Hence provide outlet arrangement consisting of effluent launder with weirs on both sides of launder.

OUTLET ARRANGEMENT:

a pressure outlet pipe. The outlet arrangement consists of effluent weir of V-notches, effluent launder, effluent box and

i) Effluent Weir:

Length of effluent weir plate on each side of launder

=
$$\pi \times (57 - 1) = 175.93$$
 say 176 m

Provide 90° V-notches **(9**) 20 cm centre to centre on both sides of the launder.

Total No. of notches =
$$176 \times 5 = 880$$

Average discharge per notch at average design flow

$$= \frac{50 \times 10^6}{24 \times 60 \times 60 \times 880 \times 1000} = 6.58 \times 10^{-4} \text{ cum} \text{ sec.}$$

The discharge through a V-notch is given by

$$Q = \frac{8}{15} C_d \sqrt{2g} \tan \frac{\theta}{2} H^2$$

for peak flow per notch, $Q = 6.58 \times 10^{-4}$ x 2.25 11 1.48 x 10⁻³ cum/s

or
$$C_d = 0.584, \Theta = 90$$

Head over V-notch at peak flow =

$$\left(\frac{15x1.48x10^{-3}}{8x0.584x\sqrt{2x9.81}}\right)^{\frac{2}{5}} = 0.065m$$

Provide 8 cm deep 90 degree V-notches at 20 cm centre to centre

(ii) Effluent launder:

to critical depth of flow. effluent launder, assume that the effluent launder discharges freely into the effluent Assume the width of effluent launder or channel to be 0.6 m. Consequently the depth at the end of effluent channel may be assumed equal tical depth of flow. Critical depth at the end of effluent launder, Y_2 is To compute depth of

$$Y_2 = (\frac{(g'xL)^2}{(b^2xg)})^{\frac{1}{3}}$$

$$Y_2 = \left[\frac{(\frac{50 \times 10^3}{2 \times 24 \times 3600})^2}{(0.6^2 \times 9.81)} \right]^{\frac{1}{3}} = 0.287 m$$

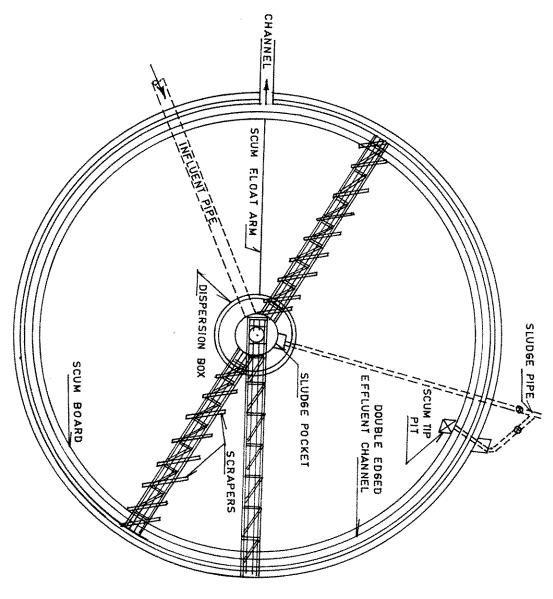
Depth of water at upper end of the trough, Y,

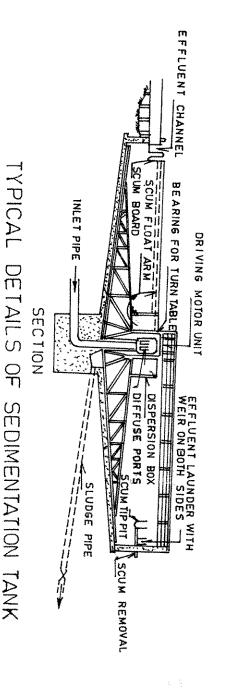
$$Y_1 = \sqrt{Y_2^2 + \frac{2x(q'xLxN)^2}{gxb^2xY_2}}$$

$$Y_1 = \sqrt{0.287^2 + \frac{2(\frac{50x1000}{2x24x3600}x2)^2}{9.81x0.6^2x0.287}} = 0.862 m$$

Provide a depth of 0.95 m.

APPENDIX_ 12.1





DETAILS

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SEDIMENTATION TANK

APPENDIX 13.1

DESIGN OF CONVENTIONAL ACTIVATED SLUDGE PROCESS

Given:

Flow = $50,000 \, \text{m}^3/\text{d}$, Raw wastewater BOD_s = $250 \, \text{mg/l}$; SS = $400 \, \text{mg/l}$; Minimum and maximum temperature = $18 \, \text{and} \, 32^{\circ} \, \text{C}$ respectively; Primary Sedimentation efficiency for BOD and SS removal = 35% and 75% respectively; Primary and Secondary excess sludge SS concentration = $40 \, \text{and} \, 10$ kg/m³, Aeration equipment oxygen transfer efficiency under standard conditions = 1.8 kg O₂/kWh.

Aeration tank volume:

- B0D of influent to aeration tank = 250x65/100 = 162.5 mg/l
- For 90% BOD removal read Θ_c for 18° C from Fig.(13.3) = 6.5 days
- Adopt for conventional activated sludge MLSS = 2000 mg/l
- Assuming Y = 0.5 and $k_d = 0.06/d$ from Eq.(13.7) calculate V = 8549 m³
- if a larger HRT value is desired, repeat calculations assuming lower value of MLSS HRT from Eq.(13.1) = 4.1 h, which is greater than 4 h, hence acceptable.
- requirements and conditions detailed in section 13.4.1. The dimensions of the tank will be decided on the basis of aeration equipment

Excess Sludge:

- Calculate $Q_{*}X_{s}$ from Eq.(13.3) = 2630461.5 g SS/d or 2630 kg/d
- For 10 kg/m³ SS concentration in secondary sludge, excess sludge volume = 263 m³/d

Sludge Recirculation:

- 0.5, hence acceptable. Calculate studge recirculation ratio from Eq.(13.10) = 0.25, which is between 0.25 and However, provide for 0.33
- Therefore sludge recirculation pump capacity $= 0.33 \times 50,000 = 16,500 \text{ m}^3/\text{d}.$

Oxygen Requirement:

- Calculate_oxygen requirement from Eq.(13.8) assuming f = 0.68; = 7018420 g/d
- acceptable Calculate kg O_z required/kg BOD removed = 0.96 which is between 0.8 and 1.0, hence

Aerator Power Requirement:

- and $\alpha = 0.8$ calculate, oxygen transfer capacity of available aeration equipment from For field conditions; temperature = 32 degree C assuming C_L $Eq.(13.9) = 1.3 \text{ kg } O_2/kW.h.$ 1 mg/l, Cູ
- Therefore aeration equipment power requirement = $7018 / 24 \text{ kg O}_2 / \text{h} / 1.3 \text{ kg O}_2 / \text{kW.h} = 225 \text{ kW}.$

Sludge Generated:

- Primary sludge solids = $50,000 \text{ m}^3/\text{ d} \times 400 \text{ g/m}^3 \times 0.75 \times 1 \text{ kg} / 1000g$ = 15000 kg/d

- Primary sludge volume = $15000 \text{ kg/d} / 40 \text{ kg/m}^3 = 375 \text{ m}^3/\text{d}$
- Secondary sludge solids (from earlier calculations) = 2630 kg/d.
- Secondary sludge volume = 263 m³/d.
- Total sludge volume = $375 + 263 = 638 \text{ m}^3/d$.

APPENDIX 13.2

DESIGN OF FACULTATIVE AERATED LAGOON

Average ambient air temperature in January is 18 deg. C and in summer 37 deg.C. cu.m./day, Raw BOD₅ = Design a facultative aerated lagoon to serve 40,000 people. Sewage flow @ 180 lpcd = 7200 ay, Raw $BOD_s = 50$ gcd or 277 mg/l and final BOD_s is not to exceed 30 mg/l in winter.

Lagoon Size

Assume detention time days

Lagoon volume $7200 \times 5 = 36,000 \text{ cu.m.}$

Let Lagoon dimensions be 70 m x 130 m x 4 m deep

Lagoon Winter Temperature

Use Eq.(13.3) to determine $T_{\rm c}$. Assume $T_{\rm c}=23^{\circ}$ C

Hence,

Estimation 아 ㅈ

Assume K at 20° C 11 0.7 per day

D/UL Estimation

Hence, K at 21° C

11

 $0.7 \times 1.035 = 0.724 / day$

Keep lagoon geometry such that flow conditions are plug-flow type (i.e. D/UL = 0.2 approx.). This will be possible if a long and narrow lagoon ($23m \times 390 \text{ m}$) is provided (see Table 13.3) or baffles are provided within the rectangular lagoon of $70m \times 130m$ to give a winding flow with the same effect. (See Fig. 13.5).

BOD_s Removal Efficiency (in Winter)

X O S.S.likely to flow out in effluent Namely, soluble BOD in effluent Soluble BOD removal efficiency See Fig.(13.4) at $K \times \Theta = 3.62$ and D/UL Overall efficiency in winter Hence, BOD of effluent = 22 BOD of VSS = $0.77 (0.6 \times 35)$ $= 0.724 \times 5$ + 5 Ħ H Ħ II 11 16 mg/l 22 mg/l 92% 3.62 86% 38 mg/l 0.2 35 mg/l (say)

above value. In other months of the year, the efficiency will be higher and effluent BOD will be less than the

Power Requirement

Area including embankments	Net lagoon area	Land Requirement	11	Power level in Lagoon	H		Power needed	н	O ₂ required/day ==	When efficiency =
S	9000 sq.m.		1.7 W/cu.m (acceptable)	62.5 KW x 1000 36,000	62.5 KW (i.e. about 80 HP)	(0.8) (2 kg O ₂ /KWh)	100 kg/hr	2,408 kg/d = 100 kg/hr.	0.86 (1.4 x 2000 kg/d).	86% and all BOD is removed aerobically,

NOTE: If the lagoon was kept as a square shaped unit or a rectangular unit with say W:L = 1:2, the D/UL value would have been between 3.0 and 4.0 (namely, approaching completely - mixed conditions) and soluble effluent BOD would have increased to 49 mg/l, thus giving a total final effluent of about 65 mg/l instead of 38 mg/l seen above. Thus, lagoon geometry plays an important part in determining efficiency.

and slopes

Area/person

II

0.337 sqm/person

13,500 sq.m. (approx)

APPENDIX 14.1

DESIGN OF TRICKLING FILTER

Problem Statement:

Design a high rate trickling filter plant to treat settled domestic sewage with a BOD $_s$ of 200 mg/l for an average flow of 50 MLD. Assume a peak factor of 2.25. The desired BOD $_s$ of effluent is 10 mg/l.

Solution:

Several design approaches are available for the design of trickling filters. Two approaches will be used to design the trickling filter viz. (i) NRC equation and (ii) Rankine's approach.

Since the BOD_s removal efficiency is high a two stage filtration system has to be used. The design of filters is done on the basis of average flow. However, the hydraulic design of the distribution arms, under drainage system, pipelines etc., is done for peak flow and checked for average flow.

i) Design Using NRC Equation

first stage filter, Assuming a BOD loading of 0.8 kg BOD, applied/m³/d excluding recirculation, the volume of

Volume =
$$\frac{BOD_{5} load}{BOD_{5} loading} = \frac{50 \times 200}{0.8}$$
= 12,500 m⁶

The efficiency of first stage filter using NRC equation,

$$E_{1} = \frac{100}{1 + 0.44 \sqrt{\frac{W_{1}}{V_{1}F_{1}}}}$$

Adopting a recirculation ratio of 2.

$$F_1 = \frac{1 + R_1}{(1 + 0.1R_1)^2} = \frac{1 + 2}{(1 + 0.1x^2)^2} = 2.0833$$

 $W_1 = 50 \times 200 = 10,000 \text{ Kg BOD}_g/d$

$$E_1 = \frac{100}{1 + 0.44 \sqrt{\frac{10,000}{12,5000x2.0083}}} = 78.6\%$$

The efficiency of second stage filter. $\mathsf{E}_{\mathbb{R}}$

$$E_2 = \frac{200-10}{200} \times 100-78.6 = 16.4\%$$

The volume of second stage filter can be computed using the equation

$$E_{2} = \frac{100}{1 + 0.44} \frac{W_{1}(1 - E_{1})}{V_{2}F_{2}}$$

Adopting a recirculation ratio of one, the value of F_z is

$$F_2 = \frac{1 + F_2}{(1 + 0.1 R_2)^2}$$

$$\frac{1+1}{(1+0.1x^2)^2} = 1.653$$

$$V_2 = 274.8 \text{ m}^3$$

ii) Rankine's Approach

Adopting an organic loading of 0.8 Kg BOD/m 3 /d as assumed in earlier case, the volume of first stage filter is 12,500 m 3

Adopting a filter depth of 1.5 m.

Filter area needed

$$=\frac{12,500}{1.5}$$
 =8333 m^2

using a circular filter,

$$dia = \sqrt{\frac{8333 \times 4}{\pi}} = 10299 m$$

three units. Since rotary distributors are available indigenously only upto 60 m, it is desirable to have a least

$$= \frac{833344}{3\pi} = 59.48m$$

Say 60 m

Applying Rankine's formula for the first stage filter and varying value of $R_1 = 0.5$, 1.5, 2.0, 2.5 and 3.0 efficiency of first stage filter can be calculated by Rankine's equation. 0.5, 0.75, 1.0,

giving values of 75; 77.78; 80; 83.33; 85.77; 87.50 and 88.88 % respectively.

These values are entered in column 2 & 3 of Table 1 respectively.

Similarly the efficiency of second stage Filter

$$\mathbf{E}_2 = \frac{1 + \mathbf{R}_2}{2 + \mathbf{R}_2}$$

Various values of $\rm R_2$ and efficiencies are entered in columns 5 and 6 of Table 1. Column 4 gives the BOD₅ passing through the first stage filter.

Now, the combined efficiency of the filters required to give an effluent BOD₅ of 10 mg/l.

Efficiency of two stage $Ec = E_1 + E_2 (1-E_1)$

For a R, value of 0.5 this will be

$$0.95 = 0.75 + E_2$$
 (1-0.75) or $E_2 = 0.8$

 R_2 value from col.5 of Table = 3.0

Similarly $\rm R_2$ values for various $\rm E_2$ values for different $\rm R_1$ values to obtain 95% efficiency are given in col.7 of Table 1.

TABLE 1 $\mathrm{R_2}$ VALUES FOR DIFFERENT VALUES OF $\mathrm{E_2}$ AND $\mathrm{R_1}$ TO OBTAIN 95% EFFICIENCY

7	ø	₹ 7	4	. (ω	2			S.No.
3,00	2.50	2.00	1.50	1.00	0.75	0.50	Recirculation Ratio of 1st stage filter	ŢĮ.
88.88	87.50	85.77	83.33	80.00	77.78	75.00	Efficienty of lst stage filter.	Ε,
22.22	25.00	28,66	33.33	40.00	44,44	50,00	BOD _s passing through lst stage filter.	S
3,00	2.50	2,00	1.50	1,00	0.75	0.50	Recirculation Ratio of 2nd stage filter.	R ₂
80.00	77.78	75.00	71.43	66.67	63.64	60.00	Efficiency of 2nd stage filter	m
And the state of t	0.50	1.00	1.50	2.00	2.50	3.00	Values for various R, values to give 95% Efficiency.	

The hydraulic loadings for different values of R, in terms of Kld/m2 for the average flow.

	Choose						0	O	= 5.89226 × (1 + H ₁)	$50 \times 10^{\circ}$ 4 \times 3×10^{3} $\pi \times (60)^{2}$
J.	Ţ	3.00	2.50	2.00	1.50	1.00	0.75	0.50	, p	дж(60) ²
AM	H									
1 for Second Stage Filter.	2 for First Stage Filter and	23.57	20.62	17.68	14.73	11.78	10.31	8.84	Hydraulic Loading (m³/d/m²)	x (1+R,) is worked out.

Organic loading (Recirculation included) for 3 filters of dia. 60m and depth 1.5 m

This is less than 1800 g/d/m³ and therefore the equations are applicable.

Choosing an organic loading 0.5 Kg/d/m³

Adopting a depth of 1.0 m

Area of filter = 2866 m²

Check for hydraulic loading

$$\frac{50.00 \times 10^{3} \times 1 \times (1 \times 1)}{2866} = 34.89 \frac{k/d}{m^{2}}$$

Which is more than permissible.

Therefore area required for maximum permissible hydraulic loading of 30 Kld/m²

$$50 \times 10^3 \times [(1+1)/(30)] = 3333.33 \text{ m}^2$$

Adopting 3 circular Filters,

$$dia = \sqrt{\frac{333333 \times 4}{3 \times 1}} = 37.6 m = 38 m$$

Adopting 3 units of 38 m dia and 1.0 m depth for 2nd Stage Filter.

iii) Hydraulic Design of First Stage Filter

This is designed for the peak flow + the recirculation of the average flow at the rates prescribed. In this case the recirculation is 2 times the average flow.

Total flow through the filters at the peak flow with 2.25 peak factor

 $= 50 \times 2.25 + 2 \times 50 = 212.5$ Mid or 2.459 m³/s

This flow is divided into 3 units

Therefore flow through each unit at peak flow = 0.82 m³/s

Adopting a velocity of 2 m/s, dia of central column

$$\sqrt{\frac{0.82\times4}{\pi\times2}}=0.722m$$

provide a central column = 0.75 m

check for velocity at average flow:

Ave. Flow =
$$50 \times 10^6 \times (1+2) = 150 \text{ MId} = 1.736 \text{ m}^3/\text{s}$$

Therefore velocity at average flow =

$$\frac{1.736}{3} \frac{4}{\pi \times (0.75)^2} = 1.31 \text{ m/s}(> 1 \text{ m/s})$$

Distributor:

Assuming rotary reaction spray type distributor with 4 arms:

Dia of filter = 60 m

Arm length = [(60 - 2) / 2] = 29m with 4 sections of 7.25m each

of the areas covered by these lengths of the arm. Therefore, the areas covered by the different lengths of the arms are calculated. The flow in the arms has to be adjusted for every section of 7.25 m length in the proportion

Let A_0 , A_2 , A_3 and A_3 be the areas covered by each length of arm starting from the centre Allowing for 0.75 m dia in centre to be used up for central column etc.,the areas are

A₁ =
$$\pi (7.625^2 \cdot 0.375^2) = 182.29 \text{ m}^2$$

A₂ = $\pi (14.875^2 \cdot 7.625^2) = 512.68 \text{ m}^2$
A₃ = $\pi (22.125^2 \cdot 14.875^2) = 843.07 \text{ m}^2$
A₄ = $\pi (29.375^2 \cdot 22.125^2) = 1173.46\text{m}^2$

The proportionate area for each length of arm 1st i.e. from column to 7.625 m.

$$A_1$$
 182.29 = 6.72 % $A_1 + A_2 + A_3 + A_4$ 2542.17 Similarly 2nd 18.91% 3rd 31.09% 4th 43.28%

Orifices:

1.5 m Assuming a dia of 25mm for the orifices with Cd value of 0.6 and head causing flow equal to

dischargethrougheachorifice = CdxA√2gh

$$=0.6x\frac{\pi}{4}x0.025^2x/2x9.81x1.5$$

0.001597 m³/s

Therefore No.of Orifices required in each arm

Total discharge through arm

Discharge through each orifice

= 128.36 say 129 0.001597

No. of orifices in each section of the arm is

Ist section (6.72 / 100) x 129 = 9

2nd section (18.91 / 100) x 129 = 25

3rd section (31.09 / 100) x 129 = 40

4th section (43.28 / 100) x 129 = 56

Spacing of Orifices:

3rd Section	2nd Section	1st Section
40 Nos.	25 Nos.	9 Nos.
40 Nos. in 725 cm i.e. 725/40	2nd Section 25 Nos. in 725 cm i.e. 725/25	1st Section 9 Nos. in 725 cm i.e. 725/9
Ħ	ti	11
18cm c/c	29cm c/c	80cm c/c

4th Section 56 Nos. in 725 cm i.e. 725/56

13cm c/c

Diameters of different sections of the arm:

The flow through velocity in the arm should be less than 1.2 mps

a) Discharge through 1st section = 0.205 m³/s

Crosssectional area with 1.2 mps = $(0.205 / 1.2) = 0.1708 \text{ m}^2$

Assuming circular section, dia of pipe

$$\frac{0.1708x^4}{\pi} = 0.466m \, say470mm$$

b) Discharge through 2nd section

$$= (1-0.0672) \times 0.205 = 0.1912 \text{ m}^3/\text{s}$$

For V = 1.2m/s

$$dia = \sqrt{\frac{0.1912x^4}{1.2x\pi}} = 0.45m \, say 450mm$$

c) Discharge through 3rd section

$$[1 - (0.0672 + 0.1891)] \times 0.205 = 0.1525 \text{ m}^3/\text{s}$$

For V = 1.2m/s

$$0.1525x4 = 0.402m say400mm$$

d) Discharge through 4th section

$$= 0.4328 \times 0.205 = 0.0887 \text{ m}^3/\text{s}$$

For $V = 1.2 \,\text{m/s}$

$$dia = \sqrt{\frac{0.0887x^4}{1.2x\pi}} = 0.3067m \, say310mm$$

Under Drainage System:

Total discharge through each filter at peak flow = 0.82m³/s.

laterals placed at $0.6~{\rm m~c/c}$ with a slope of 2.5% in each half circle. The invertheir junction with the peripheral main collecting channel is kept the same R.L. The underdrainage system is designed with a peripheral collecting channel fed by semi circular placed at 0.6 m c/c with a slope of 2.5% in each half circle. The invert level of all laterals at

Average discharge per lateral:

$$0.82$$
 = 0.0041 m³/s 100×2

nq
$$0.015 \times 0.0041$$

$$S^{1/2} = 0.025^{1/2} = 0.000389$$

The laterals are designed to flow half full to provide for proper ventilation.

... @

$$= 0.25$$
; q $= \frac{q}{d_o^2} = x \cdot 0.25 = 0.1963$

From Appendix 26

for
$$(a/d_o^2)$$
 of 0.1963; $(ar^{2/3}/d_o^{8/3}) = 0.05915$

adopting 16 cm dia

$$(ar^{2/3}/d_o^{8/3}) = (0.000389/0.16^{6/3}) = 0.051$$

From Appendix 26

corresponding
$$(a/d_o^2) = 0.1753$$

Check for Velocity at Average Flow:

Total discharge =
$$50+2\times50 = 150 \text{ M/d} = 1.736 \text{ m}^3$$

$$=$$
 0.579 m³/s

Average flow per lateral = $[0.579 / (100 \times 2)] = 0.00290 \text{ m}^3/\text{s}$

$$ar^{2/3} = 0.015 \times 0.0029$$

 $0.025^{1/2} = 0.000275$

and
$$ar^{2/3}$$
 0.000275 = 0.0365 $d_o^{8/3}$ (0.16)^{8/3}

corresponding
$$(a/d_o^2) = 0.1379$$

... Velocity =
$$[0.0029 / (0.1379 \times 0.16^{\circ})]$$
 = 0.8215 m/s (>0.6 m/s required)

inlet openings for the flow into the laterals to ensure proper ventilation. The laterals are covered with perforated blocks capable of withstanding the load of the filter media. It should be ensured that there is at least 15% of the total filter area available in the form of

that inlet area available is about 15% of the filter area. 20% of the filter area. Therefore it is to be provided with cover blocks having about 75% openings so In the present design the total surface area of the laterals at the floor level of the filter is about

DESIGN OF MAIN COLLECTION CHANNEL

checked to see if free fall conditions exist while flow from the laterals of each segment falls into it. is laid to a constant slope of 0.5%. The filter can be divided into four segments and the main channel is divided into two and the flow from each semi circle is collected in the peripheral main channel which It is desirable to provide the main collection channel along the periphery of the filter. The flow

depth of semicircular section. To provide a free fall from the invert of the laterals assume the depth of flow to be 5% less than

1st Segment:

$$q = 0.1 \times 0.82 = 0.082 \text{ m}^3/\text{s}$$

from Appendix 26

$$(y/d_o) = 0.475;$$

ੂੰ

$$(ar^{2/3}/d_o^{8/3}) = 0.1426$$

Qο

$$(a/d_s^2) = 0.367$$

for a slope of 0.5% and n = 0.015

$$ar^{2/3} = \frac{nq}{S^{1/2}} = \frac{0.015 \times 0.082}{0.005^{1/2}}$$

$$d_o = [(0.01739) / (0.1426)]^{3/8} = 0.4543 \text{ m}$$

Adopting 46 cm or 0.46 m dia & 0.5% slope

And for this $(y / d_0) = 0.4658$ and (a / d_0^2) = 0.3585

$$0.082$$
 Velocity = 0.3585×0.46^2 = 1.08 m/s (> 0.75 m/s required)

2nd Segment:

$$q = 0.25 \times 0.82 = 0.205 \text{ m}^3/\text{s}$$

vertical depression at the end of the 2nd section

=
$$(\pi D / 4) \times (0.5 / 100)$$
 = $[\pi \times (60 / 4) \times (0.5 / 100)]$ = 0.24 m

Total additional flow in this section

$$0.15 \times 0.82 = 0.123 \text{ m}^3/\text{s}$$

Flow that can be accomodated

$$0.24 \times 0.46 \times 1 = 0.1104 \text{ m}^3/\text{s}$$
 (Assuming 1 m/s velocity)

Hence choose a bigger section say 53 cm.

Redesign of 1st Segment:

$$ar^{2/3} = 0.01739$$

 $ar^{2/3} = 0.01739$

For this value,
$$(a / d_o^2) = 0.27$$
 and $(y / d_o) = 0.3778$

d 8/3

(0.53)^{8/3}

= 0.09453

0.01739

Check for Average Flow (Recirculation included)

Flow in Segment 1,

$$= (1.736 / 3) \times 0.1 = 0.0579 \text{ m}^3/\text{s}$$

$$ar^{2/3}$$
 0.0579 x 0.015 1 x = 0.06676 $do^{6/3}$ 0.005^{1/2} (0.53)^{8/3}

for this $(a/d_o^2) = 0.2113$ and $(y/d_o) = 0.314$

2nd Segment:

$$q = 0.205 \text{ m}^3/\text{s}$$

$$0.205 \times 0.015$$
 $ar^{2/3} = 0.04349$
 $0.005^{1/7}$

$$ar^{2/3}$$
 0.04349 = 0.2364 $do^{8/3}$ (0.53)^{8/3}

For this value, $(a / d_o^2) = 0.541$ and $(y / d_o) = 0.65$

Depth of Flow = $0.65 \times 0.53 = 0.3445$ say 0.35m

Depth from invert of channel to invert of lateral =

Clearance = 0.5 - 0.35 = 0.15 m ensuring free flow conditions

3rd Segment:

$$q = 0.4 \times 0.82 = 0.328 \text{ m}^3/\text{s}$$

Assuming depth of flow above semi circular section to be x

$$\int_{\rho}^{\pi} \frac{\pi R^2}{\rho} + 0.5x$$

$$\int_{\rho}^{\pi} \frac{dR^2}{\pi R^2} + 0.5x$$