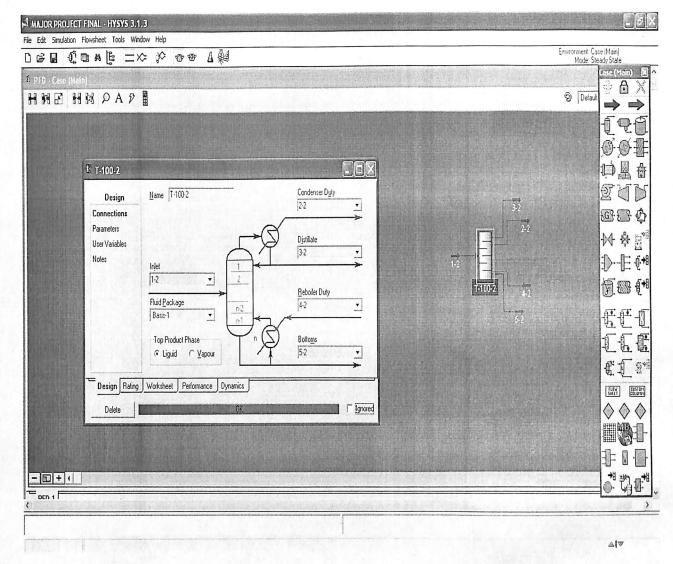


Figure 15

- 3. Double click on stream 1-2 and the temperature, pressure and molar flow (conditions) of the stream. Enter the compositions of each component in mole fraction. Click on the green button of "solver active" and this will aromatically calculate the following design parameters:
  - Conditions and compositions of distillate and bottoms.
  - Minimum number of trays, actual number of trays and feed tray location
  - Temperature & duty of condenser and reboiler

The shortcut column calculates the parameters at 100% efficiency. The efficiency of distillation column is ideally 70%, so we have to make the corrections accordingly.



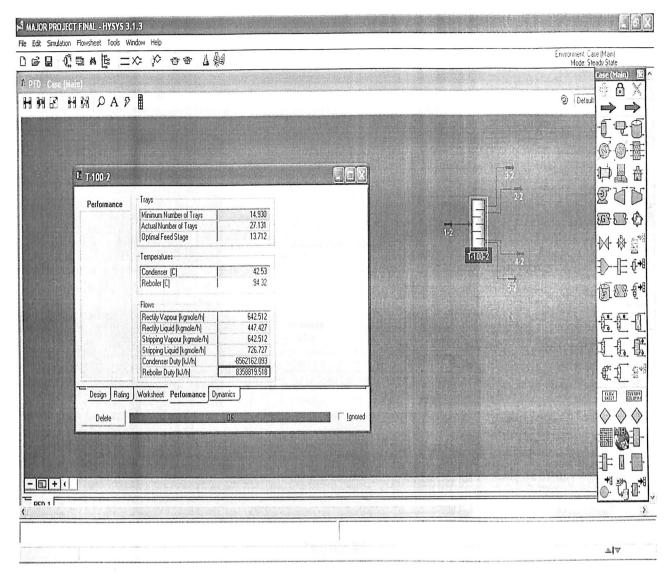


Figure 16

4. In the object palette, click on the distillation column. In PFD window, a distillation column in the name of T-100 will come. On double clicking T – 100, a pop up will appear as shown in figure 17. Under "design" tab, in "connections" bar, give the stream names of inlet stream, condenser energy stream, distillate liquid outlet, reboiler energy stream and bottom liquid outlet. Set the value of Delta P of condenser and reboiler and put the value of pressure of condenser & reboiler with the pressure difference of 80 KPa. Now, enter the value of number of stages.



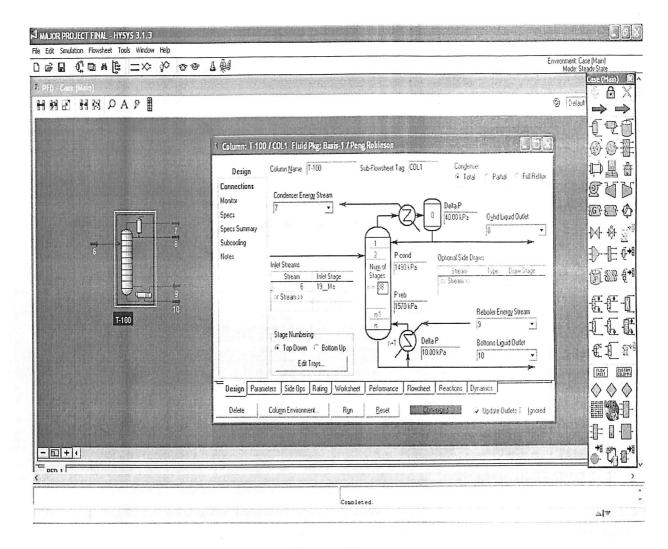


Figure 17

5. In the monitor tab, cick on "add spec" and add column component fraction with stage as "condenser", flow basis as "mole fraction", phase as "liquid" and key component obtained in the distillate with its mole fraction. Similarly, add another column component fraction for reboiler. Check the degree of freedom which should be zero as in figure 18.



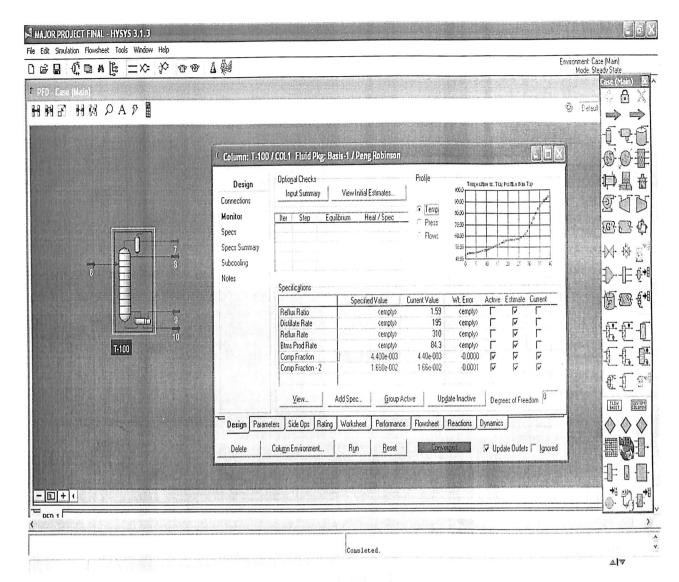


Figure 18

6. Again, click on the green button of "solver active". In this way, column T-100 can be converged. Thus, we obtain the pressure, temperature, net liquid flow & net vapour flow on the individual trays. Now, in tools menu, click on "utilities" (ctrl+U), add a utility of tray sizing and view tray sizing -1 utility. Click on "select TS" in setup bar under "design" tab & highlight "T-100". To obtain the values of tray internals, click on "results" bar under the "performance" tab.



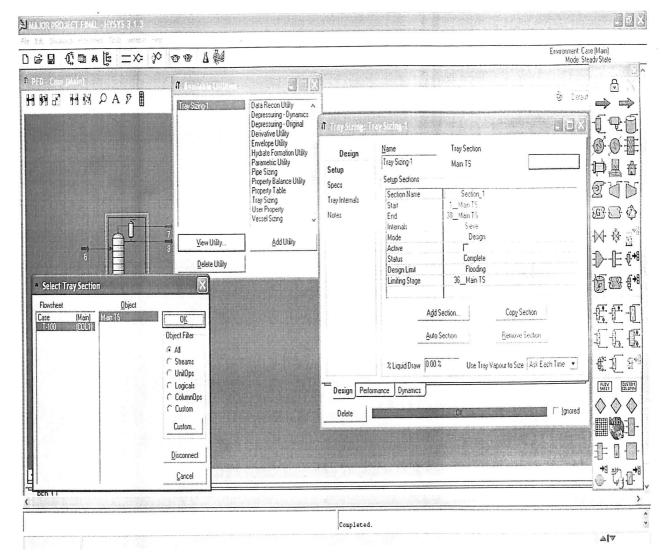


Figure 19

In this way, we can simulate a depropanizer column.



## **CHAPTER 3**

## **DESIGNING OF DEPROPANIZER TRAY COLUMN**



## 3. DESIGNING OF DEPROPANIZER TRAY COLUMN

### 3.1 CALCULATION OF BASIC PARAMETERS

The client has given us the following data.

Feed Flow rate

279.3 Kgmole / hr

Temperature

65.60 °C

**Pressure** 

20.40 bar

Vapor Feed

1

**Flooding** 

85%

Turndown

50%

**Efficiency** 

70%

### **COMPOSION OF FEED**

		Mole %	Mole Fraction
Ethane	Light non-key (C2)	00.85	0.0085
Propane	Light key (C3)	69.17	0.6917
i- Butane	Heavy Key (C4)	12.51	0.1251
n- Butane	Heavy non-Key (C4)	17.39	0.1739
i- Pentane	Heavy non-Key (C5)	00.06	0.0006
n- Pentane	Heavy non-Key (C5)	00.02	0.0002

### PURITY THAT SHOULD BE OBTAINED FROM THE DISTILLATES

Propane

Light key (C3)

98.3

0.9830



### PURITY THAT SHOULD BE OBTAINED FROM THE BOTTOMS

i- Butane Heavy Key (C4) 40.49 0.4049

n- Butane Heavy non-Key (C4) 57.58 0.5758

Writing,

Flow rate of Feed with F

Mole fraction of Feed with X<sub>f</sub>

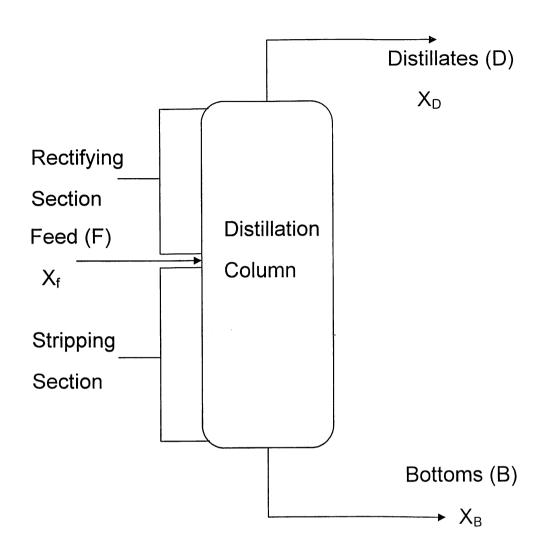
Flow Rate of Distillates with D

Mole fraction of Distillates with X<sub>D</sub>

Flow Rate of Bottoms with B

Mole fraction of Bottoms with X<sub>B</sub>





### APPLYING MASS/ MATERIAL BALANCE

To obtain the flow rate of Distillates and Bottoms and the mole fraction of non key components in Distillates and Bottoms

$$D + B = F$$

$$D + B = 279.3$$

#### APPLYING BUTANE BALANCE

$$F * X_f = D * X_D + B * X_B$$

$$279.3 * (0.1251 + 0.1739) = D * (0.4049 + 0.5758) + B * (1 - 0.983 - XD.e)$$

$$279.3 * (0.1251 + 0.1739) = D * (0.4049 + 0.5758) + (279.3 - D) * (1 - 0.983 - X_{D, e}) (1)$$



#### APPLYING ETHANE BALANCE

$$F * X_f = D * X_D + B * X_B$$

$$279.3 * 0.0085 = D * X_{D, e}$$

(2)

Solving equations (1) and (2), simultaneously, we get;

D = 195.1071 Kgmole / hr

B = 84.1928 Kgmole / hr

$$X_{D, e} = 0.0122$$

Similarly applying material balances for various components we get the composition of heavy and light key and non- key components.

		Distillate	Bottoms
Ethane	Light non-key (C2)	0.0122	0.0000
Propane	Light key (C3)	0.9831	0.0166
i- Butane	Heavy Key (C4)	0.0044	0.4049
n- Butane	Heavy non-Key (C4)	0.0003	0.5758
i- Pentane	Heavy non-Key (C5)	0.0000	0.0020
n- Pentane	Heavy non-Key (C5)	0.0000	0.0007
Total		1.0000	1.0000



From the values of Temperature and Pressure we can determine the values of K for different components using the K-value graph below.

 $\alpha$  = Volatility = (K-value of i<sup>th</sup> component) / (K-value of Heavy Key)

z = compressibility factor = Mole fraction of feed

 $Relative \ volatility = \frac{Volatility \ of \ i^{th} \ term \times Compressibility \ of \ i^{th} \ term}{Volatility \ of \ i^{th} \ term - \theta \ determined \ by \ trial \ method}$ 

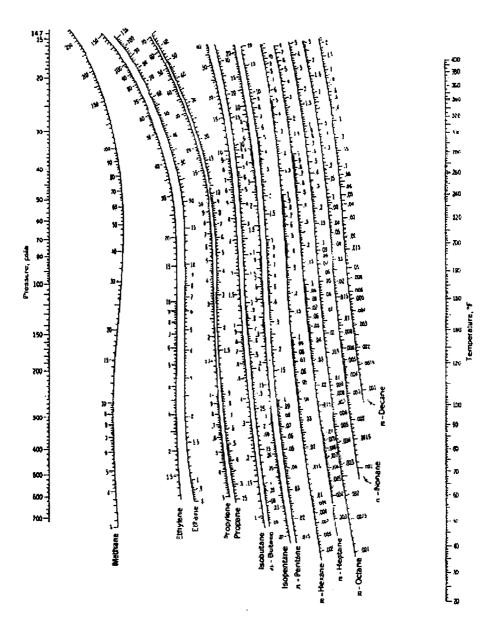


Figure 20. K-value Correlation Graph



## DETERMINING K - VALUES FROM GRAPH

	K	α	z Relati	ve Volatilities	$(\alpha^*Xd)/(\alpha-\theta)$
Ethane	3.00	4.8387	0.0085	0.01	0.02
Propane	1.26	2.0323	0.6917	1.78	2.53
i- Butane	0.62	1.0000	0.1251	-0.52	-0.02
n- Butane	0.47	0.7581	0.1739	-0.27	0.00
i- Pentane	0.22	0.3548	0.0006	0.00	0.00
n- Pentane	0.19	0.3065	0.0002	0.00	0.00
Total				1.00	2.53
		Тор	Middle	Bottom	
K (Propane Light Ko	ey)	0.95	1.26	2.24	
K (Butane Heavy Ke	еу)	0.4633	0.62	1.276	
$\alpha = K_{LK} / K_{HK}$		2.0505	2.0322	1.7554	

$$\beta_{LK/HK}$$
 \*  $(K_{top} HK) \land \theta_{LK} = K_{top} LK$ 

$$\beta_{LK/HK}$$
 \*  $(K_{bottom} HK) \land \theta_{LK} = K_{bottom} LK$ 

$$\beta_{LK/HK} * (0.4633) ^ \theta_{LK} = 0.95$$

$$\beta_{LK/HK} * (1.276) ^ \theta_{LK} = 2.24$$



Taking logarithm on both sides

$$LOG_{10} (\beta_{LK/HK}) + \theta_{LK} LOG_{10} (K_{top} HK) = LOG_{10} (K_{top} LK)$$

$$LOG_{10} (\beta_{LK/HK}) + \theta_{LK} LOG_{10} (K_{bottom} HK) = LOG_{10} (K_{bottom} LK)$$

$$LOG_{10} (\beta_{LK/HK}) + \theta_{LK} LOG_{10} (0.4633) = LOG_{10} (0.95)$$
(3)

$$LOG_{10} (\beta_{LK/HK}) + \theta_{LK} LOG_{10} (1.276) = LOG_{10} (2.24)$$
(4)

Solving (3) and (4) simultaneously, we get;

 $\theta_{LK}$ 

0.8466

BLK/HK

1.8223

 $\theta$  is determined such that the sum of the relative volatilities from the above table is as close to vapor feed which is 1.

 $\theta$  (by trial method) 1.2425

A' and B' are any Gilliand Constants

B' = 0.48 from Gilliand Curve i.e. figure 2.

$$A' = \frac{Reflux\ ratio - Minimum\ reflux}{Reflux\ ratio + 1}$$

A' = 0.2320

$$N_r/N_s = ((z_{LK} * (X_{B, LK})^2 * D) / (z_{HK} * (X_{D, HK})^2 * B))^{0.206}$$

$$N_r/N_s = 1.0268$$



$$\label{eq:Minimum number of stages} \begin{aligned} & \text{Minimum number of stages} = \frac{ln \frac{(X_{D,LK})(X_{B,HK})\theta_{LK}}{(X_{B,LK})(X_{D,HK})\theta_{LK}}}{ln \, \beta_{LK} \frac{lK}{HK}} \end{aligned}$$

= 13.1886

### Minimum number of stages = 13

Minimum Reflux = 
$$\sum (\alpha * Xd) / (\alpha - \theta) - 1$$
  
= 2.53 - 1

#### Minimum reflux = 1.53

Number of theoretical stages = 
$$\frac{B' + Minimum number of stages}{(1 - B')}$$
$$= 26.2859$$

### Number of theoretical stages = 26

## Actual number of stages = 38

Stripping Feed Tray Location = Actual Number of stages / 
$$(N_r/N_s + 1)$$
  
= 18.5266

## Stripping feed tray location = 19



Rectifying Feed Tray Location = Actual Number of Stages – Stripping Feed Tray Location

= 19.0246

**Rectifying feed tray location = 19** 

Reflux Ratio = Minimum Reflux \*1.5

Optimum reflux or Reflux ratio = 2.2935

#### 3.2 CALCULATION OF COLUMN INTERNALS

Plate spacing  $(l_t) = 24$  inches

Within the standard values

From Hysys,

First stage  $\rho_L = 459.7 \text{ Kg/m}^3$ 

First stage  $\rho_V = 34.18 \text{ Kg/m}^3$ 

Last stage  $\rho_L = 462.6 \text{ Kg/m}^3$ 

Last stage  $\rho_V = 40.99 \text{ Kg/m}^3$ 

Top Section Liquid flow rate (L) = 13865 Kg/hr

Top Section Vapor flow rate (V) = 22445 Kg/hr

Bottom Section L = 30558 Kg/hr

Bottom Section V = 25677 Kg/hr

Surface Tension First stage =  $4.685 \times 10^{-3} \text{ N/m}$ 

Surface Tension Last stage =  $3.881 \times 10^{-3} \text{ N/m}$ 

Top Molecular Weight = 44.15

Bottom Molecular Weight = 57.75



### 3.2.1 Calculation of Column Diameter

$$F_{LV} = \frac{L}{V} \sqrt{\frac{\rho_V}{\rho_L}}$$

$$F_{LV \, top} = \frac{L_{top \, section}}{V_{top \, section}} \sqrt{\frac{\rho_{V \, top}}{\rho_{L \, top}}}$$

$$F_{LV \text{ top}} = \frac{13865}{22445} \sqrt{\frac{34.18}{459.7}} = 0.19$$

$$F_{LV\,bottom} = \frac{L_{bottom\,section}}{V_{bottom\,section}} \sqrt{\frac{\rho_{V\,bottom}}{\rho_{L\,bottom}}}$$

$$F_{LV \text{ bottom}} = \frac{30558}{25677} \sqrt{\frac{40.99}{462.6}} = 0.38$$

By using figure 5,

$$K_{1 \text{ top}}$$
 after correction =  $8.1 \times 10^{-2} \times \left(\frac{4.685}{20}\right)^{0.2} = 0.0606$ 

$$K_{1 \text{ bottom}}$$
 after correction = 5.5 × 10<sup>-2</sup> ×  $\left(\frac{3.881}{20}\right)^{0.2}$  = 0.0397

$$u_f = K_1 \sqrt{\frac{\rho_L - \rho_V}{\rho_V}}$$

$$u_{f\,top} = \ K_{1\,top} \sqrt{\frac{\rho_{L\,top} - \ \rho_{V\,top}}{\rho_{V\,top}}}$$

$$u_{ftop} = 0.0606 \sqrt{\frac{459.7 - 34.18}{34.18}} = 0.0606 \times 3.5284 = 0.2138 \text{ m/sec}$$



$$u_{f\,bottom} = K_{1\,bottom} \sqrt{\frac{\rho_{L\,bottom} - \rho_{V\,bottom}}{\rho_{V\,bottom}}}$$

$$u_{f \, bottom} = 0.0396 \sqrt{\frac{462.6 - 40.99}{40.99}} = 0.0396 \times 3.2071 = 0.1274 \text{ m/sec}$$

$$\hat{u}_{top} = 85\% \text{ of } u_{ftop} = 0.1817 \text{ m/sec}$$

$$\hat{\mathbf{u}}_{\text{bottom}} = 85\% \text{ of } \mathbf{u}_{\text{fhottom}} = 0.1083 \text{ m/sec}$$

Maximum Volumetric Flow Rate,

$$Top = \frac{V_{top \ section} \ in \ Kg/hr}{\rho_{V \ top} \ in \ Kg/m^3 \times 3600}$$

Top = 
$$\frac{22445}{34.18 \times 3600}$$
 = 0.1824 m<sup>3</sup>/sec

$$Bottom = \frac{V_{bottom \, section} \, in \, Kg/hr}{\rho_{V \, bottom} \, in \, Kg/m^3 \times 3600}$$

Bottom = 
$$\frac{25677}{40.99 \times 3600}$$
 = 0.1739 m<sup>3</sup>/sec

Net Area (A<sub>n</sub>),

$$Top = \frac{Maximum\ Volumetric\ Flow\ Rate\ of\ Top}{\widehat{u}_{top}}$$

Top = 
$$\frac{0.1824}{0.1817}$$
 = 1.004 m<sup>2</sup>

$$Bottom = \frac{Maximum\ Volumetric\ Flow\ Rate\ of\ Bottom}{\hat{u}_{bottom}}$$

Bottom = 
$$\frac{0.1739}{0.1083}$$
 = 1.6057 m<sup>2</sup>



$$A_d = 0.12 A_C$$

$$A_n = A_C - A_d$$

$$A_n = A_C - 0.12 A_C$$

$$A_{C} = \frac{A_{n}}{0.88}$$

$$A_{C top} = \frac{A_{n top}}{0.88}$$

$$A_{C \text{ top}} = \frac{1.004}{0.88} = 1.141 \text{ m}^2$$

$$A_{\text{C bottom}} = \frac{A_{\text{n bottom}}}{0.88}$$

$$A_{C bottom} = \frac{1.6057}{0.88} = 1.824 \text{ m}^2$$

$$D_{C} = \sqrt{\frac{A_{C} \times 4}{\pi}}$$

$$D_{C top} = \sqrt{\frac{A_{C top} \times 4}{\pi}}$$

$$D_{C \text{ top}} = \sqrt{\frac{1.141 \times 4}{\pi}} = 1.205 \text{ m}$$

$$D_{C bottom} = \sqrt{\frac{A_{C bottom} \times 4}{\pi}}$$

$$D_{C \text{ bottom}} = \sqrt{\frac{1.824 \times 4}{\pi}} = 1.524 \text{ m}$$

Diameter of column = 1.524 m



#### 3.2.2 Calculation of Liquid Flow Arrangement

Maximum Volumetric Liquid Rate = 
$$\frac{L_{bottom \, section} \, in \, Kg/hr}{\rho_{L \, bottom} \times 3600}$$

Maximum Volumetric Liquid Rate = 
$$\frac{30558}{462.6 \times 3600}$$
 = 0.0183 m<sup>3</sup>/sec

From figure 6, at  $D_C = 1.524$  m and Liquid Rate = 0.0183  $m^3/\text{sec}$ .

We obtain Cross Flow (Single Pass) Liquid flow arrangement.

#### 3.2.3 Calculation of Provisional Plate Design

Column Diameter  $(D_C) = 1.524 \text{ m}$ 

Column Area  $(A_C) = 1.824 \text{ m}^2$ 

Downcomer Area  $(A_d) = 0.2189 \text{ m}^2 (12\% \text{ of Column Area})$ 

Note: As per Hysys Downcomer Area =  $0.3153 \text{ m}^2$  (17.3% of Column Area). We will be using this value for further calculation as Hysys is a simulator so as to obtain the desired result.

$$A_n = A_C - A_d = 1.824 - 0.3153$$

Net Area  $(A_n) = 1.5087 \text{ m}^2$ 

$$A_a = A_C - 2A_d = 1.824 - 0.6306$$

Active Area  $(A_a) = 1.194 \text{ m}^2$ 

Hole Area  $(A_h) = 0.1194 \text{ m}^2 (10\% \text{ of Active Area})$ 

Note: As per Hysys Hole Area = 0.1678m<sup>2</sup> (14.05% of Active Area in Hysys)

From the figure 9, Weir Length  $(l_W) = 50.44$  inches



Weir height  $(h_W) = 2$  inches Within the standard values

Hole size  $(d_h) = 0.1969$  inches Within the standard values

Hole pitch  $(l_p) = 0.5$  inches Within the standard values

Plate thickness= 0.1969 inches Within the standard values

The Material Used For Plates is Stainless Steel.

#### 3.2.4 Calculation to Check Weeping

 $\text{Maximum Liquid Rate} = \frac{L_{\text{bottom section in Kg/hr}}}{3600}$ 

Maximum Liquid Rate =  $\frac{30558}{3600}$  = 8.4883 Kg/sec

Minimum Liquid Rate = Turndown × Maximum Liquid Rate

Minimum Liquid Rate =  $0.50 \times 8.4883 = 4.2441$  Kg/sec

$$h_{ow} = 750 \left[ \frac{\text{Liquid Rate}}{\rho_{\text{Lhottom}} \times l_{\text{W}}} \right]^{2/3}$$

$$Maximum h_{ow} = 750 \left[ \frac{Maximum Liquid Rate}{\rho_{L bottom} \times l_W} \right]^{2/3}$$

Maximum 
$$h_{ow} = 750 \left[ \frac{8.4883}{462.6 \times 1.2812} \right]^{2/3} = 44 \text{ mm liquid}$$

Minimum 
$$h_{ow} = 750 \left[ \frac{4.2441}{462.6 \times 1.2812} \right]^{2/3} = 28 \text{ mm liquid}$$

Minimum  $h_W + h_{ow} = 51 + 28 = 79$  mm liquid

From figure 8,  $K_2 = 30.6$ 



$$\breve{u}h_{(min)} = \frac{[K_2 - 0.90(25.4 - d_h)]}{(\rho_{V \text{ bottom}})^{1/2}}$$

$$\tilde{u}h_{(min)} = \frac{[30.6 - 0.90(25.4 - 5)]}{(40.99)^{1/2}} = 1.9118 \text{ m/sec}$$

$$\label{eq:Actual Minimum Vapor Velocity} Actual \ \underbrace{\text{Minimum Vapor Velocity}}_{.} = \frac{\text{Turndown} \times \text{Maximum Volumetric flow rate}}{A_h}$$

Actual Minimum Vapor Velocity = 
$$\frac{0.50 \times 0.1739}{0.1678}$$
 = 0.5181 m/sec

So minimum operating rate will be well above Weep Point

#### 3.2.5 Calculation for Plate Pressure Drop

Maximum Vapor Velocity = 
$$\frac{0.1739}{0.1678}$$
 = 1.036 m/sec

From figure 12, using the values of

Plate Thickness / Hole Diameter = 1

$$A_h / A_p = A_h / A_a = 14.05$$

We obtain  $C_0 = 0.87$ 

$$\text{Dry Plate Drop } (h_d) = 51 \left[ \frac{u_h}{C_O} \right]^2 \frac{\rho_{v \text{ bottom}}}{\rho_{L \text{ bottom}}}$$

$$h_d = 51 \left[ \frac{1.036}{0.87} \right]^2 \frac{40.99}{462.6}$$

 $h_d = 6 mm$ 

Residual Head (h<sub>r</sub>) = 
$$\frac{12.5 \times 10^3}{\rho_{L \text{ bottom}}}$$



$$h_r = \frac{12.5 \times 10^3}{462.6}$$

$$h_r = 27 \text{ mm}$$

Total Drop 
$$(h_t) = h_d + (h_w + h_{ow}) + h_r$$

$$h_t = 6 + (51 + 44) + 27$$

$$h_t = 128 \text{ mm}$$

Total Plate Pressure Drop ( $\Delta P_t$ ) =  $9.81 \times 10^{-3} h_t \rho_{L \, bottom}$ 

$$\Delta P_t = 9.81 \times 10^{-3} \times 128 \times 462.6$$

$$\Delta P_t = 580.8776 \ Pa$$

#### 3.2.6 Calculation for Downcomer

Downcomer Clearance  $(h_{ap}) = h_w - 5 = 50.8 - 5$ 

 $h_{ap} = 45.8 \text{ mm}$ 

$$A_d = 0.3153 \text{ m}^2$$

Downcomer Clearance Area  $(A_{ap}) = l_w \times h_{ap} = 1.281 \times 0.0458$ 

$$A_{ap}=0.0587~m^2$$

A<sub>ap</sub> is less than A<sub>d</sub> hence we will use this in the formula to calculate head loss in the downcomer

 $Downcomer\ head\ loss\ (h_{dc}) = 166 \left[ \frac{Maximum\ liquid\ rate\ in\ Kg/sec}{\rho_{L\ bottom} A_{ap}} \right]^2$ 

$$h_{dc} = 166 \left[ \frac{8.4883}{462.6 \times 0.0587} \right]^2 = 16 \text{ mm}$$



Downcomer Backup  $(h_b) = h_t + (h_w + h_{ow}) + h_{dc}$ 

$$h_b = 128 + (51 + 44) + 16$$

 $h_b = 239 \text{ mm}$ 

$$h_b < \frac{1}{2}$$
 (plate spacing + weir height)

$$239 < \frac{1}{2}(609.6 + 50.8) = 239 < 330.2$$

Hence tray spacing is acceptable

$$Downcomer \ Residence \ Time \ (t_r) = \frac{A_d h_b \rho_{L \ bottom}}{L_{wd}}$$

$$t_{\rm r} = \frac{0.3153 \times 0.239 \times 462.6}{8.4883}$$

 $t_r = 4.1068 \text{ sec (Which is greater than 3 sec)}$ 

#### 3.2.7 Calculation of Entrainment

For  $F_{LV} = 0.38$  from figure 7,

 $\Psi = 0.005$  which is well below 0.1



#### 3.2.8 Calculation for Number of Holes

Area of One Hole = 
$$\frac{\pi}{4} \times d_h^2 = 1.9637 \times 10^{-5} \text{ m}^2$$

Number Of Holes = 
$$\frac{\text{Hole Area}}{\text{Area of One Hole}}$$

Number Of Holes = 
$$\frac{0.1678}{1.9637 \times 10^{-5}}$$

Number Of Holes = 8545



## 3.3 DATASHEETS OBTAINED FROM ASPEN HYSYS

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11	Condenser  I_Main TS	1 000 1 000	1 000	1.000	1.44		1 000				
13	2_Man TS	\$ 000	1 090	1 000	1 000	: 595	1 000				
14	3_Main TS	1.000	1 500	1 000	1 900)	1000	1,000				
15	4Man TS	1,000	1 900	1 000	1 000	1 (60)	1 000				
16	5_Man TS	1 000	1 000	1 000 1 000	1000	1 (A) (1 1 (30))	1 000 1 000				
17 18	6_Main TS 7_Main TS	1 000	1 000 1 000	1 000	1 (60)	t 3(0)	1 000				
19	8_Man 15	1 000	1 690	1 000	1.00		1.3490				
20	9_tlan TS	1 900	1 600	1.000	16%	1.34	1 (900				
21	10_litain TS	1 000	1 000	1 000	1,000	: 000	1 000				
222	11_ldain TS	1.000	1,000	1 000	1 000	1 (50)	1 000				
23	12_Main TS	1.000	1 600 1 600	1,000 1,000	1 000 1 000	1 000	1 600 1 600				
24 25	13_IAain TS 14_IAain TS	1 000	1 600	1 600	1 000	1,000	1 030				
28	15_Idam TS	1.000	1 000	1 600	1 000	1 000	1 000				
27	16_Idain TS	1.000	1 000	1 000	1000	1 600	1 000				
28	17_IAnn TS	1 000	1 800	1 900	14分	;	1 (00				
29	18_1Aan TS	1 000	1 000	1 000	1.007	1.74(4)	1,000				
30	19_Main TS	1 000	1 000	1 600	1 660	1 000 1 000	1 600 1 606				
31 32	20_Main TS 21_Main TS	1.000 1.000	1 000	1090	1000	1 000	1000				
33	22_Main TS	1.000	1 000	1 000	1 600	1.5%	1 000				
34	23_Main TS	1 000	1.000	1 600	1 (60)	1506	1 000				
35	24_Main TS	1 000	1 000	1 050	1 050	1 000	1 000				
35	25_Main TS	1 000	1 600	1 000	1.000	1.000	1 000				
37	26_IAam TS	1 (90)	1 60% 1 660	1 000	100	1.4.	1 600 1 600				
38 39	27tánn TS 28tánn TS	1 000 1 000	1 666	1669	1 660	1.0%	1 000				
40	29_ Main TS	1.000	1 000	1 000	1 000	1 000	1 000				
41	301Aain TS	1 000	1 000	1 000	1 000	1 000	1 000				
42	31_ldain TS	1 000	1 600	1 000	1,000	1,000	1 606				
43	32_IAbin TS	1 000	1 000	1 000	1,000	1 600	1 000				
44 45	33_Main TS	t 000	1 500	1 000	1 000	1.000	1 000				
46	34_Main TS 35_Main TS	1 000 E(h)(i	1 500 1 600	1 990	1989	1 (4)	1000				
47	36_Main TS	1,000	1 600	1 000	1 900	1 0.00	1 000				
48	37_Main TS	1000	1000	1 000	1000	1000	1 000				
49	38_Main TS	1.000	1 000	1.000	1 000	1,000	1.000				
50	Reboiler	1 000	1.500	1.000	1 000	1000	1 000				
51	Stages Condensor	Overati Efficiency 1 000	n-Pentane 1 000		1	<del> </del>					
52 53	I_Man T5	1 000	1 000		N-2Countries in a						
54	2_Man TS	1 000	100		P. Carlotte						
55	3_Main TS	1 000	1 (69)				1				
56	4_fAam TS	1 000	1 000								
56 57 58 59	5_Man TS	1 000	1 000	ł							
59	6_Main TS 7_Main TS	1 000	1 000			-					
60	8_Moen TS	1 000	1 000								
61	9_Madi TS	1 000	1 000								
62	10, Idain TS	1.000	1 000	L .		<u> </u>					
63	Hyprotech Ltd.		H	(SYS v3.1.3 (Build 4	827)	· · · · · · · · · · · · · · · · · · ·	Page 3 of 23.				
	Deemsed to TEALLE	IL:					The second of the same				



				Case Name CHECCUMENTS AND SETTING MULANALIK KANDENA DESKTOPM							
Н	YPROTI	TEAM LND ECH Dalgary, Albei	ta	Unit Set	51						
J.	C.7 * E V E ( 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	evitios Callada		Date/Time Thu Apr 23 02 14 21 2009							
			Di	istillati	on: T-100	) (continued)					
Ł				Stage Efficiencies							
t	Stages	Overall Efficiency	n-Pentane	Ottoge 200	:	l					
T	11_Main TS	1 000	1 000		1						
1	12_Main TS	1 000	1 000								
	13_Main TS	1,000	1 000								
	14_Main TS	1 000	1 000				į				
	15_Main TS	1 000	1 600		į		.				
	t6_Main TS	1 000	1 000				1				
	17_Main TS	1 000	1 366		•	1					
	18_Man TS	1 000	1 000								
١	19_Main TS	1 000	1 000		i	İ	: I				
	20_train TS	1 000	1 000		1						
	21_Main TS	1.000	1 000		1	1					
	22_tam TS	1 000	1 000								
	23_Main TS	1 000	1 000								
	24_Main TS	1 000	1 600				i				
	25_Main TS	1 000	1 690								
		1 900	1 000		1		İ				
	25_Main TS	1	1 000				i				
	27_Main TS	1 000	į.								
	28_Main TS	1 000	1000								
	29_Main TS	1 000	1 000	1	ì		ĺ				
ŀ	30IAain TS	1.000	1 000		•						
l	31_Main TS	1,000	1 000		į						
l	32_Main TS	1 000	1 000								
l	33_Main TS	1 000	1000								
l	34_Main TS	1 000	1000		-						
l	35Main TS	1000	1000								
l	35_Nam TS	1 000	1 000		: 1						
l	37_Main TS	1 000	1000				1				
l	38_ Main TS	1 000	1000				1				
	Reboiler	1.000	1 000								
				SOL	VER						
l			Column Solving	Algorithm	HYSIM Insklo-Out						
t		Solving Ópt	ons			Acceleration Par	amelers				
L	Maximum florations			10000	Augularata Kilisi	one & #4 Model For a motors		······································			
L	Equippoism Error T	oletance		1.00005							
L	Heat/Spec Error To	lorance		5 000e-004							
Ĺ	Save Solutions as t	ndal Estimate		Qn							
Ĺ	Super Critical Hand	iling Model		Simple K							
ł	Trace Level			Low							
Ļ	Init from Ideal K's			Off	<u> </u>	Damping Para	meters				
Ļ		al Estimate Genera	tor Parameters		Azeotrope Check						
ļ	Horative IEG (Good	for Chemicals'i		Off	Fixed Clamping F	Factor					
ł											
+					<u> </u>		······································				
1				SIDE ST	RIPPERS						
-				SIDE RE	CTIFIERS						
†			The state of the s	PUMP A	ROUNDS						
				VAP BY	PASSES						
٠.	Hyprotech Ltd.		11		3 (Bulki 4827)			Page 4 of			
ı	COMPRESSED IN SEC.	and the second second	n		O PENING ANE ()			2 · · ·			



t			Case N	ame: CNDCCUN	ÆN1	S AND SETTUKSSIMAYANIK KANODIAIOESKTOPIM				
2	TEAM LND  VYDROTECH Cabgary Alberts		Unit Set Si							
3	HYPROTECH Cabgary Alberto CANADA									
5			Date/Time: Thu Apr 23 02 14 21 2009							
S		-		T 400 /		mating and \				
7		L	vistillati	ion: T-100 (	CO	ntinuea)				
9										
10			RAT	ING						
11			Tray Sections							
12 13	Tray Section	Main TS @C	OLI		T					
14	Tray Clameter (m)	1 500	•							
15	Weir Height (m)	5,0004-00	2 .		_					
16	West Length (m)	1 200			╀	<u> </u>				
17	Tray Space (m)	0.5500			+					
18	Tray Volume (m2)	09719			$\vdash$					
19 20	Disable Heat Loss Calculations Heat Model	No.			+-					
21	Rating Calcutations	olf	<del>-  -</del>		T					
22	Tray Mold Up (m3)	8.8350-00	)2							
23				sels		***				
24			ves							
25	Vessel	Reboiler (80	OL1	Condenser @COL1	+					
25	Diameter (m)	1 193		1 193	+-					
27	Length (m)	2 000		1 789 2 000 *	+					
28 29	Volume (m3) Orientation	Horizon:	<u>.                                      </u>	Horizontal	+					
30	Vessel has a Boot	No	-	lio .	†					
31	Boot Diameter (m)	***		14.	$\top$					
32	Boot Length (m)				I					
33	Hold Up (m3)	1 000		1 000						
34		Other Eq	uipment In	Column Flowshee	≘ŧ					
35			· • • • • • • • • • • • • • • • • • • •	T						
36 37					*********	**************************************				
38			Pressur	e Profile						
39			Pressu	ra (kPa)		Pressure Drop (kPa)				
40	Condenser		1496	) kPa	•	40 00 kPa				
41	1_latin TS			OkPa	-	ତ ୫୯ <u>୦</u> ୫ କ <b>ନ</b> ର				
42	2_Main TS			l kPa		Ø 8300 €F.a				
43	3_tain TS			hPa		9 ±199 •± 0 € 8199 •€0				
44	4_Main TS 5_Main TS			2 kPa 3 kPa	-	୍ର ମେଶ - ମିଧ				
45	6_Main TS	l		4 APa		0 8108 kPa				
47	7_Main TS	1		5.kPa		∆ 8108 kPa				
48	8_Main TS	100	153	5 kPa		0.8008 kFla				
49	9 <sub></sub> Jain TS			6 kPa		© 8108 <b>kPa</b>				
50	10_Main TS	1		7 kPa		O SION APA				
51	11_Main TS			8RPa	i	1 8th's eFA √ 5th eFs				
34	12_Main T\$			98Pa 88Pa	ļ	್ ಪ್ರಾಥಾಗಿ ಶಿಪ್ಪು ಕೊತ್ತು				
55 55	13_Main TS 14Main TS	l		1 kPa	ĺ	⊈ 810° <b>⊾</b> #a				
55	15_Main TS			1 %Pa		C 8108 kPa				
56	16_Main T\$	1		2 kPa		© 8106 kPa				
57	17_Main TS		154	ЗкРа		0.8108 kFa				
58	18Main TS	Į		4 %P0		3 \$106 APA				
59	19_Main TS			SAPa		0.8908.8Pa				
60	20 Main 15			5 kPa curs		5.44/1.4Fa				
42 43 44 48 48 50 51 52 53 54 55 56 57 58 60 61 62	21_Main TS 22_Main TS			AAPa SaPa	ì	1 (2000 <b>- Fo</b> - 1 (2000 <b>- Fo</b>				
			1537 FF a HYSYS va.1.3 (Bulld 4827)			Page 5 of 23				
63	Hyprotech Ltd.	1	HYSYS va 1	.3 (Build 4827)		Page 5 of 23				



I			Case Name:	C (DOCUMENTS AND	O SETTBIGSIMAYAM KANODIA DESKTOPUN				
3	TEAM LIND  TEAM LIND		Unit Set. SI						
4	CANADA		Oxfe/Time: Thu Apr 23 02 14 21 2009						
6									
7		D	istillation:	T-100 (contir	nued)				
8	23 Main TS		1548 ¥Pa		> 2*69.4Fa				
10	24_Main TS		1549 KPa	. who we are	ህ 8106 እም <sub>ፅ</sub>				
111	25_Main TS	1	1549 kPa	į	0.8108 kP.s				
12	≥ _Main TS		155G-kPa		ପ୍ରଥମେ ଅନ୍ୟ				
13	27_Main TS		1561 kPs	Ì	0 \$108 +Pa				
14	28Main TS		1552 kPa		0.8105 kPa				
15	29_Main 15		1563 kPa	1	+ tip≥ + \$1.5				
16	30Main TS	į	1554 489	; ;	etre • Car				
17	31_Main TS	1	1554 ¥Pa		ି ଅପରେ •ମିଜ ଠି ଅପରେ •ମିଜ				
18	32_Main TS		1555 kPa						
19 20	33_Main TS	İ	1565 RPa		0.8108 kPa 0.8108 kPa				
120	34_Main TS		1657 kPa 1580 kPa		∪ සංග්ය ⊁ෙන ට වර්ගිය ⊁Fක				
21 22	35_Main TS	1	1588 kPa		0 0100 sea				
22 23	36_Main T\$		1558 kPa 1559 kPa		ପ ଅଟେକ କଥନ ପ ଅଟେକ କଥନ				
24	37_Main TS 38_Main TS	1	1560 kFa		0.0 00 00				
25	Reposer		1570 «Pa	• •	% ଏହି କଳିଲ				
26	TAN APPROXITE			· · · · · · · · · · · · · · · · · · ·					
27		Pre	essure Solving Op	tions					
28	Pressure Tolerance 1 000e-004 Pr	esswe Crop Tolerance	1.000e-004 * Can	nping Factor	1 CCD * Max Press serations 100 *				
29			PROPERTIES						
30			Branartine : £ @COL1						
31 32			perties : 6 @COL	.1					
32 33	Vapour/Phase Fraction	Overall 0.0000	Liquid Phase 1 0000						
34	Temporature (C)	65.60	65 (4)		j				
35	Pressue (NPa)	2040	20140						
38	Molas Flow (kgmole/h)	279.3	279 3						
37	Mass Flow (kg/h)	1.346e+004	1.3456+004						
38	Striftjeal Liq Vol Flow (m3/h)	25.49	25 40						
39	Molas Emhalpy (k.Mkgmolo)	-1 233 <del>6+</del> 005	-1 233e+665						
40	Mass Emhalpy (kJ/kg)	-2559	-2559						
41	Molar Entropy (k.t/kgmola-C)	107.4	\$07.4						
42	Mass Entropy (kJ/kg-C)	2 229	2 229		:				
43	Heat Flow (NJh)	-1: <b>44</b> 4e+007	-3 444+-007						
44	Molas Donarly (kginolerm3)	9 358	9 35.4 a						
45	Mass Density (kg/m3)	451.0	451 G						
46	Std Ideal Liq Mass Density (kg/m3)	528.0 521.6	528 0 531 6						
47	Liq Mass Density @Std Cond (kg/m3)	531 6 161 9	1619						
48	Molas Heat Capacity (kL/kgmole-C) Mass Heat Capacity (kL/kg-C)	3.360	3 360						
50	Thermal Conductivity (Win-K)	* 229e-002	7.229e-002		<u></u>				
51	Viscosity (cP)	8 0489-002	8 048A-002						
52	Surface Tension (dyne/cm)	3 736	3,736						
53	Molecular Weight	48 19	48 19						
54	Z Factor	7 740a-002	7 740e-002						
58		Pro	perties: 8 @CO	<u>L1</u>					
56		Overall	Vapour Phase	Liquid Phase					
57	Vapeun/Phase Fraction	0.0000	0.0000	1 0000					
58	Temperature (C)	42 52	42 52	42.52	**************************************				
59	Pressure: (MPa)	1490 1	1450	1490					
60	lablar Flow (kgmcla)hi	195.0	6.0000	195.0					
61	tass Flow (kg/hi	8580	0.0000	8580					
62	Skit kleat Liq Vol Flow (m3/h)	16 98	0 0000	16.95	Page 6 of 23				
63	Hyprolech Ud.		17375 v3.1.3 (Build o	90Z1)	Specified by user				

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\* Specified by user



4		_	Case Name.	C IDOCUMENTS AN	D SETTINGSMAYANK KANODIA/DESKTOPM
2	HYPROTECH Cazgary	4lbesta	Unit Set	\$I	
4 5	CARADA		Date/Time	Thu Apr 23 02 14 21 :	7000
<b>6</b> 7		Di	stillation:	T-100 (conti	nued)
ð ñ		Веало	ation I 9 @COI	: 4	
		Overall	rties: 8 @COL	Liquid Phase	
10 11	Molar Enthalp, (kJ/kgmole)	-1 176e+605	-1 035e+005	-1 1764 - (V.)	
12	Mass Enthalp, (kJ/kg)	-2073	-2376	-2573	<u> </u>
13	Molar Enricey (kJ/kgmole-C)	99.09	140 1	99.09	
14	Mass Entropy (kJ/kg-C)	2 252	3 252	125.	
15	Heat Fig. (A.40)	2 294 cm (n) 1	9,666	-2 <u>204</u>	The state of the s
115	Molar Density (kgmole/m3)	10.44	44.75.30	19.3	
17	Mass Density (kg/m3)	4621	32.92	402.1	i
18	Std Ideal Eig Mass Density (kg/m3)	505.2	502.3	505.2	
19	Liq Mass Density (I)Stil Cond (kg/m3)	566.9	505.0	505.0	
30	Motar Heat Capacity (kJ/kgmole-C)	1403	92.72	143 3	· · · · · · · · · · · · · · · · · · ·
21	Mass Host Capacity (kJ/kg-C)	3 190	2 122	3 190	
33	Thermal Conductivity (Wm-K)	8 4326-602	2 155%-002	8 <b>432</b> 6-607	
23	Viscosity (cP)	3 365+ 002	9 5534-003	6.3686-000	
24	Surface Tension (dyne/cm)	4 854		4 51 h	
25	Malecular Weight	43.99	43.70	45.90	
26	Z.Fáctor	5 464e 662	0.7536	5.404e-012	
<u> 27</u>		<del>-1</del>	rties : 10 @CC		
:8		Overall	Vapour Phase	Liquid Phase	
29	Vapour/Phase Fraction	0.0000	0,0000	1 0900	
30	Temperature (C)	94 32	9432	54.00	
31	Pressure. (SPa)	1570 *	1570	1570	
32	Molas Flow (kymcse/h)	84.25	0.0000	\$4.5	, AND AND AND AND AND AND AND AND AND AND
33	Mass Floo (kg/h)	4341	0 (%)	41	
34	Std Ideal Liq Vol Flow (m3/h)	\$511	0.0000	g 511	<u> </u>
35	Molar Emhaloy (kJ/kgmole)	-1 390e+005	-1 2456+005	-1 3904-006	
36	Mass Erahalpy (kJA-g)	-2406	-2157	-2400	
37	Malar Entropy (kJ/kgmole-C)	1033	144 7	1033	
32	Mass Entropy (kJ/kg-C)	1784	2 509	1.79.4	
39	Heat Floa (kJh)	-1 1710+607	0 0000	-1 171e+007	
40	Molas Density (kgmole/m3)	7.975	0.7163	7 975	
41	Mass Density (kg/m3)	402.9	41.24	462.0	
42	Statistical Liq Mass Density (Lg/m3)	5735	5717	573%	and an opposite an amount of the first of the amount of th
-	Liq Mass Density @Std Cond (kg/m3)	575.0	5733	575.0	
43	Molar Heat Capacity (kul/kgmsle-C)	1925	136.5	192 5	
44 45	Mass Heat Capacity (kJ/kg-C)	3 323	2 365	3 323	
	Thermal Conductivity (W/m-K)	6 2959-002	2 4529-002	6 255e-002	
46 47	Viscosity (cP)	8 844e-002	1.0389-602	8.8446-002	harana a a a a a a a a a a a a a a a a a
48	Surfaco Tensson (dyna/cm)	3 639	***	3 639	
46	Molecular Wealth	57.93	57.72	57 93	
540	2 Pactor	6 444e-500	@ #1*4	P 4434990	
51					enterente en en en enterente en entre en entre en entre en en entre en entre en entre en entre en entre entr
52			SUMMARY		
53	Figw Basis.		1/colar	The comp	estion option is selected
54			Feed Compositi	on	
55	6				
55 56	Flow Rate (kgmoloth) 279.3	206			
57	-44				
58	Ethane 0.00	1			1
<b>30</b>	Propane 0.60	•			į
90	-Butano 0:2	<b>§</b>	İ	1	
51	n-Butano 0 17:	į.			
52	Percane 0.00	i i			
53	Hyprotech Lid.		SYS v3.1,3 (Bulki)	4827)	Page 7 of 23
-	Interest to TEAMINE	111	THE CALL DIST PROPERTY.	2.9.19.2. Francisci com de la como de la com	' opecified by user



J			Case Nan	N C VDC/CUMEN	TS AND SETTURGSMAYA	K KANCDIA/CESKTOPIM					
2		TEAM UND		Usuf Sot SI							
	<b>TYPROTECH</b>	Cospony Alberta CastaDa	Lean Sot								
当			Cate/Time	Cats/Time Thii Apr 23 02 14 21 2000							
<u>\$</u>			PA 2443 -1	- T 400 /	4:						
			Distillatio	on: T-100 (co	intinued)						
9											
10		***	SUMM	481							
11		6									
12 13	n-Pertano Flow Basis	0.0000	1,5olar	The	dimeres transplantes select	te.t					
14	1 PAN DOSIG		Feed Fl								
15		Ø									
16	Flow Rate (kgmole/h)	279.3000			:						
17 18	Ethane (hgmc/afh)	2 3741									
16	Propose (kgmoleft)	193 1918									
20	i-Butane (komokuh)	34 9404									
21	n-Butarin (kgmolerh)	48 5703									
22	i-Pertane (kgmole/h)	0 1976									
23 24	n-Pentane (kgmele/h)	0 0559	Produ	cts							
25	Flow Basis		litotar		compression aption is select	le(1					
26			Product Com	positions							
27	Dan Dan Taran	8	10								
28 29	Flow Rate (kgmc/e/h)	195 047:	84 2528								
30	Ethane	0 0122	0 (000)								
31	Propane	0 9833	0.0166								
32	i-Butane	0.0044	0.4945								
33	n-Butane	0.0001	0.5762								
34 35	⊩Pentane	0 0000	9 (929 9 0007								
35	Flow Basis		1,fölar	The	) сатровлют оснол ів веж	led:					
33			Product	Flows							
38		8 105 2172	10 84 3528								
3 <u>9</u>	Flow Rate (kgmole/h)	195 0472	84 3548								
40	Ethane (kgmc/e/h)	2 3740	e se se production de la constantion de la const								
42	Propane (kg/mole/h)	191 7935 -	1 - 363			: :					
43	1-Butane (kgmoleth)	0.6581	04 0823								
44	n-Butano (Agmole/h)	0.0215	49 5488								
45 46	i-Pontane (kgmolefi) n-Pontane (kgmolefi)	0.0000	0.0359								
47	Flow Basis		k#≾lar		) दलगाइनक्रमीला द्रक् <b>र</b> ात्रक्ष क्रमीण	:1र्ज					
48	-		Product Re	coverles							
49	Shan Data diameters	8	10 64 2528								
50 51	Flow Rate (kgmcle/h)	195 0472	C4 73.74								
51 52	Ethans (%)	100,0000	0.0000								
63	Propone (%)	99 2762	0.7230								
54	(%)	2 4560	97 5440		BROWN DA						
55	n-Butores (%) i-Pontano (%)	0 0443 0 0050	94 9537 169 6000								
56 57	n-Pontano (%)	0.000	100 0000								
57 58 59			COLUMN P	ROEII ES							
60	Roflina Ratio	1 589 Repute Fran	THE RESERVOIS ASSESSMENT OF THE PERSON OF TH	The Flore Cobbbs	Solosto for S. Paris	M.C.					
61 62			Column Prof	IIOS FIOWS							
63	Hyprotech Ud.		HYSYS.v3.1.3	(Bulli 4827)		Page 8 of 23					
	Livensed to TEAM LND		7 / 4 7 7 70, 240, 250			1 Specified by week					



CADOCUMENTS AND SETUNGSMAYARK KANODAYDESKTOPM Case Hame TEAM UND Calgary Alberta Und Set CACCADA Thu Apr 20 02 14 21 2 69 Cate Time €. Distillation: T-100 (continued) **COLUMN PROFILES** Net Lig (kgmoleth) | Net Vap (kgmoleth) | flet Feed (kgmole h) | flet Dears (kgmoleth) Pressure (APa) Temperature (C) 12 Condenser 42.52 14:4: 3:43 534. 13 1500 1\_\_Main T5 44 53 500 2\_\_Main 15 313 7 44 89 1531 3\_ Main TS 45 15 1532 313.2 508 f 4\_Main TS 45 42 1532 3125 5083 5/37 G 5\_\_fAain TS 45 72 1533 3117 500 7 6 Main TS 46 07 1534 3107 1535 300 € 555 E 7 Main TS 45 47 5/34 B 8\_Main TS 46 90 1536 N/8 2 9\_\_Main TS 47 46 1536 000.7 **5**23 3 10\_Main 15 48 (6) 1537 4, 4, 5 7.71 11\_Main TS 48 73 1538 303 1 5000 12\_Main TS 49.47 456 1 1530 301.0 ---10\_Main TS 50.08 4185 O 1540 238.7 14\_Main TS 51 16 1541 2563 493 8 15\_Main 18 1541 2938 491.4 16\_Main 18 53 07 1540 291.2 489 8 17\_\_Main 15 54 03 1543 288.6 496.2 18\_ Main 78 55 10 1544 226.5 483.6 19 Main 78 500.9 1535 481 55.11 20\_Main TS 452.5 537.1 56.33 1545 21\_Main TS 56 50 1549 537 2 452.9 22\_Main TS 56 67 1547 537.0 452.9 35 23\_Main 15 56 91 1549 536 5 452.8 24\_ Main 15 57 27 1549 535 8 452.3 37 25\_Main TS 451.5 15.17 5345 57.82 26\_Main TS 1550 532 5 450-2 58 66 529.8 449.3 27\_Main TS 59 94 1551 1552 576 4 445.6 26 Main TS 61 81 1553 522 6 442 1 29\_Main 15 54 40 67 75 1554 5193 439.3 30\_Main TS 1554 5173 235 B 31\_Main TS 71 72 433 1 75 97 1555 517 4 32\_Main TS 433 1 33\_Main TS 80 10 1555 519 1 434 9 521 9 46 34\_Main TS 83 77 1557 437.7 47 534 3 1558 35\_Mairi TS 86 83 527 1 4400 48 1959 35 Main TS 89 Y 48 1555 246.2 44 37\_Main 19 94.30 50 38\_\_Main 15 1560 129.1 444 6 9273 51 1570 444 9 Reboiler 64 32 52 Column Profiles Energy Liquid Enthalpy (kul/kgmole) Vapour Enthalpy (k,l/kgmic/e) Heat Loss (kJ/h) Temperature (C) Condenser 42 52 -1 176e+005 -1 935e+005 1\_Main TS -1 042g+005 44 53 -1 175a+005 4 544e+0/5 -1.178a+005 2\_Main TS 44 69 1.4700.65 -1.1/29+605 3\_blain TS 45 15 4.6486+9.5 41 182e+305 4 JAnn TS 45 47 -3-64"e+00.5 -1 155e+305 5\_ Man 15 45 75 60 6 Main TS -1 187e+005 4 64(0)+005 46 07 31 4 050s4005 7\_Main 75 46 47 -1 1516+005 -1 0518+00A a Man TS 46 93 -1 19401905 Page 9 of 23 Hyprotech Ltd. HYSYS v3.1.3 (Build 4827) Specified by user Licensed to TEAM LND



CODOCUMENTS AND SETTINGSMAYARD KANDDIADESKTOPM Casa Hama that Set Catgar, Alberta CASSADIA Date/Time Triu Apr 23 02 14 21 2000 Distillation: T-100 (continued) **COLUMN PROFILES** 10 Heat Loss (kith) Vapour Enthaley (£34 grocks Liquid Enthaloy (k.t/kgm/de) Temperature (C) 9\_ Main TS 2" 4 13 -1 203m+ 5.5 16\_\_Main 18 45 (4) 29 Cr. Walter 2 14 45 73 -1.203-4-005 11 Main 15 -1.213±+005 41 0004+005 12 Main 15 49.47 4 000 arch -1-218-00° 13\_Main 75 50.28 3 055a+005 14\_Main TS 51 16 -1-224e+005 41 06564 005 15\_Main TS 52 09 -1 229e+005 -1 234e+565 -\$ 970e+165 16\_Main TS 530 35000 17\_Blain TS -1 236s+165 5.4 GB 18\_Main TS is @7464 × 5 -1 2440+365 55 10 19 Main TS 56 11 1.4840.005 15 3760 + 3.5 20\_Main 15 56 33 -1-248e+005 -t 077e+3/6 21\_Main TS 56 50 -1 2496-605 -1 078e+905 22 Main TS 56.67 -1.250e+005 # 079w+605 26 27 28 29 30 +1-079++005 23 Main TS FR G1 of 05169005 24 Main 75 -1 253e+905 ~1.000++005 57.27 25\_Main TS 57 82 -1.256e+009 -3 **08**3e+965 -1.261e+005 A 0864-000 29\_Main TS 58 66 27\_Main TS 59 94 11 7560+19F A Gata-1055 28\_Main TS 61 61 -1.2800+305 14 000 Barrio 15 32 4 1116+075 29\_\_Main TS €4 40 -1 2946+609 -1 126e+005 30\_Main TS 67.75 -1 5119+005 -1-145e+906 -1.330++005 31\_Main TS 71 72 32\_Main TS -1 348e+305 4 167e+035 75 97 -1 363e+005 4 1866-005 33\_Main TS 80 10 83 77 -1 376e+006 4 207e+x5 34 Main TS -1 38 50 - 300 64. E3 . 1 7 2 3 1 4 × 16 35\_Main TS 4 3,890,000,000 3 2340 ere 6 35\_Main TS 40.37 40 -1-391e+005 -1 241e485 37\_Main TS 91 19 41 -1 397e+395 -8 244e+006 62 73 39\_ Main TS -1 245e+005 42 94 32 -1 3506+005 Robolio 43 FEEDS / PRODUCTS 44 Molar 45 Flow Basis Floars (lugmodent Enthalpy ik.2%gmca Temp 1C6 Type Chity (hillin) State 46 Stream 6.7480+006 47 Energy Condenser 48 DIAN Legist 156 0 49 I\_Man TS 50 2\_Main TS 51 3\_Main TS 52 4 Man 15 53 5\_Main TS 54 6\_Main TS 55 Main TS 56 57 Main TS SR 10\_ Idam TS 59 11 Main TS 12\_Ham TS 13\_lain TS 14 Main 75 Page 10 of 23 HYSYS v3.1.3 (Bulke 4827) Hyprotech Ltd. " Egypt theat by user Lecenand to TEAM NO



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┨		Case Hamo	O ECCUMENTS AND SETTINGEN	MARINE KAINGUAQESKTOF		
H	TEAM UNIC	Unit Set	SI			
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┝						
		Distillation: T	-100 (continued)			
L		DVNAMCE				
┞		DYNAMICS				
١	V	essel Dynamic Specific	ations			
I	Vessel	Reboter (\$0.09,1	Condonser (\$1.55)			
L	Diameter (m)	1 193	Let :			
╀	Height 0 (m) Volume 0 (m3)	1,560	2000			
t	Volume 0 (m3) Liquid Volume Percent (%)	(0.00)	50.00			
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+	Fixed P Spec Active	Not Active	Het Acteu			
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İ				A		
1		Holdup Details				
1						
4		Pressure (APA)	र्राट्सिक्स <b>म</b> जन्महोत	toria gradus en last <del>a</del> emito		
ł	Condenses	0.0000	0.000	· · · · · · · · · · · · · · · · · · ·		
ł	Condenser 1_Max T5	0.0000	0 0000			
1	7_Man TS	0.0000	0.0000	<b>4</b> + 40		
	3_Main TS	6.0000	0.6000	· ·		
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	5_Main TS	6 <b>600</b> 0	0.0000	14		
4	6_Main TS	0.0000	0.0000	***		
4	7_Main T5	0.0000	0.4465 1.76679			
4	8_ Main TS 9_ Main TS	0.0000	0.9999	•••		
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3	Hyprotech Ud.	HY5YS v3.1.3 (Build 45	327)	Page 12 of		

# DESIGNING OF DEPROPANIZER TRAY COLUMN USING ASPEN HYSYS AS A SIMULATOR

Ţ		Case Hante	C DOCUMENTS AND SET 1,003	mayara kanoqia:CES#TOP
}	TEAM LIST	Unit Set	SI	
7	CANADA	Cate/Time	Thu Apr 23 02 14 21 2009	
J		Distillation	T-100 (continued)	
		Distillation.	1-100 (Continued)	
		Phospate «KPB»	\$ 250 miles (m) 25	Elektropist Legistis Mi
•	24_Main TS	iji (sijesi)	0.000	
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	HYPROTECH	Calgary Alberta	U	Init Set	51			
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5 £			<u></u>					
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8								
9				SETUP				
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11 12	Tray Section	Main TS SCOL1	Liquid Draw		200 - 1	Court of the court		
13		Section_1						
14	Section Start	1,	Main is		<u> </u>	<u> </u>		
15	Section End	38_1	Mam 75	<del></del>		***************************************		
16	Internals		Seco					
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27	Internals		Sieve Design			<u> </u>		
28 20	Mode  Number of Flow Paths		U-1-1-11		THE REPORT OF THE PERSON NAMED IN			
30	Section Diameter (m)	1			·			
31	Tray for Properbas		783					
32	Tray Spacing (mm)		603 6	•				
33	Tray Thickness (mm)		5 000					
34	Featury Factor		1 000	-	······			
35	Max Delta P (hi of liq)		203.2 mm	•				
36	Max Flooding (%)		85 00	•				
37	Packing Correlation HETP (m)							
38	HETP (m) Packing Type							
40	r acking 175	·						
41			IRA	Y INTERNALS			-	
42		Şe.	ction_1					
43	Section Start		1_Han TS					
44	Section End		38_fdam TS					
45	Internals		5:876			***************************************		
46	Seeve Hole Pitch	(mm)	1270			***************************************	_	***************************************
47 48	State Halo Diameter Valve Mari Density (	umm) (kgmp3)	300				1	
49	Valve Mari Thickness	(mm)						
50	Hose Alea (% of AA)	(%)						
51	Valve Onlice Type							
52	Valve Design Marsial							
53	Bulktre Cap Skit Hergiti	(mm)					<del>-  </del>	
54	Side Wea Type	17000	50 80 °	**************************************				
56 56	Wour Height Man Won Loading (in	(mm)	89 42 *					
57	Downcomer Type		Vorpcal					
58	Downcomer Clearance	(mm)	45.80				and a second second	
59	Max DC Backup	(%)	50 60 .					
60	State DC Top Width	(mm)				····		
61	Side DC Bettern Width	(mm)						
62	Centre DC Top Width	(mm)				· · · · · · · · · · · · · · · · · · ·		Dear to see
63	Hyprotech Lid.		HYSY	5 v3.1.3 (Bulk) 48	21)			Page 14 of 23 per flest by user
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	_			Casa Name	C (DOCUM	ents and set hassma	YALIK KALIODIAIDESK <b>TOP</b> U
3 3	AYPROTECH	TEAM LTIC Caspaty A	1	Unit Set	Şı		and the second s
4	Corrects Innovation	CANADA		Date/Time	Thu Apr 23	02 14 21 2889	
5							
€ 7 8		Tray	Sizing: Tray S	Sizing-1 (	continu	ied)	
9	Centre DC Bottom Worth	(mm)					
10	O C DC Top Width	(mm)					
15	O.C. DC Bostom Width	(mm)	4.51				
12	QISIDO Top Width	(mm)	Nº 4				
13	O.S. DO Bottom Width	(mm)		<u> </u>			
14 15			•	TRAY RESULTS	6		
16			Skatisn <sub>u</sub> 1	<u> </u>			
17	Section Start		t_ Main TS				
18	Section End		38_Man TS	<u> </u>			
19	Internals		Sveve				
20	Section Crameter	im)	1 524	<u> </u>			
2	Max Flooding	196)	77.65	<u> </u>			
22	X-Sectional Area	(m2)	1 824	<b></b>			
23	Section Height	(m)	23.16 8.625	1			
24 25	Section DeltaP	ikPai	2050	<del> </del>			
35	Flow Length	(mm)	825.5	1		error en en en en en en en en en en en en en	
27	Flow Wish	(mm)	1446				
28	Max DC Backup	146)	36 30	1			
29	Max Weir Load	im3/h-m)	§1 <b>5</b> 6				
30	Max DPI Tray	(NPa)	0 2490				
31	Tray Spacing	(mm)	609 6 *				
32	Total Weir Length	(mm)	1281				
33	West Height	(mm)	<u>50.80°</u>	<u> </u>		all the second statement and the second seco	
34	Active Area	(m2)	1 154	1		<del></del>	:
35	DC Clearance	(mm)	45.80 *				
36	DC Area	(m2)	<u>0 3153</u> 1 281	1			
37 38	Side West Length Hote Area	(m2)	0 1678				
39	Estumpted of of Hotes / Ally		8545				
40	Renet Area	(m2)	0.0000				
41	Rebet - S	(mm)	.41				
42	Relief - #	imm)					
43	Robet - B	(mm)					
44	Side DC Top Width	rmarij	349.3				
45	Side DC Bim Width	(mm)	349.3				
46	Side DC Top Length	(m)	1 281			***************************************	
47	Side DC Blm Length	(m)	1 281 0 3153				
48	Side DC Top Area	(m2)	0.3153	<del> </del>		······································	
49 50	Side DC Elim Area Centre DC Top Width	(m2) (mm)	0.0000	1		**************************************	
51	Centre C4C Btm Visith	(mm)	0.0000				
52	Contro DC Top Length	(m)	0.0000				
53	Centro DC Btm Length	(m)	0.0000				
54	Centre DC Top Area	(m2)	0 0000	1			
55	Contre DC Btm Aroa	(m2)	0 0000				
50	OC DC Top Width	(mm)	0,6000				
57	O C DC Birn Width	(mm)	0 0000				
58	DC DC Top Length	(m)	0.000	-		<del>ndinamananan</del> an etimber etimber etim maaritim maaritim maaritim et	A N. A PANCEMBA, INC. PERSONAL PROPERTY AND AND AND AND AND AND AND AND AND AND
59	O.C. DC Birm Length	(81)	0.7650	-}			
60	O.C. DiC Top Area	(m2)	(1.00)(k)(j.) 	+			
61	OC DC 8tm Area	(m2)	0.0000	1			
62 63	the state of the s	(mm)	9 (9000 HV)	1 SY5 v3.1.3 (Bulld	4827)		Page 15 of 23
100	Lisensed to TEAM LND	<del>49</del>	01:	MIN AN' I'N (DANN	777.1	······································	eserved by user



耳			Case Name:	0.000	AMENTS AND SET	Tergsurayarak Kanioqraic	ESKTOPW
3		M LNC pary Alberta	Und Set	SI			
4	CAPETO INCOME. CAP	:ADA	Cate-Fim-	Tիև Ap	23 02 14 21 20A/4		
5 E		l			<del></del>		
7	Tr	ay Sizing: Tray :	Sizing-1	(conti	nued)		
9	O'S DC Bitm Width (mi	n) 0.0000					
10	OS DC Top Length ()	n) 0.6000			-		
11		0.0000					
븯	OS DC Top Area (m						·····
13	GISI DC Btm Area (m		L		<u> </u>		
15		P/	ACKED RES	JLTS			
16		Section_1					
<u>17</u>	Section Start	1_ tisa					
18	Section End	38_Ms			· A		
19	Internals		5 624		<u></u> :		
20	Section Diameter (m) Max Flooding (%)		72.05	<b></b>	·		
<del>2</del> 2	X-Sectional Acea (m2)	1	1 824				
23	Section Height (m)		23 16				
24	Section DeltaP (kPa)		8.425				
25	DP per Length (kPa/m)						
26	Flood Gas Velocity (m3/h-m2)		•••				
27	Flood Gas Velocity (m/s)					***************************************	
28	Estimated # Pieces of Packing				· · · · · · · · · · · · · · · · · · ·		
29	Estimated Mass of Packing (kg)						······································
31	Estimated Packing Cost (US\$) HETP (m)		<u> </u>		· · · · · · · · · · · · · · · · · · ·		
32	HETP Conelation					l periodici mentre anticologica con esta de la malegació de marcia de la cologica de marciales cologica.	·
33							
-3-7	Packing Correlation	1					
34	Packing Correlation Packing Type		***				
-		TRAY PRESSURE D					
34 35		TRAY PRESSURE D					
34		TRAY PRESSURE D		1			
34			DROPS		of bod i	Esy Delta P 101 of 6	qı
34 35 36 37 38 39		TRAY PRESSURE D	DROPS	1 Çalta P (ht ç (mo)	ot pod i	Ear Delta Frint of S	41
<del>                                     </del>	Packing Type	Detta P	DROPS Section_	Cielta Pinte	9 (bq)		5.935
<del> </del>	Packing Type  F_Man TS	Deta P (kFa)	Section_	Cielta Pinte			9 939 5 9 <b>38</b>
<del> </del>	Packing Type  F_Man TS	Detta P (kPa) 9 30	Section_	Cielta Pinte	45 41 45 43 46 43		9 935 5 938 5 933
<del> </del>	Packing Type  F_Man TS	Deta P (ISB) 0 20 0 30 0 30 0 30	DROPS Section_ 092 092 093 093	Cielta Pinte	46 41 46 43 46 43 46 43		0 935 5 938 5 933 5 935
<del> </del>	Packing Type  F_Man TS	Deta P (\$F5) 0 X 0 X 0 X 0 X	DROPS Section_ 092 093 093 094	Cielta Pinte	46 41 46 41 46 41 46 41 46 41		0 005 5 038 5 033 5 025 5 916
   기 의 의 가 의 의 왕 국	Packing Type  F_Man TS	Deta P (\$50) 0 20 0 20 0 20 0 20 0 20	DROPS Section_ 092 093 093 094 095	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 39		0.935 5.938 5.933 5.925 5.916 6.908
<del> </del>	Packing Type  F_Man TS	Deta P (EPs) 0 20 0 20 0 20 0 20 0 20 0 20 0 20	DROPS  Section_  000 000 000 000 000 000 000 000 000	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 30 46 30		0 005 5 038 5 033 5 025 5 916
<del> </del>	Packing Type  F_Man TS	Deta P (\$50) 0 20 0 20 0 20 0 20 0 20	DROPS  Section_  092 093 093 094 095 096	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 30		0 935 5 938 5 933 5 925 5 916 5 909 5 894
<del> </del>	Packing Type  F_Man TS	Deta P (kFa) 0 % 0 % 0 % 0 % 0 % 0 %	DROPS  Section_ 092 093 093 094 095 096 096 007	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 30 46 30 46 30		0 035 5 938 5 933 5 935 5 916 6 909 5 834 5 083
<del> </del>	Packing Type  F_Man TS	Deta P (AFa) 0 % 0 % 0 % 0 % 0 % 0 % 0 %	DROPS  Section_  092 002 093 093 094 095 095 096 097	Cielta Pinte	45 41 45 41 46 41 46 41 46 40 46 30 46 30 46 30 47 31 47 31 48 30		0 035 5 938 5 933 5 975 5 916 6 909 5 834 5 834 5 838 5 848 5 848 5 848
<del> </del>	Packing Type  F_Man TS	Deta P (NFa)  0 20 0 20 0 20 0 20 0 20 0 20 0 20 0	DROPS  Section_  092 002 003 093 093 094 055 095 096 007 100 102	Cielta Pinte	45 41 45 41 46 41 46 41 46 40 46 39 46 30 46 36 47 31 48 30 48 28		0 035 5 938 5 933 5 975 5 916 6 909 5 894 5 884 1 956 5 648 5 829 5 808
<del> </del>	Packing Type  F_Man TS	Deta P (NFa)  0 20 0 20 0 20 0 20 0 20 0 20 0 20 0	DROPS  Section_  092 002 0093 0093 0093 0095 0095 0096 0096 0097 1000 1002 1003	Cielta Pinte	45 41 45 45 46 45 46 45 46 30 46 30 46 30 46 30 47 31 48 30 48 28 48 28 48 28		0.035 5.938 5.933 5.925 5.916 6.908 5.834 5.838 5.848 5.829 5.808 5.784
<del> </del>	Packing Type  F_Man TS	Deta P (NFa)  0 20	DROPS  Section_  092 002 093 093 094 055 095 096 097 100 102 103 106	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 30 46 30 46 30 47 31 48 30 46 38 46 28 46 28 46 22		0.035 5.938 5.933 5.975 5.916 6.908 5.834 5.838 5.848 5.829 5.808 5.784 6.758
<del> </del>	Packing Type  F_Man TS	Deta P (NPa) 0 20 0 20 0 20 0 20 0 20 0 20 0 20 0 2	DROPS  Section_  092 000 000 000 000 000 000 000 000 00	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 30 46 30 46 36 47 31 48 30 46 28 46 28 46 22 46 19		0.035 5.938 5.933 5.925 5.916 6.909 5.834 0.850 5.648 5.829 5.808 5.758 5.758
<del> </del>	Packing Type  F_Man TS	Deta P (NPa) 0 20 0 20 0 20 0 20 0 20 0 20 0 20 0 2	DROPS  Section_  092 000 000 000 000 000 000 000 000 00	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 30 46 30 46 30 47 31 48 30 46 38 46 28 46 28 46 22		0.035 5.938 5.933 5.975 5.916 6.908 5.834 5.838 5.848 5.829 5.808 5.784 6.758
   기 의 의 가 의 의 왕 국	Packing Type  F_Man TS	Deta P (NPa) 0 20 0 20 0 20 0 20 0 20 0 20 0 20 0 2	DROPS  Section_  092 000 093 094 095 096 096 096 096 100 100 100 100 100 100 100 100 100 10	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 30 46 30 46 36 47 34 48 28 46 28 46 28 46 22 46 15		0.035 5.938 5.933 5.975 5.916 5.894 5.883 0.866 5.648 5.629 5.808 5.788 5.758 5.700
   기 의 의 가 의 의 왕 국	Packing Type  F_Man TS	Deta P (NPa)  0 20 0 20 0 20 0 20 0 20 0 20 0 20 0	DROPS  Section_  092 000 093 094 095 096 096 096 096 100 100 100 100 100 100 100 100 100 10	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 30 46 30 46 36 47 34 48 28 46 28 46 28 46 28 46 28 46 28 46 28 46 28 46 28 46 28 46 28 46 30 46 30		0.035 5.938 5.933 5.975 5.916 5.934 5.833 5.835 5.848 5.736 5.736 5.736 5.707 5.672
   기 의 의 가 의 의 왕 국	Packing Type  F_Man TS	Deta P (NFa)  0 20 0 20 0 20 0 20 0 20 0 20 0 20 0	DROPS  Section_  092 000 0993 0993 0994 095 096 096 096 096 007 100 100 100 100 100 100 111	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 30 46 30 46 30 46 30 46 38 46 28 46 22 46 10 46 11 46 15		0.035 5.938 5.933 5.975 5.916 6.909 5.830 6.648 6.629 5.808 5.784 6.758 7.700 5.672 6.672 6.672 6.641 4.405 5.004
<del> </del>	Packing Type  F_Man TS	Deta P (NFA)  0 20  0 30	DROPS  Section_  092 000 0993 0993 0994 095 096 096 096 097 110 110 1993	Cielta Pinte	46 41 46 41 46 41 46 41 46 40 46 36 46 36 46 36 46 36 46 36 46 35 46 35 46 32 46 19 46 11 47 37 46 11 47 37 46 18 51 18 52 20		0.035 5.938 5.933 5.935 5.916 6.909 5.831 6.832 6.838
<del> </del>	Packing Type    Main TS	Deta P (NFA)  0 20	DROPS  Section_  092 000 093 093 094 095 096 096 096 096 096 100 100 100 100 111 100 100 111 100 1	Çelta Pintiş	46 41 46 41 46 41 46 41 46 40 46 30 46 30 46 30 46 30 46 30 46 30 46 25 46 72 46 10 46 11 47 32 46 11 48 11 48 12 48 13 48 14 48 15 48 15 48 15 48 16 48 br>48 48 48 48 48 48 48 48 48 48 48 48		0.035 5.938 5.933 5.935 5.916 6.906 5.834 6.838 6.839 5.808 5.784 6.758 7.702 6.672 6.641 4.704 5.004 5.009
<del>지</del>	Packing Type  F_Man TS	Deta P (NFA)  0 20	DROPS  Section_  092 0002 0002 0003 0004 0005 0006 0006 0006 0006 0006 0006	Çelta Pintiş	46 41 46 41 46 41 46 41 46 40 46 36 46 36 46 36 46 36 46 36 46 35 46 35 46 35 46 11 47 47 47 47 47 47 48 48 48 52 48 19 48 19 48 18 48 28 48 28 48 38 48 br>48 48 48 48 48 48 48 48 48 48 48 48		0 035 5 938 5 933 5 925 5 916 6 909 5 844 5 848 5 848 5 758 5 706 5 707 5 672 6 672 7 641 4 904 5 909 5 901



Case Name: CADOCUMENTS AND SETTINGS MAYARK KANYODIADESKTOPM

TEAM LIND
Cargary Alterna
Carradia

Contectine
Thu Apr 23 02 14 21 2001

## Tray Sizing: Tray Sizing-1 (continued)

9.				
9		Section_1		
10	23_Man TS	0.2794	#202t	501 <u>2</u>
11 12 13	Ç4_Main TS	ে 🖰 ক	• • •	6.013
늯	25 Main TS	620 W		. 41
14	25 Man TS	62.35 B	4	+ .:18
15	27 Man TS	e 2451		5 827
15	28_Man TS	0.2405	62 32	5.048
16 17	29_Man TS	0,2411	52.41	5 293
	30_Mart TS	0.2419	#2 t "	5 175
18	31_Mars 15	0 2428	40 5°	5 306
19	32_Main TS	0.2438	51.12	5.486
20	33_Main TS	0 2449		5 697
21		0.2458	机 1000	5.912
22	34_Mart TS	0.2466	54:4	€ 104
23	35_Mark TS	0.2473	54 30	6 258
24	36 Man TS	<b>!</b>	54.56	6 369
25	37_Mam TS	0 2477	<b>,</b>	5 434
26	38_Man TS	0 2480	54 65	7444

#### DOWNCOMER RESULTS

90 39	Section_1							
1912 1913 1914 1915 1915 1916 1917 1918 1918 1918 1918 1918 1918 1918	EC Backup	DC Backup (fit of liq) (mm)	EC Head Loss (mm)	DC Res Time seles hit	tigmer Res Emilie s5er (mal5)			
1_Main 15	2014	1240	3 139	4.00				
14 2_tApin TS	20.35	124 0	3 127	4760	40% ₹			
15 3_Main TS	20 35	1240	3 128	4 (62				
36 4_1Anin TS	20 35	1240	3 127	4 960	****			
5_Main TS	20 34	124 0	3,124	4 454	****			
8 6_Main TS	20 34	1240	3-129	4 6 6 6				
5 7 Main TS	20.34	1240	3 115	4 4 7 1				
0 8 Main TS	20 33	123.9	3 129	4 * " ! . 1	***			
9_Main 15	20.32	1239	3.102	4975	***			
2 10 Main TS	20 31	123.8	3 093	4 580	141			
13 11_Main TS	20 30	123.8	3 092	1460	•••			
4 12 Main TS	20.29	123.7	3 0 68	4 694	£6.+			
5 13 Main 15	20 28	123.6	3 053	4,792	y vanc y			
6 14 Main TS	20.26	123.5	3 0 3 6	4 "12	****			
7 15_Main 76	20.25	123.4	3016	4 129	Aug It			
8 16 Main TS	20 23	123.3	2 996	4 736	***			
0 17_Main TS	20 21	123.2	2002	<b>3</b> ***.				
0 18_t/ain 75	20 19	123.1	2.947	4 (4				
19 Main 75	24 49	( 621	10.48	1004	***			
2 20_Main TS	24 50	1494	10.52	3 (951	4667			
3 21 Main TS	24 51	1494	10 54	3 050	• • • •			
4 22_tānin 18	24 52	149.4	10.85	3063	~			
8 23_Main TS	24 52	149.5	10.57	3.186	<b>♦</b>			
56 24_Main TS	24 53	1495	10.58	1 ( ž. i.	W			
7 25 Main TS	24 54	145-6	17.0	1, 2,	•			
56 26 Main TS	24 54	149.5	175.04					
9 27 Main 15	24.58	149.5	1769					
\$0 28_Main TS	24 62	15/0 1	10.80	\$656				
31 29_Main TS	2470	150 6	10 97	3 0 2 1	•••			
61 29_Main TS 62 30_Main TS	24 83	1513	11.76	2997				
3 Hyprotech Ltd.			1.3 (Build 4827)		Page 17 of 23			

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" Specified by user



4		TEAM LNC	Case f	Name: CIDOCUMEN	ITS AND SETTINGSWAVAN	IK KANDDIADESKTOP
H	YPROTE	Calipany Alterta	Unit S	ot SI		
-		eater Caradia	Chate/F	imie – Fins Apir 23 (12	14 (1 28)	
1		Tray Sizir	ng: Tray Sizir	ng-1 (continue	ed)	
╁			Sec	tion_1		
-	31_Main 15	25 61	1525	11.69	2 %4	
3	32_Main 7S	15.05	150.9	12.44	, , , , , ,	• •
	33_Main T5	26.62	*## V	10.45		
4	34Main 75	25.76		1) fo	2 144	•
4	35_Main TS	26.62	:53.5 :53.7	14 95	2.767	
+	35_Main TS 37_Main TS	26.20	:60.5	14 79	: 779	••
	38_Main T5	26 39	160 9	14.56	2.765	
+	35_1,4411113	DC Velocity	DC Velocity	DC Design Vel	DO Design Vel	DC Load Factor
1_		(m3/h-m2)	(mirs)	(m3:h-m2)	(m/5)	
	1_Main TS	95 95	2 657e-202	463.0	2 12al	167
1	2_Main TS	95.78	27614-302	<b>46.</b> 16		م پر در سورت
l	3_Mam TS	95 50	2.60%-002	46:0		257 met
-	4_Main T5	96.77	2 6606-663	463.0	0.505/ 0.1287	2 57 2 57
	5_Main TS	95.73	2 65%-002	403.3	0 128	2.57
4	6_Main TS	96 57 96 50	2 656+-002	463.4	0:25	2.57
	8_Idam TS	95 51	2 6576-002	463.6	9 1288	259
1	5_Main TS	55 30	2 6509-602	463 6	0.1288	2 68
	10_Main TS	95 25	0.6464-000	464.0	3.524	2.5%
1	11_Main 75	95 38	2 64% 002	464		2.5%
1	12 Main 15	94.88	2 635 002	464 %	0 t25*	2.59
	13_Main TS	91.61	2,6255-007	<b>364</b> 8	0 1291	2 69
]	14_Mom TS	94 37	2 62 (#-802	465 2	6 1292	2 60
]	15_Main TS	94 97	2 613e-002	465.5	ê 1293	2 61
]	15_Main TS	93.73	2 504e-002	465.9	0 1294	2 61
	12_Main TS	93 37	2 \$94e-G02	466.2	0.1296	7.63
4	18_ Main 78	#2 99 	2 5834-002	496.6	0.1259	263 186
	19_Main TS	175.4	4 8716-002 4 8506-002	467.0 466.9	0 +29+   0 +29+	1 75
	20_Mam TS	175 F	4 8848-992	405.5	4. •	1 *
4	21_Main TS	1760	4 58%- 302	460 %	6.126*	1.79
4	22_Main TS 23_Main TS	176 1	4 8916-502	166.6	0 1297	1.79
4	24_Main TS	1762	4 8940-602	406.8	0.1297	1 75
	25_ Main TS	176 4	4 899002	466 ö	0.1197	1.75
	29_Main TS	1760	4 907±-602	466.6	9.1297	1.73
	27_Main TS	177 1	4 920% 300	405.9	० द∷क्ष	1.73
]	28_Main TS	178 0	4 9446-302	400.9	4.91	· • • • •
	29_Main TS	1,64,1	4 9846-202	368 £	M 1218	1.7
1	30_Main TS	151 8	5 0504-992	466.6	0.12%	179
4	31_Main TS	185.2	f. 1449-002	466.0	0.1/94	17
4	32_Main TS	189.5	5 2646-002	465.2	0 1292 0 1292	16
4	33_Main TS	194.3	5 398e-002 5 827e-002	464 1 463 1	0 1289 0 1284	1 64
4	34IAam TS	1290	5.52% 002 5.63%-002	462.2	0 1284	16
4	35_Main TS 36_Main TS	2010 2081	5 70% 2002	451 5	6 1282	16
4	IT_Main TS	208.3	5 787 6 000	461.1	10,01	1.6
	38 Main TS	200 5	5.815e-762	460.9		1 6
9			FLOODING RESULTS	ACCES TO THE PARTY OF THE PARTY		
23456789012348678901		and the second s				
3	Hiprotoch Ud.		HVEVD	1.3 (Bulld 4827)		Page 18 of
_	Exercised to: TEAM LIV		2.21 20 - 97 1097,	part per plant with the sail		" Spetified by uses

Lizensed to: TEAM LIND



TEAM UND Casgary, Alberta CAMADA

C (DOCUMENTS AND SETTINGS) MAYALIX KANODIA/CESKTOPM Case Name. Umn Set Cale Time The Apr 23 02 14 21 2500

## Tray Sizing: Tray Sizing-1 (continued)

~ I			Section_1		
2 C 1 2 C 1 4 5 C C 7 8 0 0 2 7 2 2 C 2 M M M M M M M M M M M M M M M M		Flooding (%)	Vapour Load (ACT_m3/h)	West Load (m2d) mi	Log ∺thover Wear Jmma
3	1_Man TS	52 **	197.	2574 (	22 <b>5</b> 9
3	2_than 15	12 77	1991		- 19 Ma
5	3_Man TS	52 77	196 :	23 Hz	17 00
6	4_Man TS	62.75	1860	23.47	23.75
7	5_Maso TS	52.73	195.6	22.50	20.70
8	6_Main TS	52.70	155 ~	2:11	23.69
9	"Main TS	52.66	185.5	22/3	23.68
20	8_Main TS	52.61	185-3	79.50	23 %
1	9_Man TS	52 55	185 1	10 43 (	23 64
15	10_Main TS	52.48	184.8	2.4	27.62
3	11_Masri T5	52.49	1842	23.49	22.59
14	12_Main TS	52 35	184 1	23 95	23.56
స	13_Men TS	52 19	183 8	23.38	23 50
36	14_Mam TS	52.06	193.4	20 23	20.47
27	15_Main TS	51.92	182.9	23 15	23,42
28	16_Moin TS	51.76	182.4	29.67	23/36
29	17_Man TS	51 59	1820	22.40	23.30
30	18_Macn TS	51.42	1H1 5	t e e	23/24
31	19_Main TS	59 66	1767	1 19	35.81
32	20_Man TS	59.76	170.9	4.7.24	.5.85
100	21_Main 78	\$9.85	1710	4.5 GB	35-88
1	22_Main TS	59 89	1710	. 43.31	35 89
15	23_Main TS	59 92	171 1	43.34	35.91
15	24_Main TS	59 96	171 1	47.77	35 93
7	25_Main TS	60.01	171 1	4147	35.95
3	26_Masn TS	60 09	171.2	43.44	35.99
~ 	27_Main TS	69 24	171.4	4	୍ର ହେଉ
**	Ze_Main TS	60 52	1718	4.1	35 17
#U	25 Main TS	61.00	1227	44 17	5£ 38
**	30_Macri TS	61 81	1742	44 75	35.70
-	31_Main TS	62 99	1765	45 54	37 18
43	- ;	64 54	1797	46.65	37.77
44	32_Main T6	66 29	183.3	47.83	38 42
45	33Main TS 34klein TS	68.01	186.9	48.19	39 <b>06</b>
40		69 53	1900	49.97	19 60
**	D5_Maon TS	79 72	1925	ec =4	40.02
48	36_Man TS 31_Man TS	71 57	194.3	14.25	45 31
29	- 1	72.05	195.3	51 %	40.46
	38_Mart TS		Entrainment	Entracoment	Weep Volcidy
51 52 53 54 55 55 57 56 59 60 61 62		Flood Capacity	(%)	(kg/h)	(ets's)
53	1_Main TS	0.4255	051	70.07	0 1232
54	2_Man 19	0 4365	Ø5s	77.79	€ 1229
55	1_Mars TS	0.4265	Ø 5 s	TO SH	0.1227
56	4_Main TS	9,4394	5.51	• ; • •	11225
57	5_Main TS	5 <b>4004</b>	1.56		s: 1222
58	6_Man 15	0.4264	350	69.2	2 1222
59	"_Men TS	0 4263	0.50	69 85	÷ 1220
60	e_Mach TS	0.4763	0.50	(29-62	© 1218
61	9 Main TS	9 4262	0.50	69.40	0 1216
62	10_Main TS	0.4262	0.50	59 (4	0.1219
	typrotech Ltd.		YSYS v3.1.3 (Build 4827)		Page 19 of 23



:		Case Name	C (DOCUMENTS AND SETTINGS MAYAN KANODIA DESKTOPM
HYPROTECH	FEAM (NE) Onigate Alteria	Unit Set	Si
1	Cattaga	Crate/Time	Ing Apr 23-02-14-21-3669
6		<u> </u>	

Tray Sizing:	Tray Sizing-1	(continued)

9 10	***************************************	S	ection_1		
10	11 Main TS	0.4260	2.5%	48 SL	. 1219
<del></del>	12_fdan TS	0.4201		4.6 46	. 1212
	12 Main TS	9.4261	949	10 a 2	1216
73	14_Alam TS	0.4262	. Δ.	31.57	1 1215
15	15_Main 75	0.4260	⊕ <b>4</b> 9	6105	U 1009
15	16_Mam TS	0.4259	Q-48	t6.47	© 1209
17	17_Alam TS	0.4259	÷ 46	65.86	0 1009
18	18 Main TS	0.4259	9.46	65-23	6 1210
19	19 Man TS	0.4258	0.29	€2.45	6 1210
30	20_Main TS	0.4258	5/ <b>27</b>	6. 5.	1297
21	21_Main TS	0.4057	4. <b></b>		1264
<del></del>	22_Man TS	3 4057	:. <b>*</b>		1.02
73	22_Man TS	0.4257	2.27	62.03	0.1000
74	24_Main 75	0.4256	3.27	6.410	⊕ 1 <b>8</b> 97
75	25_Main TS	0 4255	6 27	69-19	ŷ 1193
28	26 Main TS	9 4254	a 27	69.36	0 1487
77	27_Main TS	9.4252	027	69.69	Q 1178
28	28_Main TS	0.4249	0.27	To as	0.1160
30	29_Man TS	0.4245	6.27	71.45	@ 1147
30	30 Man TS	\$ <b>42</b> (89		• 4,•	1123
~	31_Mass TS	0.4392	p. 2a - <sup>3</sup>	** %	- 1093
34	32_Main TS	0 4225	929	€0.56	0 1090
32	33_Main TS	0 4218	0.30	85.15	1028
30	34_Marin TS	0.4212	oji	96.60	9.9964-002
35	35_Main TS	0.4207	0 32	94.52	9 7894-002
12 12 13 14 15 16 17 18 19 20 21 22 24 25 28 29 30 31 32 33 33 34 36 36 37 38	36_Man TS	0 4204	9 33	98 20	9 602 <b>a-002</b>
20	37_Man TS	0.4202	033	100 9	9.4%e-002
31	38_Mam TS	0 4201	Q 34	162.6	9479-402

#### LIQUID PROFILE (FROM TRAY)

41		1 Main TS	2_Main TS	± Main T≎	4 Main T\$
42 14055	ilov (kg/h)	1 350+604	1.38%e+004	1 339++(+34	1.3996+004
		3 378e-003	8 3900-003	8 3916-303	8 395m <b>-00)</b>
	ilar Weight	44 15	44 25	44.24	44 45
		44 53	44 89	45.15	45.42
		459 7	459?	459.8	460 0
46 Densit		8 2699-602	9,2776-602	4 (9% o ++.)	5 310 <b>5 002</b>
47 Viscos		4 685	4 67 1	1005	4 060
Harries .	e Tensian (dime/cm)	5_Main 7S	9 Main TS	7 Main 18	8 Main TS
49	The state of the s	1 3859+004	1 3896+004	1 3894 - 004	1.395
50 Mass		8 3859-003	8.380e-003	8 3730-(0)	9 3 <b>65a-003</b>
38 Liquid		44 57	4471	44 97	45 05
	utan VYesqt4	45 72	46.07	46.47	45 93
	erabure (C)	450 2	450.4	400 ?	451.1
54 Densi		CONTRACTOR OF THE PROPERTY OF	8.35Ke-462	8 334n-fei2	8 4164-QQ2
55 Visco:	**************************************	§ 331⊕€02 4 €56	4 652	4 648	4 (344
	e Tension (dene/cm)	<del></del>	<del></del>	11 Main TS	12_Main TS
57		9_Main TS	10_Main TS	1.6676+304	1 .85 004
58 M859	THE RESERVE THE PERSON NAMED IN COLUMN 1	1 3884+034	1 3074+004		5 3109-003
59 Lique	Fray imási	8 355e-003	5 343e-523	8 328-313	46 92
60 Molyc	utar WeigM	45 26	45 49	45.75	
61 Temp	oratino (C)	47 46	48 66	46 -1	49.47
62 Densi	ce (kg/m3)	461.5	4520	462 %	453.0
63 Hypn	dech Lid.	HYS	YS v3.1 3 (Build 4827)		Page 20 of 23

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Speciding by in en



4			Case Name C (DOC) 9/5	ents árið setthyng þavarð.	KANOGAYÇERK <b>TÇEW</b>
١,	HVPROTECH Carpar	utili. Gildeta	Unit Set SI		
Ĭ	CARE CARE	DA.	Cole/Time Thu Apr 23	02 (4 2) 2009	
<u> </u>					
9	Tra	y Sizing: Tray	/ Sizing-1 (continu	ed)	
Ħ	Viscosity (cP)	8 4516-003	8 490e-9001	8 \$1.0 <del>4</del> 192	8 5129 <b>40</b> 0
ō	Surface Tension (dynelan)	4 640	4.637 (		÷ 637
7		15 Mais 75	14, Mars 15	the state of the s	27 Main \$8
2	Mass Flora (kg/h)	1 -840+004	1 (#1.6+ %)4 8 0986-003 (	الانتخاب وهي الانتخاب وهي الانتخاب وهي الانتخاب وهي الانتخاب وهي الانتخاب وهي الانتخاب والانتخاب والتنظيف والت الانتخاب والتنظيف والتنظيف الانتخاب والتنظيف والتنظيف والتنظيف والتنظيف والتنظيف والتنظيف والتنظيف والتنظيف و	₹2100-0003
3	Liqued Flow (m3/8)	5 290e-003 46 32	49.63	46 94	4" 27
4	Molecular WergM  Temperature IC:	50 28	51 16	52.09	53.0."
6	Density (kg/m3)	463 7	4643	4664	465.7
7	Viscosity (cP)	8 0150-002	8 657o-602	୭ ଜନ୍ମିକ ପ୍ରକୃ	8 734a-9 <b>0</b> 0
8	Surface Tension (dyne/cm)	4 532	4 634	4 6 34	1651
9		17 Main 75	18 Main 16	to Main IS	20 Main TS
20	Mass Flox (kg/h)	1 5736-4004	1.57(9+304	<u> </u>	, 1/14+064 + 200 - 200
21	Liquid Flow (m2/s)	8 178e-000	8 145-0-002	1 "566 (65)	1 50 m-602 49 23
22	Malecular Weight	47 58	47.89	48 16 56 11	55 33
23	Temperature (C)	54 08 466 4	55 10 457 h	467.6	467.7
24 25	Density (kg/m3) Viscosity (cP)	8,767a-W2	8.7974-002	8 8229-602	8.8169-602
26	Surface Tension (dynis/cm)	4 647	4 655	4 664	4 650
27		21 Main TS	22_Main TS	23Main 15	24 Main TS
26	Mass Flow (kg/h)	2.6%+604	25046-034	2/14/44/44	् राङ्क्स्स्य
8	Liquid Flox (In3/s)	1 5404-302	1 (414-(0)	194,5 %.	1 54 m/-102
30	Molecular Weight	49.00	49.71	4: 43	48 50
31	Temperature (C)	55.50	56 67	54.94	£7 27
32	Density (kg/m3)	<b>457</b> 6	457 6	467.6	467.7
Ų	Viscosity (cP)	8 5149-0)2	9.615e-002	6.6214-(6)2	8.8329-002
34	Surface Tension (dynidem)	4 642	4 637	4 631	4 605
35		25_Main TS	26 Main TS	27_Main TS	28_ Main TS 2 621e+504
36	Mass Flow (kg/h)	2.6026+904	2 6076+904	2.616e+004 1.111e-1e0	1 (1) (4-002
37	Liquid #York (MR.Vs)	1 545a-002 48 68	1.547e-392 495e	47.84	49 99
3E	Malegular Weight Temperature (C)	57.82	58 66	(4.34	ं। क्ष
39		467.6	468 1	468.4	468 6
40	Density (kg/m3) Viscosity (cf1)	3.8499-002	3.877e-002	89104-007	8167e-002
41		4617	4 605	4 587	4 540
43	·	29_Main TS	30_Main TS	31_Minn TS	32_Main TS
44	1	2.6546∘€04	2 68%004	2.7386+604	2.767 <b>6+004</b>
45		1 572e-C02	1 592 <del>0-0</del> 02	1622+600	1 (4.0002
46	Molecular Weight	50.79	51.76	52.92	45
47	Temperature (C)	64.40	67.75		75.97
48	Donaity (kg/m3)	459 1	469 1	4(3) (6	468 (
40	'Viscosity (cP)	0 02% W2	9 075e-002	9 1116-002	9.11°a-002
50		4519	4.459	4 376 25 Main TS	4 284 36_ Main TS
51	·	33 Main TS	34_Main TS	29734+304	. 013e4 <b>€0</b> 4
52		7 (S(m)+004 1 772e-002	2.921e+004 1.743e-002	1 778+ 300	1.80%=-000
53 54		75 to	55.97	रेटक्र	57.15
55		80 16	\$3.77	Annual Company of the	2.00
50	4	9304	205 €	2:14	\$(,3.)
57		9 0976-607	9 06 1e 000	0.635%-002	ମ ଜ <b>ିଲେ ଓଡ଼ି</b>
56		4 (84	4 (90	4 511	3 05
36	AND AND AND AND AND AND AND AND AND AND	37 Main TS	38 Main 75		
ρſ,	Mass Flow (kg/h)	3 04144-004	3.0560+004		
61	Liquid Flow (m3/4)	1 625002	1 835e-602		
Ŋ.	Malecular Veletyra	57 50	97.75		Page 21 of 23
_			HYSYS v3.1.3 (Build 4827)		



<u> </u>		TEAM (170)	Case Hame - (E/X)	SUBMISHID SELTING WAS	MA KATO DIA CESKIOPM
3	YPROTECH	Casgary Alberta	Unit Set SI		
4	CISECTOLS IN SPATION	CARADA	Date/Time Thu Apr	23 02 14 21 2000	
5					
6 7		Tray Sizing: Tray	/ Sizing-1 (contin	nued)	
8					
4	Temperature (C)	91 19 463 5	<u></u>		
10	Density (kg/m3) Viscosity (cP)	8 935e-002	b this markets and		
12	Surface Tension (dyne/cm)	3 508	588t j		
13		VAP	OUR PROFILE (TO TRAY)		
14		1 Main TS	2 Main 78	3 Main TS	4 Main TS
15 16	Mass Flow (kg/h)	2 244e+004	2.240#+904	2.2474+344	2,24%+004
<del>17</del>	Gas Flow (ACT m3m)	656 G	656.4	JF ≥	455.2
18	Molecular Weight	24 0%	44.15 ,	44.1	14 27
13	Temperature (C)	44 55	ti t		AS 12 AS 28
20	Density (kg/m3)	34 18	427	9 6360-963	9.6436-603
21	Viscosity (cP)	9 625a-003	9 631e-092 1531	15)2	1532
222	Fluid Pressure (kPa)	5 Main TS	6 Mais TS	7 Main TS	8_Main TS
24	Mass Flow (kg/h)	2 247e+004	2 2476-004	2.2474+494	2 2474+004
25	Gas Flow (ACT_m3h)	8545	653.8	652.9	652.0
26	Molecular Weeght	44 34	44 43	44.53	44 64
27	Temperature (C)	46.01	26 47	# (3	47 49 34 49
28	Density (kg/m3)	34 33	34 17	3 669-693	2.6794-000
29	Viscosiny (cP)	9 651e-(9)3	0 659e-603 1534	1505	1530
30	Fluid Pressure INPA	1533 9 Main TS	10 Main TS	11 Main TS	12_Main TS
31 32	Mass Flow (kg/h)	2 2460+004	2 2456+604	2 2450 • 004	2 2436 • 604
33	Gas Flow (ACT_m3/h)	6510	650 6	ध्वतः इ	647.5
34	Molecular Weight	44 77	44 91	45.4%	45.32
35	Temperature (C)	48.05	48.70	4, 4;	59 28 34 64
36	Dansily (kg/m3)	34 50 9 691±-003	.45° } ~ ∴∴440) }	5.4 m² (	9 Na-003
37	Viscosity (cP) Flied Pressure (kPa)	1536	1537	15 38	1534
38 39	Fluid Pressure (kPa)	13 Main TS	14_falson TS	15_Main TS	. 16_Main TS
40	Mass Flow (kg/h)	2 2426+004	2.2494+034	2.237e+004	2.234e+004
41	Gas Flow (ACT_m3/h)	646.2	641.8	(4) 4	+541 <del>5</del>
42	Molecular Weight	45 49	45.56	4: :-	45.95
43	Temperature (C)	51 16	52.09	40.97	54 08 34 81
44	Density (kg/m3)	34 69	34 73 1	9 (3 a · · )	3(0)4-003
49	Viscosity ICPs	9.751e-003 1540	9.756-09.3 E	18-41	1642
46	Fluid Pressure (kPa)	17_Main TS	18 Main TS	19_Main 78	20_Main TS
47 48	Mass Flow (kg/h)	2 2310+004	2 2286+004	2 0096+004	2 1934+904
49	Gas Flow (ACT_m3/h)		638 9	6708	601.9
50	<del></del>	46 13	45 31	55.33	46 42
51	Temperature (C)		Fe) 1:	<i>₩</i> 31	56 50 34 98
52			34 87 1 3 8384 397 1	(4.547 (4.5) (7.5) (4.5)	0 x 40 4 00 3
53	Viscosity (cP)		194.1	1.4	1545
54	The second secon	7	22_Main TS	.3_Main 19	24 Main TS
46		21_Main TS 2 1950+094	2 106e+604	2 1080+(414	2 1100+004
56 57			600 B	930 4	5 <u>69</u> ş
56		46, 47	45 52	48.61	46.74
55			58.63	60.27	61.82
64			35.74	3.11	년 18
91	Viscosity (cP)		5 651e (4))	99500 000	4 H (N. 1940)
63	The state of the s	1540	11.3	1 Arg. 1	Page 22 of 23
6	Hyprotech Ltd.		HYSYS v3.1.3 (Bulls 4827)		Specified to user

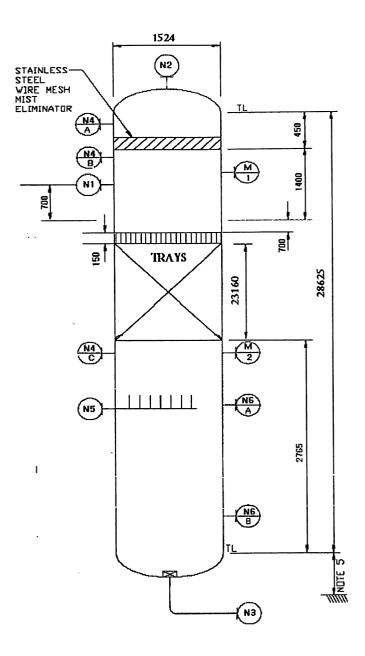
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Γ			Case Name C (C)CCUM	ENTS AND SETT WGSWAYAUK	KAHODIA DESKTOPI
Ļ	YPROTECH	TEAM LITE Caligary, Alberta	Unit Set SI		
	GERRIALE BROYATION	CattaDa	Cole/Time Thu Apr 23	02 14 21 2000	
T		Trav Sizing: Tray	/ Sizing-1 (continu	ıed)	
1			28 Marc 78 1	21 Nov. 7. 4	26_ Main TS
╀		25_Main 75 2.114e+004	2.11(m+0)d		2.14_9+004
4	Mass Flow (kg/li) Gas Flow (ACT_m3/h)	590 )	54-8 5	€ <sub>j</sub> ti ¥	કાર્સ 4
+	Molecular Weight	46.95	47.27	47.7	±9 46
+	Temperature (C)	58 66	59 94	9131	64.40
1	Density (kg/m3)	35.27	35-41	33.65	35.92
	Viscosit, (cP)	0 9804-000	0.0046-003	· )(\d-q-(n))	् विकासी <u>त्रे</u>
Ι	Fluid Pressure (kPa)	1549	1550	(27.0	1662 32 Main FS
		29_Main TS	30_Main TS	31_1Aa6 77	<u>), jobe-004</u>
1	Mass Flow (kg/h)	2 190e+00d	2.201e+004	<u>2.4 m • 354</u> €00.00	600 7
4	Gas Flow (ACT_m3h)	595 6	5067 5060	5194	50 30
4	Molecular Weight	49 41 67 79	73.73	76.47	80 10
4	Temperature (C) Detruits (kd/h3)	36.36	36.95	الداء " و	58.4
+		1 00026-002	1.0036-602	1.0104.492	1 015 <del>6-00</del> .
<u>:</u>	Viscosity (cP) Flood Pressure (kPa)	1553	1564	1614	1565
5	1.0924 (1023187 (55.0)	33 Main TS	34_Main TS	C. Main 78	${\mathcal N}_{\mathbb Z}$ Main ${\sf TS}$
5	Mass Flow (kg/h)	2 3720+004	2.4336+304	, 435m+3414	. ૧૫૧૦ છે.
7	Gas Flow (ACI_m3h)	600.6	6112		<sub>6</sub> 203
5	Molecular Weight	54.95	53.56	÷, 30	67.08
9	Temperature (C)	63.77	66.83	55 27	91 11
ō	Density (kg/m3)	35 17	3981	40.31	40 6
7	Viscosity (cP)	1 0156-002	1 021e-003	1 02%+ 000	1 025a-00
2	Fluid Pressure (kPa)	1556	1557	1055	155
3		37_Main TS	39_Main TS		
4	Mass Flow (kg/h)	2.55%+004	2.5684-004		
5	Gas Flow ACT_multip		516.4		
6	Molecular Weight	57.42	57.72		
7	Temperature (C)		64 15 40 99		
18	Density (kg/m3)		1 0276-002		
9	Viscosity (CP)		1560		
0	Fluid Pressura (kPa)	1934	1.752		
1 2 3 4 4 15 16 17 16 55 55 55 55 55 55 55 55 55 55 55 55 55					
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62				والمترابع والمتراجع والمتراجع والمتراجع والمتراجع والمتراجع والمتراجع والمتراجع والمتراجع والمتراجع	Page 23 of



### 3.4 LAYOUT OF THE DESIGNED COLUMN



Note: All the dimensions are in mm.



DETAIL "B"
COLUMN BOTTOM

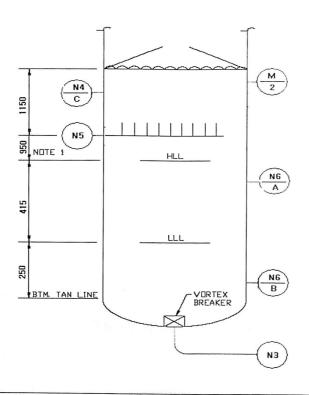
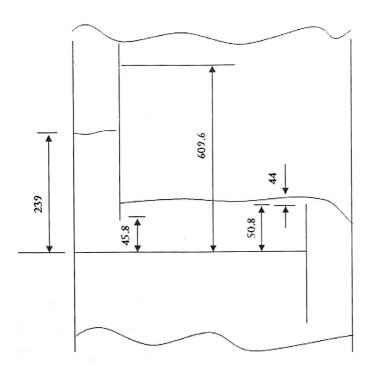


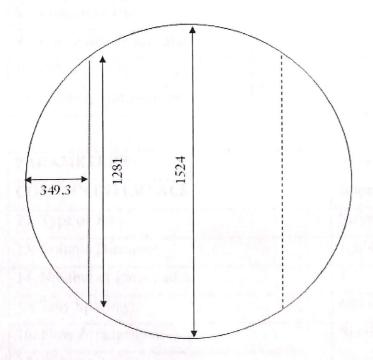
Table 2: Nozzles and Connections Details

Mark	No.	Size (in) / Rating	Service
M1	1	24 / #300	Manhole
M2	1	24 / #300	Manhole
N1	1	4 / #300	Feed
N2	1	8 / #300	Overhead
N3	1	10 / #300	To reboiler
N4A/N4B/N4C	3	2 / #300	DP guage
N5	1	12 / #300	Reboiler outlet
N6A/N6B	2	3 / #300	Level Transmitter





TRAY (FRONT VIEW)



TRAY (TOP VIEW)



## **RESULTS**

Important Results are summed up as under:

Table 3: Important Results of Column Design

DADAMETERS	THEORETICAL	VALUE FROM ASPEN
PARAMETERS	VALUE	HYSYS
COLUMN PARAMETERS		
1. Minimum Number of Stages	13	15
2. Minimum Reflux	1.53	1.53
3. Number of Theoretical Stages	26	27
4. Actual Number of Stages	38	38
5. Stripping Feed Tray Location	19	19
6. Rectifying Feed Tray Location	19	19
7. Reflux Ratio	2.2935	2.293
8. Condenser Duty	-	8.562 * 10 <sup>6</sup> KJ / hr
9. Condenser Temperature	-	42.53 °C
10. Reboiler Duty	- MANA	8.359 * 10 <sup>6</sup> KJ / hr
11. Reboiler Temperature	- Hay rown	94.32 °C

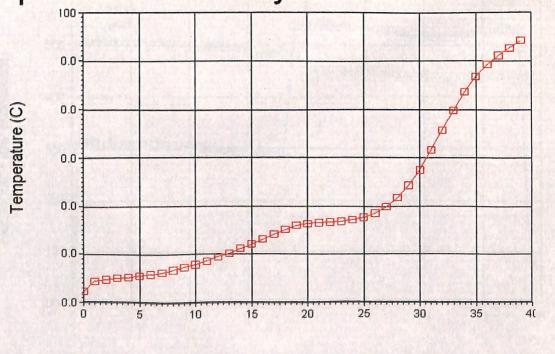
same as the Values taken from Aspen Hysys	
Sieve Tray	
1.524 m	
1	
609.6 mm	
Single Pass Cross Flow	
1.824 m <sup>2</sup>	
$0.3153 \text{ m}^2$	
1.5087 m <sup>2</sup>	
1.194 m <sup>2</sup>	
$0.1678 \text{ m}^2$	



22. Weir Type	STRAIGHT
23. Weir Height	50.8 mm
24. Weir Length	1281 mm
25. Hole Diameter	5 mm
26. Maximum Weir Loading	$89.42 \text{ m}^3 / \text{h} - \text{m}$
27. Hole Pitch	12.7 mm
28. Plate Thickness	5 mm
29. Total Plate Pressure Drop	580.8776 Pa
30. Downcomer Type	Vertical
31. Downcomer Clearance	45.8 mm
32. Downcomer Clearance Area	0.0587 m <sup>2</sup>
33. Downcomer Backup	239 mm
34. Downcomer Residence Time	4.1068 sec
35. Number Of Holes	8545

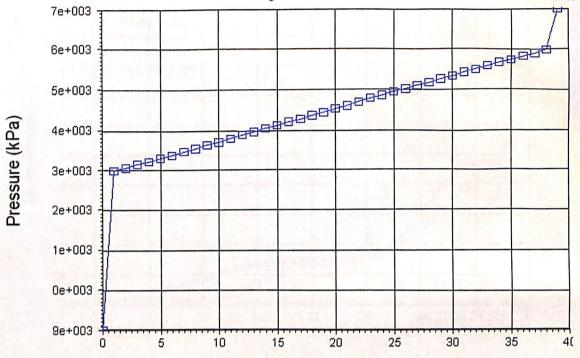
### **GRAPHS**

Temperature vs. Tray Position from Top

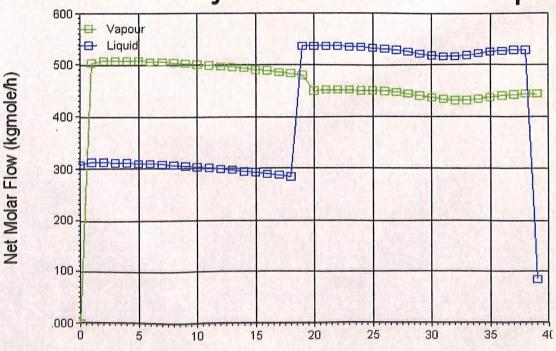




# Pressure vs. Tray Position from Top

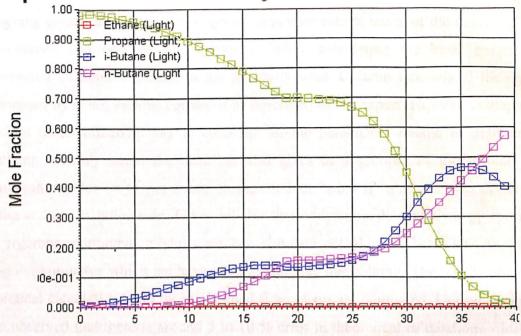


# Flow vs. Tray Position from Top





Composition vs. Tray Position from Top





#### **CONCLUSION**

Designing of distillation column plays an important role in terms of the degree of separation to be achieved and cost of the operation. While calculating the basic parameters, Fenske Underwood Gilliland correlations are generally used. Column internals of the column can be determined by using various equations as described in this report. However, computer simulator provides us an effective way to calculate all the parameters related to designing. Use of a simulator not only saves the time but also gives us a glimpse of the column run. Process simulation enables us to run many cases, conduct "what if" analysis, and perform sensitivity studies and optimization runs. Given reliable thermodynamic data, realistic operating conditions, and rigorous equipment models, we can simulate actual plant behavior. Aspen Hysys is a computer simulator which we have used to simulate our column. The results obtained from the theoretical calculations and those obtained from hysys are compared. During comparison, it has been observed that there is around 5 to 10 % error in theoretical calculations while keeping the hysys results as reference. Thus, designing of depropanizer column is done and simulation results are compared.



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