WATER TREATMENT EAP215/3 WATER SUPPLY AND TREATMENT

ENGINEERING

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References:

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- 4.Water Supply by Twort A.C.,Law F.M. and Crowley F.W., Edward Arnold.
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Typical Water Treatment Process



Why water needs to be treated

- Palatable and pleasant to drink
- No colour and smell
- Free of pathogenic organisms
- Not corrosive affecting humans as well as equipment

Classes of Treatment

Class	Treatment	Water source
A	No treatment	Spring water found in hilly areas and used individually
В	Disinfection	Spring water found in hilly areas and used by the public
С	Normal treatment process	River and impounded reservoir (upstream and downstream of the river)
D	Special treatment	Need to remove pollutants such as colour, algae, organic and inorganic matters from the river or lake water . Water for special industrial use.

Propose treatment process

Parameters	Treatment process
Suspended solids	Course screen and fine screen
Dissolved solids	Micro-screen
Algae	Micro-screening, pre-chlorination, copper sulphate, carbon adsorption, rapid filter
Turbidity	Coagulation, flocculation, sedimentation, or dissolved air flotation
Colour	Coagulation, flocculation, filtration, carbon adsorption
Taste and smell	carbon adsorption, chlorination
Hardness	Coagulation, filtration, lime softening

Parameters	Treatment process
Iron and manganese : >1 mg/L <1 mg/L	Pre-chlorination Aeration, coagulation, filtration
Pathogen (MPN/100mL) <20 20-100 >100	Post chlorination Coagulation, filtration, post chlorination Pre-chlorination, coagulation, filtration, post chlorination
Free ammonia	Adsorption, post chlorination

Pre-treatment

- Raw water storage
- Screen
- Pre-chemical treatment
- Aeration
- Pre-sedimentation







Raw Water Storage

- To even out variations in flow
- To improve the water quality
- To even out the variations in water quality
- When water is stored in a reservoir for a period of 1 month to several months – there is a great decrease in pathogenic organisms. Even a few day will improve its physical and microbiological characteristics.

- Storage in the form of impounded reservoir or a small open storage reservoir.
- Improvement is a result from natural settling of suspended solids.
- Storage to improve water quality should be equivalent to 7-15 days of average water demand
- Problems: siltation, eutrophication

Stratifications of lake during summer and winter



Thermocline layer may be defined as a change in temperature with 1m depth with greater than 1°C. The depth of epilimnion is related to the size of the lake and could be as little as 1 m in small lake besides the storm activity in spring when stratification is developing (Davies and Cornwell, 1998). Arhonditsis and Brett (2005) reported that the level of dissolved oxygen at the epelimnion layer for Lake Washington was as high as 11 mg/L in March to as low as 9 mg/L in August for the reporting years 1995 to 2001. For the same period the observed values of total phosphorus and total nitrogen were at the higher range of 22 ug/L and 450 ug/L respectively in May

Treatment process



Chemical pre-treatment

- -to destroy algae using copper sulphate
- -to kill bacteria using chlorine especially during the outbreak of cholera etc.
- Chemicals will be dosed as the raw water enters the treatment plant

Course screen

- To screen large material such as leaves, debris and other sizeable clogging materials.
- Trash racks:
- Clear opening 75 to 100mm
- Inclined or vertical
- Manually or mechanically cleaned

- Bar racks:
- Clear opening 20mm to 75mm
- Inclined or vertical. Normally inclined at 30⁰ from vertical
- Manually or mechanically cleaned
- Flow rate through the screen between 0.6 to 1 m/s, could be increased to 1.2 or 1.4 m/s at maximum flow.

Fine screen

- To remove materials too small to be deflected by course screen. Water velocity < 60cm/s at maximum design flow.
- Types:
- Traveling water screen 6 to 9mm mesh cloth, normally installed after course screen.
- Basket screen 1/8 in to $\frac{3}{4}$ in most common
- Dish or drum screen vertical position, water spray cleaning



Drum screen (2m - 5m diameter) and microstrainer (opening 15 to 64 microns)



Fig 8.1 Examples of screens: (a-i) manually-raked bar screen, (a-ii) mechanically-raked bar screen, (b) drum screen, (c) microstrainer

Intake











Aeration

- Increase DO in the water and removing dissolved gases such as carbon dioxide, methane and hydrogen sulphate
- Reduce taste and odour and corrosive property of water
- To oxidise iron and manganese

Types of aerators

- Water into air aerators cascade aerator cone aerator, slat and coke tray aerator, draft aerator, spray aerator
- Air into water aerators diffuse aerator, draft tube aerator
- Combination aerators mechanical aerators, pressure aerator





FIGURE 8.2 - MULTIPLE-PLATFORM AERATOR



FIGURE 8.3 - SPRAY AERATOR

Examples of air-into-water aerators are:-

- (a) venturi aerators, and
- (b) draft tube aerators.



FIGURE 8.4 - VENTURI AERATOR





FIGURE 8.5 - DRAFT TUBE AERATOR





Sprayer with combination of cascade & draft

diffuser



Pre-chemical treatment

- (i) pre-chlorination
- Raw water is contaminated but not so turbid
- To control algae and other microorganisms
- Dose around 1 mg/L
- (ii) copper sulphate treatment
- To control algae and aquatic plants
- Certain freshwater algae are toxic, e.g. blue-green algae will cause colouration, corrosion of concrete and steel and toxic to freshwater. Others are green, diatoms, flagellate.
- Dosage depends on temperature and alkalinity, normally at 0.12 mg/L to 0.3 mg/L.
- Safe dosage for most aquatic fish at 0.5 mg/L

2. RAPID MIX

- The desired reactions between the coagulant and water are irreversible and take place in a fraction of a second.
- mixing must be very fast.
- Coagulant should be injected at the eye of the impeller for maximum turbulence.

-----RAPID MIX

- The loss of head through the mixing stage is low, but the electrical equipment adds to the maintenance work.
- The power required is 3 to 15 watts per I/s.

Rapid mixing - coagulation

- Two types:
- 1. Mechanical propeller or paddle type mechanical mixers.
- 2. Hydraulic V notch weirs, Parshall flumes, orifices, throttles valves etc.

Mechanical mixer

 Typical design values for most mechanical rapid mix system provides detention time t of 10 to 60 sec and G values of 600 to 1000s⁻¹.



Hydraulic mixing:

- Baffles; high level of turbulence necessary.
- Weirs; injection from several points across width of channel.





- Hydraulic mixing:
- Total head loss across a throttled valve should not exceed 3.2m for mixing coagulant chemicals
- Energy provides by a weir with an effective fall of 30cm provides a G value of 1,000s⁻¹ at 20^oC.
- Weir mixers normally use for flow rate <150 MLD due to maintenance problems with multiple orifice chemical feed manifolds.

-----Hydraulic mixing



Weirs



-----Hydraulic mixing

- Hydraulic jump but at low flows much of the water may by-pass the turbulent 'roller'.
- Jets.
- Orifice in pipeline





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-----Hydraulic mixing

- The sequence of dosing (if several chemicals are used) is best determined by experimentation at the plant concerned.
- If the chemical is dosed under water, the system should be designed to allow visual inspection of the flow of coagulant.
- Air has been used to induce agitation, but the bubbles attached to flocs may prevent settlement.

Hydraulic jump

- Occurs when a supercritical flow meets a subcritical flow
- The resulting flow transition is rapid and involves a large energy loss due to turbulence
- F_r >1 : supercritical flow; water velocity>wave velocity; disturbances travel downstream; upstream water levels are unaffected by downstream control
- F_r<1 : Subcritical flow; water velocity<wave velocity; disturbances travel upstream and downstream; upstream water levels are affected by downstream control

Hydraulic jump



Hydraulic Jump

$$\frac{d_1}{d_2} = \frac{1}{2} \left[\sqrt{\left(1 + 8 F^2\right)} - 1 \right]$$

Where:

 d_1 = depth of H₂O upstream of the jump (m). d_2 = depth of H₂O downstream of the jump (m).

F = Froude No. =
$$\frac{V_1}{(gd_1)^{\frac{1}{2}}}$$

where V_1 = velocity of flow upstream of the jump (m/s) g = 9.81 m/s². For a jump:

$$\frac{d_2}{d_1} > 2.4$$

$$F_r > 2$$

When Fr between 4 & 9, the energy consumed in turbulence can be between 45 & 70%, which is adequate for rapid mixing. The head losses is typically 0.3m or greater

Criteria for Rapid Mixing:

- Performance of a square tank is better than cylindrical tank.
- Stator baffles may provide an advantage for appropriate mixing.
- Flat blade is better than propeller impeller.
- Introducing the coagulant at the axle of the agitator may encourage coagulation.



Mechanical rapid mixer



1. DOSING

- Dosing pumps (positive displacement) are calibrated to supply chemicals at a known rate.
- The calibration can be checked using a graduated tube.
- Lime is dosed as a slurry, not as a solution. For this reason it must be kept constantly agitated, and pipes conveying the slurry should be flushed with water regularly



DOSING

- For small works constant head devices can be used to dose chemicals .
- the head governing the flow is constant whatever the level of the solution in the container .
- Another system doses a fixed volume of coagulant each time a tank is filled.
- Plants without a power supply may use sodium carbonate to raise the pH since it can be dosed as a solution.



Terminology

- <u>Suspension</u> a dispersion of solid particles in a liquid. These particles scatter light, making aqueous suspensions cloudy or turbid.
- <u>Colloid</u> -Colloidal particles are very small (1 to 200 x 10⁻⁹ m) and form <u>stable</u> <u>suspension</u> because the random brownian motion- of the water molecules is enough to prevent the particles settling.

-----Terminology

- Floc a collection of colloidal particles held together.Flocs generally have little strength and a density close to the density of water.
- <u>Coagulation</u> -the process of making conditions favorable for colloidal particles to join together to form flocs by biological or chemical processes.
- Flocculation -is the growth of flocs

- In normal English, the term 'coagulate' means to 'change from fluid to more or less solid state, clot, curdle, set or solidify'.
- Flocculate is a small portion of matter like a flock of wool.
- Source: Water and Wastewater Engineering System, Barnes et al., 1981, page 199.

Settling velocity for different particle sizes

Particle size (mm)	Size for	Settling velocity (ms ⁻¹)
10	Gravel	0.73
1	Course sand	0.23
0.1	Fine sand	1.0 x 10 ⁻²
0.01	Silt	1.0 x 10 ⁻⁴
0.0001	Large colloids	1.0 x 10 ⁻⁸ (0.3 m/year)
0.000001	Fine colloids	1.0 x 10 ⁻¹³ (3 m/million yr)



Sec.

Fig 7.2 Applications of unit processes



GE Osmonics

Filtration and Separation Spectrum



Coagulation - mechanisms

 In general colloidal suspensions in water consists of clay particles and other organic matters which have negative charges and always repel each others. Brownian motion prevents the settlement of particles under

- Colloids may consist of:
- 1. colloidal clay causes turbidity
- 2. large organic molecules cause colour
- 3. proteins, carbohydrates and fats in sewage
- 4. oxides of iron, manganese and silica
- 5. precipitated aluminium hydroxide etc.

- Colloids are *hydrophilic* when the factor causing stability due to the attraction between particles and water. It means water loving. Such compounds have an affinity to water and are usually charged or have polar side groups to their structure that will attract water.
- Colloids are *hydrophobic* when no attraction between particles and water. It means water hating. These compounds are repelled by water and are usually neutral (no charge.)

- Hydrates of iron and aluminium form *hydrophobic* colloid in water.
- Proteins, starches or fats (*hydrophilic*) become stable as a result of mutual electrostatic repulsion.
- Colloidal particles have a very large ratio of surface area to volume. For e.g. 1cm³ of material if divided into cubes of 10⁻⁵ mm on a side (mid-point of colloidal range) would have a surface area of 600m².

- Destabilisation has to be made by adding coagulants with positive charges.
- Thus reducing the repelling forces between the particles and promotes the agglomeration of colloids.



- The most well-known repulsive force is caused by the interaction of double layers of the surfaces (electrostatic stabilization)
- The most important attractive force is called the London-van der Waals force, which arises from magnetic field within the particles and in the space between them
- These two types of forces form the basis of the Derjaguin, Landau, Verwey, and Overbeek theory of colloid stability.

- Other forces include those associated with the hydration of ions at the surfaces (repulsive) and the presence of adsorbed polymers which can cause either repulsion (steric interaction) or attraction (polymer bridging).
- The repulsive force due to displacement of fluid (as particles approach one another on a collision course where the fluid between them must move out of the way) is called hydrodynamic retardation.
Guoy-Stern colloidal model (Figure in the next slide)

- Stationary charged layer on the surface is surrounded by a bound of water in which ions of opposite charge drawn from the bulk solution produce a rapid drop in potential. This drop within the bound-water layer is called stern potential.
- A more gradual drop, called zeta potential, occurs between the shear surface of the bound-water layer and the point of electroneutrality in the solution



Negatively charged particle, the diffused double layer and location of zeta potential

CHEMICALS

- <u>Alum</u> widely used as a primary coagulant. available as lumps, grains or powder, or as a solution which is strongly acidic.
- For optimum floc formation the pH should generally be between 6 and 8.
- Acid is formed when alum reacts with water.
- 1 mg/l alum reacts with 0.39 mg/l 95% hydrated lime as Ca(OH)₂.

Iron salts

- Operate over a wider pH range than alum.
- Iron flocs are heavier than aluminium flocs but may be harder to remove from filter sand.
- Ferrous sulphate is sometimes cheaper than alum because it is a byproduct of steel pickling. It must be oxidised to the ferric state usually by chlorine, when the two are known as chlorinated copperas.
- Ferric sulphate and ferric chloride may also be used.

Coagulant dosage

Coagulants	Dose (mg/l)				
Aluminium Sulphate (Alum) Al ₂ (SO ₄) ₃ .18H ₂ O	10 – 50				
Ferric Sulphate $Fe_2 (SO_2)_3$	10 – 50				
Sodium Aluminate Na ₂ Al ₂ NO ₄	5 – 30				
Ferrous Sulphate FeSO ₂	5 – 25				

pH correction

- is necessary to raise the pH, and hydrated lime is usually preferred for this purpose on cost grounds.
- However lime has a low solubility and so is usually fed as a slurry-needs constant agitation.
- Where constant agitation is not provided sodium carbonate (washing soda) is used, because it dissolves readily in water.
- Sulphuric acid is used if the pH must be lowered.

Coagulant aids

- (i) Polyelectrolytes are used to make flocs larger, tougher and denser. They have long molecules with charged sites.
- Three types of polyelectrolytes: cationic, anionic and nonionic or ampholytic.
- The best type must be chosen by experiment.
- Polyelectrolytes may be used to reduce the requirement for coagulant and alkali.

-----Coagulant aids

- Synthetic polyelectrolytes (e.g. polyacrylamides) are banned in some countries, because the monomer an impurity is a health hazard.
- (ii) <u>Activated silica</u> produces a similar result to polyelectrolytes, but must be prepared with great precision.
- (iii) <u>Purified clays</u> may be added to waters with low turbidity or high colour to add weight to the flocs and reduce the coagulant requirement.

FLOCCULATION

Laminar settling: Stokes' law

$$V = \frac{g}{18} \frac{\left(\rho_s - \rho_l\right)}{\mu} d^2$$

Where

V is settling velocity ρ_s is density of particle ρ_l is density of liquid d is diameter of particle





-----FLOCCULATION

- Flocculation forms particles of larger 'diameter'.
- Flocs grow when particles are colliding with each other and the chemistry of the system enables particles to stick together.
- <u>Perikinetic Flocculation</u> (Brownian motion). Bombardment of colloidal particles by molecules causes collisions. The effect on larger flocs is negligible.

-----<u>FLOCCULATION</u>

 Orthokinetic Flocculation- Smoluchowski's model. If y < (r1+r2) and V1 > V2 the two particles will collide.



Orthokinetic Flocculation

• The velocity gradient G can be written as:

$$G = \frac{V_1 - V_2}{y}$$

• The velocity gradient must be induced by impellers rotated slowly by electric motors or by frequent changes in direction of the flow

Smoluchowski's model

• The Smoluchowski model leads to the result that the total number of collisions per unit volume in unit time is given by:

$$N = \frac{2}{3}n^2 G d^3$$

- where n is the number of particles per unit volume
- and *d* is the particle diameter

So the rate of flocculation is proportional to:

- the velocity gradient
- the volume of the sphere of influence of the particles
- the square of the numerical concentration of the particles



DESIGN OF FLOCCULATORS

- Facts:
- Common jar-test equipment gives no information regarding full-scale velocity gradients.
- Typical values of velocity gradient are between 20 to 100 s⁻¹.
- The Camp Number is the product of velocity gradient and detention time, Gt, and reflects the total number of collisions. Typical value are 2x10⁴ to 2x10⁵.



FLOCCULATORS

- Detention times should be greater than 10 minutes.
- The quality of floc produced can be assessed by observing how the floc reflects light.
- The average velocity gradient can be related to the energy dissipated:

$$G = \left(\frac{P}{\mu V}\right)^{1/2}$$



Design

- Power input is given by the following equation:
- $\mathsf{P}=\mathsf{F}_\mathsf{D}\mathsf{V} = \frac{1}{2}C_D\rho AV^3$
- P is in watts, F_D is the drag force (mkg/s²), C_D is the drag coefficient, A is the submerged area of the paddle in square metre, and V is the relative velocity of the paddle in m/s. V is approximated as 0.75 x peripheral velocity of the paddle=1.5πrn, where n is in rps

-----design

- Rotation of the paddle varies from 2 to 15
 revolution per minute
- Peripheral velocity of paddle from 0.2 to 0.8 m/s

(i) <u>Hydraulic Flocculator</u>

- The flow is directed around, or over and under, fixed baffles so that the changes in direction produce the required turbulence.
- There are no moving parts.
- they cannot be adjusted to suit varying conditions

-Hydraulic Flocculator

- The rate of energy dissipation is given by the mass flow rate ρg and the head loss h.
- *P*=Q*ρ*g*h*

$$G = \sqrt{\frac{P}{V\mu}}$$
$$\sqrt{\frac{Q\rho gh}{V\mu}}$$

--<u>Hydraulic Flocculator</u>

 For turbulent flow the head loss is approximately proportional to the square of the flow rate (i.e. h=KQ²) so the effect of variations in flow rate can be assessed using:

$$G = \sqrt{\frac{KQ^3 \rho g}{V\mu}}$$

---<u>Hydraulic Flocculator</u>

 The head loss occurring in a baffle flocculator can be estimated by assuming that the kinetic energy is lost at each change in direction.



 $\Delta H = (v_1^2 + 5v_2^2 + 4v_3^2)/2g + \text{normal channel friction}$

----<u>Hydraulic Flocculator</u>

- Instead of baffle walls, pipes may be used to produce jets.
- The velocities in such flocculators are typically in the range 0.15 to 0.5 m/s

Mechanical Flocculator

- Velocity gradients are caused by the rotation of a paddle about a horizontal shaft or of an impeller about a vertical shaft.
- The paddle tip speed should not exceed 0.6 m/s to prevent excessive local velocity gradients.
- Vertical-shaft systems need no underwater bearings and produce a more uniform distribution of velocity gradients

The mixing effect of the paddle or impeller some of the water may pass through the vessel in a much shorter time, as illustrated by the table below:

Number of tanks in series	1	3	5
Percentage of the flow that passes through the flocculation stage	39	20	11
in less than half of the theoretical detention time			

Example: coagulation

- A treatment plant has a capacity of 2 million litres per day. Estimate the Camp number, size of the tank and power input in the coagulation tank based on the following data:
- Retention time = 1.5 minutes
- Velocity gradient, $G = 700s^{-1}$
- Dynamic viscosity = 0.87×10^{-3} kgm⁻¹ s⁻¹

Retention time = 1.5 minutes Velocity gradient, $G = 700s^{-1}$ Dynamic viscosity = $0.87x10^{-3}$ kgm⁻¹s⁻¹ Q = 2 MLD

Camp No =Gt = 700 x 1.5x60 = 63,000 Volume V = Qt = $\frac{2,000}{24x60}$ x1.5 = 2.08 m³

Velocity gradient $G = \left(\frac{P}{\mu V}\right)^{1/2}$ Therefore power input, $P = G^2 \mu V = 700^2 x 0.87 x 10^{-3} x 2.08 = 886.7$ watt





Example

 A sinuous channel has 15 round-the-end cross-walls. Water is passed along with a velocity of 0.2m/s between cross-wall and 0.5 m/s round the ends. The flow is 0.3 m³/s and the nominal retention time is 25 min. for a temperature of 10°C, estimate the additional loss of head, the power dissipated, velocity gradient and Camp Number.

Properties of saturated water

Temp.	Saturation Pressure P _{saty} kPa	Density p. kg/m/		Enthalpy of Vaporization	Specific Heat cj. 1%g - K		Thermai Conductivity k. W/m - K		Dynamic Viscosity µ, kg/m - s		Prandti Number Pr		Volume Expansion Coefficient 8 14K
7, °C		Liqu/d	Vapor	h _{ie} , kJ/kg	Uquid	Vapor	Liquid	Vapor	Liquid	Vapor	Liquid	Vapor	Liquíd
0.01 5 10 15 20	D.6113 0.8721 1.2276 1.7051 2.339	999.8 999.9 999.7 999.1 998.0	0.0048 0.0068 0.0094 0.0128 0.0173	2501 2490 2478 2466 2454	4217 4205 4194 4186 4182	1864 1857 1862 1863 1867	0.561 0.571 0.580 0.589 0.598	0.0171 0.0173 0.0176 0.0179 0.0182	1.792×10^{-3} 1.519×10^{-3} 1.307×10^{-3} 1.138×10^{-3} 1.002×10^{-3}	$\begin{array}{c} 0.922\times10^{-5}\\ 0.934\times10^{-5}\\ 0.946\times10^{-5}\\ 0.959\times10^{-5}\\ 0.973\times10^{-5}\\ \end{array}$	13.5 11.2 9.45 8.09 7.01	1.00 1.00 1.00 1.00 1.00	$\begin{array}{c} -0.068 \times 10^{-3} \\ 0.015 \times 10^{-3} \\ 0.733 \times 10^{-2} \\ 0.138 \times 10^{-3} \\ 0.195 \times 10^{-3} \end{array}$
25 30 35 40 45	3.169 4.246 5.628 7.384 9.593	997.0 996.0 994.0 992.1 990.1	0.0231 0.0304 0.0397 0.0512 0.0655	2442 2431 2419 2407 2395	4180 4178 4178 4179 4180	1870 1875 1880 1885 1892	0.607 0.615 0.623 0.631 0.637	0.0186 0.0189 0.0192 0.0196 0.0200	$\begin{array}{c} 0.891 \times 10^{-3} \\ 0.798 \times 10^{-3} \\ 0.720 \times 10^{-5} \\ 0.653 \times 10^{-3} \\ 0.596 \times 10^{-3} \end{array}$	$\begin{array}{c} 0.987 \times 10^{-5} \\ 1.001 \times 10^{-5} \\ 1.016 \times 10^{-5} \\ 1.031 \times 10^{-5} \\ 1.046 \times 10^{-5} \end{array}$	6.14 5.42 4.83 4.32 3.91	1.00 1.00 1.00 1.00 1.00	$\begin{array}{c} 0.247 \times 10^{-3} \\ 0.294 \times 10^{-3} \\ 0.337 \times 10^{-3} \\ 0.377 \times 10^{-3} \\ 0.415 \times 10^{-3} \end{array}$
50 55 60 65 70	12.35 16.76 19.94 25.03 31.19	988.1 985.2 983.3 980.4 977.5	0.0831 0.1045 0.1304 0.1614 0.1983	2383 2371 2359 2346 2334	4181 4183 4185 4187 4190	1900 1908 1916 1926 1936	0.644 0.649 0.654 0.659 0.663	0.0204 0.0208 0.0212 0.0216 0.0221	$\begin{array}{c} 0.547 \times 10^{-5} \\ 0.504 \times 10^{-3} \\ 0.467 \times 10^{-3} \\ 0.433 \times 10^{-3} \\ 0.404 \times 10^{-3} \end{array}$	$\begin{array}{c} 1.062 \times 10^{-5} \\ 1.077 \times 10^{-5} \\ 1.093 \times 10^{-5} \\ 1.110 \times 10^{-5} \\ 1.125 \times 10^{-3} \end{array}$	3.55 3.25 2.99 2.75 2.55	1.00 1.00 1.00 1.00 1.00	$\begin{array}{c} 0.451\times 10^{-3}\\ 0.484\times 10^{-3}\\ 0.517\times 10^{-3}\\ 0.548\times 10^{-3}\\ 0.578\times 10^{-3} \end{array}$
75 80 85 90 95	38.58 47.39 57.83 70.14 84.55	974.7 971.8 968.1 965.3 961.5	0.2421 0.2935 0.3536 0.4235 0.6045	2321 2309 2296 2283 2270	4193 4197 4201 4206 4212	1948 1962 1977 1993 2010	0.667 0.670 0.673 0.675 0.677	0.0225 0.0230 0.0235 0.0240 0.0246	$\begin{array}{l} 0.378 \times 10^{-3} \\ 0.355 \times 10^{-3} \\ 0.333 \times 10^{-3} \\ 0.315 \times 10^{-3} \\ 0.297 \times 10^{-3} \end{array}$	$\begin{array}{c} 1.142 \times 10^{-5} \\ 1.159 \times 10^{-5} \\ 1.176 \times 10^{-5} \\ 1.193 \times 10^{-5} \\ 1.210 \times 10^{-5} \end{array}$	2.38 2.22 2.08 1.96 1.85	1.00 1.00 1.00 1.00 1.00	$\begin{array}{c} 0.607 \times 10^{-3} \\ 0.653 \times 10^{-3} \\ 0.670 \times 10^{-3} \\ 0.702 \times 10^{-3} \\ 0.716 \times 10^{-3} \end{array}$
100 110 120 130 140	101.33 143.27 198.53 270.1 361.3	957.9 950.6 943.4 934.6 921.7	0.5978 0.8263 1.121 1.496 1.965	2267 2230 2203 2174 2145	4217 4229 4244 4263 4286	2029 2071 2120 2177 2244	0.679 0.682 0.683 0.684 0.683	0.0251 0.0262 0.0275 0.0288 0.0301	$\begin{array}{c} 0.282 \times 10^{-3} \\ 0.255 \times 10^{-3} \\ 0.232 \times 10^{-2} \\ 0.213 \times 10^{-3} \\ 0.197 \times 10^{-9} \end{array}$	$\begin{array}{c} 1.227 \times 10^{-8} \\ 1.261 \times 10^{-5} \\ 1.296 \times 10^{-5} \\ 1.330 \times 10^{-6} \\ 1.365 \times 10^{-5} \end{array}$	1.75 1.58 1.44 1.33 1.24	1.00 1.00 1.00 1.01 1.02	$\begin{array}{c} 0.750 \times 10^{-3} \\ 0.798 \times 10^{-3} \\ 0.858 \times 10^{-3} \\ 0.913 \times 10^{-3} \\ 0.970 \times 10^{-3} \end{array}$
150 160 170 180 190	475.8 617.8 791.7 1,002.1 1,254.4	916.6 907.4 897.7 887.3 876.4	2.546 3.256 4.119 5.153 6.388	2114 2083 2050 2015 1979	4311 4340 4370 4410 4460	2314 2420 2490 2690 2710	0.682 0.680 0.677 0.673 0.669	0.0316 0.0331 0.0347 0.0364 0.0382	$\begin{array}{c} 0.183 \times 10^{-3} \\ 0.170 \times 10^{-3} \\ 0.160 \times 10^{-3} \\ 0.150 \times 10^{-3} \\ 0.142 \times 10^{-3} \end{array}$	1.399×10^{-5} 1.434×10^{-5} 1.468×10^{-5} 1.502×10^{-5} 1.537×10^{-5}	1.16 1.09 1.03 0.983 0.947	1.02 1.05 1.05 1.07 1.09	$\begin{array}{c} 1.025\times10^{-3}\\ 1.145\times10^{-3}\\ 1.178\times10^{-3}\\ 1.210\times10^{-3}\\ 1.210\times10^{-3}\\ 1.280\times10^{-3}\end{array}$
200 220 240 260 280	1,553 8 2,318 3,344 4,688 6,412	864.3 840.3 813.7 783.7 750.8	7.852 11.60 16.73 23.69 33.15	1941 1859 1767 1663 1544	4500 4610 4760 4970 5280	2840 3110 3520 4070 4835	0.663 0.650 0.632 0.609 0.581	0.0401 0.0442 0.0487 0.0540 0.0605	$\begin{array}{c} 0.134 \times 10^{-3} \\ 0.122 \times 10^{-3} \\ 0.111 \times 10^{-3} \\ 0.102 \times 10^{-3} \\ 0.094 \times 10^{-3} \end{array}$	$\begin{array}{c} 1.571 \times 10^{-6} \\ 1.641 \times 10^{-5} \\ 1.712 \times 10^{-5} \\ 1.788 \times 10^{-6} \\ 1.870 \times 10^{-5} \end{array}$	0.910 0.865 0.836 0.832 0.854	1.11 1.15 1.24 1.35 1.49	$\begin{array}{c} 1.350\times10^{-3}\\ 1.520\times10^{-3}\\ 1.720\times20^{-3}\\ 2.000\times10^{-3}\\ 2.380\times10^{-3}\end{array}$
300 320 340 360 374.14	8,581 11,274 14,586 18,651 22,090	713.8 667.1 610.5 528.3 317.0	46.15 64.57 92.62 144.0 317.0	1405 1239 1028 720 0	5750 6540 8240 14.690	5980 7900 11.870 25,800	0.548 0.509 0.469 0.427	0.0695 0.0836 0.110 0.178	$\begin{array}{c} 0.086 \times 10^{-3} \\ 0.078 \times 10^{-3} \\ 0.070 \times 10^{-3} \\ 0.060 \times 10^{-3} \\ 0.043 \times 10^{-3} \end{array}$	$\begin{array}{c} 1.965 \times 10^{-5} \\ 2.084 \times 10^{-5} \\ 2.255 \times 10^{-5} \\ 2.571 \times 10^{-5} \\ 4.313 \times 10^{-5} \end{array}$	0.902 1.00 1.23 2.06	1.69 1.97 2.43 3.73	2.950 × 10 ^{. 5}

Note 1: Kinematic viscosity μ and thermal diffusivity α can be calculated from their definitions, $\nu = \mu/\rho$ and $\alpha = k/\rho c_{\rho} = \nu/Pr$. The temperatures 0.01°C, 100°C, and 374.14°C are the triple-, boiling-, and critical-point temperatures of water, respectively. The properties listed above (except the vapor density) can be used at any pressure with negligible error except at temperatures near the critical-point value.

Note 2: The unit kJ/kg+°C for specific heat is equivalent to kJ/kg - K, and the unit W/m - °C for thermal conductivity is equivalent to W/m - K.

Source: Niscosity and thermal conductivity data are from J. V. Sengers and J. T. R. Watson, Journal of Physical and Chemical Reference Data 15 (1986), pp. 1291–1322. Other data are obtained from various sources or calculated.

Solution

- Additional head loss $\Delta H = (nv_1^2 + (n-1)v_2^2)/2g$
- Where n-1 is the number of baffles
- V_1 is the velocity between the baffles, m/s
- V_2 is the velocity at the baffle slots, m/s
- Therefore additional head loss
- = $(16x0.2^2+15x0.5^2)/2x9.81=0.224m$
- P=pghQ=1000x9.81x0.224x0.3=659w
- Vol. = txQ=25x0.3x60=450
- $G=(p/\mu x Vol.)^{1/2}=[659/(1.31x10^{-3}x450)]^{1/2}=33.4/s$

Example

 A flocculator is designed to treat 75,000 m³/day is 30m long, 12m wide and 4.5m deep. It is equipped with 0.3m paddles supported parallel to and moved by 4 horizontal shafts which rotate at a speed of 2.5 rpm. The centre line of the paddle is 1.7m from the shaft, which is at mid depth of the tank. Two paddles are mounted on each shaft, one opposite the other. Assuming that the mean velocity of the water is approximately 1/4 the velocity of the paddles, that the drag coefficient of the paddles is 1.8 and that the temperature of water is 10° C,

- Determine:
- (a) the differential velocity between paddles and the water
- (b) the useful power input and the energy consumption
- (c) the detention time
- (d) the values of G and Gt
Solution

$$P = F_D V = \frac{1}{2} C_D \rho A V^3$$

• P=power input (w), F_D=drag force (mkg/s²), C_{D} =drag cofficient, ρ =density (kg/m³), A=submerged area of the paddles(m^2), V = relative velocity of the paddle (m/s). V may be approximated to 0.75 times the peripheral velocity of the paddle or equals $1.5\pi rn$ where r is relative radius of the paddle (m) and n is the no. of revolution per second. Please do the rest!

• Example

Two sets of jar tests are conducted on raw water containing 30 NTU and an alkalinity concentration of 40 mg/L expressed as CaCO₃. Given the data as shown in Table 1, find the optimal pH, coagulant dose and the theoretical amount of alkalinity that would be consumed at the optimal dose.

The reaction occurs when water is added with alum is shown below: $AI_2(SO_4)_3.14H_2O + 6HCO_3^- \rightarrow 2AI(OH)_3 + 6CO_2 + 14H_2O + 3SO_4^{-2-1}$ Molecular weights of the elements are shown as follows: Oxygen = 16, sulphur = 32, aluminium = 27, hydrogen =1 and carbon = 12.

Table 1

Jar Test 1									
Jar	1	2	3	4	5	6			
pН	5.0	5.5	6.0	6.5	7.0	7.5			
Alum dose (mg/L)	15	15	15	15	15	15			
Turbidity (NTU)	14	8	4.5	6.0	9	13			
Jar Test 2									
Jar	1	2	3	4	5	6			
pН	6.0	6.0	6.0	6.0	6.0	6.0			
Alum dose (mg/L)	6	8	12	14	16	18			
Turbidity (NTU)	14	10	4.5	4.0	6	13			





The amount of alkalinity used can be obtained from the following equation: Al₂(SO₄)₃.14H₂O + 6HCO₃⁻ \rightarrow 2Al(OH)₃+6CO₂ + 14H₂O + 3SO₄²⁻

- Where 1 mole of alum used 6 moles of
- MW of alum as follows:

•
$$AI_2 = 27 \times 2 = 54$$

- $S_3 = 32 \times 3 = 96$
- $O_{12} = 16 \times 12 = 192$
- $H_{28} = 1 \times 28 = 28$
- $O_{14} = 16 \times 14 = 224$
- Total = 594

- From the graph optimum dose of alum is **13.5**mg/L
- Molarity or molar is expressed in mg/L by the following expression;
- $Mg/L = Molarity \times MW \times 10^3$
- =(Mol/L) (g/Mol) (10³ mg/g)..... (1)
- .:. Mole of alum added per litre =

$$\frac{13.5 \times 10^{-3} \text{ g/L}}{594 \text{ g/Mol}} = 2.27 \times 10^{-5} \text{ Mol/L}.$$

- Where it will consume:
- 6(2.27x10⁻⁵) = **1.36x10⁻⁴** M or molar
- Note: 1 molar solution has 1 mole substance per litre solution.
- •
- Molecular weight = 61 i.e. (1+12+48),
- ∴ from equation (1),
- (1.36x10⁻⁴ mol/L) (61 g/mol) (10³ mg/g) =
 8.31 mg/L has been used

This can be expressed in the following equation :

mg/L as CaCO₃ = $(mg/L \text{ spesis}) \left[\frac{\text{EW CaCO}_3}{\text{EW spesis}} \right]$ (2)

Where EW = equivalent weight

$$\therefore$$
 mg/L asCaCO₃ = $8.31x \frac{50}{61}$ = 6.811 mg/L HCO₃⁻ as CaCO₃
Note: EW= equivalent weight = $\frac{\text{molecular weight or atomic}}{n}$

where in precipitation reaction is the valence of the element in question, in the acid/base reaction, n is the number of hydrogen ionsthat the molecule transfers

Theory for CHEMICAL PRECIPITATION

- Involves the addition of chemicals to alter the physical state of dissolved and suspended solids and to facilitate their removal by sedimentation.
- May result in the net increase in the dissolved constituents in the wastewater.
- Have been developed for the complete secondary treatment of the untreated wastewater including for the removal of nitrogen or phosphorous or both.

Chemical Precipitation for Improving Plant Performance

- Was used to enhance the degree of suspended solids and BOD removal.
- Since 1970 chemical precipitation used for more complete removal of organic compounds and nutrients contained in wastewater.

The most common chemicals used are shown in the following table

Chemical	Formula	Molecular	Dry Density	Liquid
		weight	$(lb/ft^3) x$	density
			$16.0185 = kg/m^3$	(lb/ft^3)
Alum	Al ₂ (SO ₄) ₃ .18H ₂ O*	666.7	60-75	78-80(49%)
	$Al_2(SO_4)_3.14H_2O^*$	594.3	60-75	83-85(49%)
Ferric	FeCl ₃	162.1		84-93
chloride				
Ferric	$\operatorname{Fe}_{2}(\operatorname{SO}_{4})_{3}.$	400		70-72
sulphate	$Fe_2(SO_4)_3.3H_2O$	454		
Ferrous	$FeSO_4$) ₃ .7H ₂ O	278.0	62-66	
sulphate				
(copperas)				
Lime	Ca(OH) ₂	56 as Ca O	35-50	

PERIODIC TABLE OF THE ELEMENTS

(Modified from a table published and copyrighted by Sargent-Welch Scientific Company, Skokie, Illinois, and used with their permission)

Group	
	VIIIA
1 1.0079	2 4.0026
252.7 1 BOILING V OXIDATION STATES	-268.9
-259.2 H POINT, °C 30 65.37	-269.7
	0.126 ПС
Hydrogen IIA MELTING - 419.5 7 (1) Atomic weights are 1971 values. Parentheses	IVA VA VIA VIIA Helium
3 6.939 4 9.0122 POINT, °C 7.14 ZII SYMBOL (1) Home weights are 1971 values. Fachnices 5 10.811	6 12.01115 7 14.0067 8 15.9994 9 18.9984 10 20.183
1330 2770 Indicated most stable or best known isotope.	4830 4830 -195.8 33.5.4.2 -183 -2 -188.2 -1 -246
1205 Li 1277 Be (2030) B (2030) B (2030) B (2030) B (2030)	3727 C -210 N -218.8 O -219.6 F -248.6 No
Libburg Berullum (grint) (2)	
NAME Induits at the boiling point.	Carbon Nitrogen Oxygen Pluorine Neon
11 22.9998 12 24.312 (3) Names and symbols for elements 104 and 105 13 26.9915 (3) Names and symbols for elements 104 and 105 14 20 (3) Names and 105 14 20 (3) N	$14 \ 28.086 \ \ 15 \ 30.9738 \ \ 16 \ 32.064 \ \ 17 \ 35.453 \ \ 18 \ 39.948$
are proposed but not vet officially accepted.	2680 280 444.6 -34.7 -185.8
	$\begin{array}{c ccccccccccccccccccccccccccccccccccc$
Sodium Magnesium IIIB IVB VB VIB VIIB IB IIB Aluminum I	Silicon Phosphorus Sulfur Chlorine Argon
19 39.098 20 40.08 21 44.956 22 47.90 22 50.942 24 51.996 25 54.938 26 55.947 27 59.932 29 59.71 20 52.54 20 55.59 24 50.70	
$\begin{bmatrix} 1 \\ 2 \\ 3 \\ 3 \\ 3 \\ 3 \\ 3 \\ 3 \\ 3 \\ 3 \\ 3$	32 72.59 33 74.922 34 78.96 35 79.909 36 83.80
100 2000 2000 2000 2000 2000 2000 2000	2830 613 685 58 -152
	5.32 Gel 5.72 AS 4.79 Se 3.12 Br 2.6 Kr
Potassium Calcium Scandium Titanium Vanadium Chromium Manganese Iron Cobait Nickel Copper Zinc Galilium	Germanium Arsenic Selenium Bromine Krypton
37 85.47 38 87.62 39 88.905 40 91.22 41 92.906 42 95.94 43 (98) 44 101.07 45 102.905 46 106.4 47 107.870 48 112.40 49 114.82 1	50 118.69 51 121.75 52 127.60 53 126.904 54 131.30
688 ¹ 1380 ² 2927 ³ 3580 ⁴ 3300 ^{5,3} 5560 ^{6,5,4,3,2} - ⁷ 4900 ^{2,3,4} 4500 ^{2,3,4,6,8} 3980 ^{2,4} 2210 ¹ 765 ² 2000 ³	2270 4,2 1380 ±3,5 989.8 -2,4,6 183 ±1,5,7 -108.0
38.9 Bh 768 Sr 1509 V 1852 7r 2468 Nh 2610 Mo 2140 To 2500 Du 1966 Dh 1552 Dd 960.8 A o 320.9 Cd 155.2 In 2	231.9 Co 630.5 Cb 449.5 To 113.7 L -111.9 Vo
1.53 TID 2.6 SI 4.47 I 6.49 ZI 8.4 ND 10.2 MO 11.5 IC 12.2 RU 12.4 RII 12.0 FU 10.5 AG 8.65 CU 7.31 III 7	
Rubidium Strontium Yttrium Zirconium Nioblum Molybdenum Technetium Ruthenium Rhodium Palladium Silver Cadmium Indium	Tin Antimony Tellurium Iodine Xenon
55 132.9054 56 137.34 57 138.91 72 178.49 73 180.948 74 183.85 75 186.2 76 190.2 77 192.2 78 195.09 79 196.967 80 200.59 81 204.37 8	82 207.19 83 208.980 84 (210) 85 (210) 86 (222)
690 1640 3470 $\frac{1}{2}$ 5400 5425 590 65.4.3.2 5900 7.6.4.21 5500 2.3.4.6 5300 2.3.4.6 4530 2.4 2970 3.1 357 2.1 1457 3.1 1457	1725 4.2 1560 3.5 - 2.4 - $\pm 1.3.5.7$ (-61.8)
$\begin{bmatrix} 28.7 \\ CS \end{bmatrix}$ CS $\begin{bmatrix} 714 \\ Ba \end{bmatrix}$ Ba $\begin{bmatrix} 920 \\ La \end{bmatrix}$ La $\begin{bmatrix} 2222 \\ Hf \end{bmatrix}$ Hf $\begin{bmatrix} 2996 \\ Ta \end{bmatrix}$ Ta $\begin{bmatrix} 3410 \\ W \end{bmatrix}$ $\begin{bmatrix} 3180 \\ Be \end{bmatrix}$ Be $\begin{bmatrix} 3000 \\ OS \end{bmatrix}$ CS $\begin{bmatrix} 2454 \\ Lr \end{bmatrix}$ $\begin{bmatrix} 1769 \\ Pt \end{bmatrix}$ $\begin{bmatrix} 1063 \\ Au \end{bmatrix}$ $\begin{bmatrix} -38.4 \\ Ha \end{bmatrix}$ Hg $\begin{bmatrix} 303 \\ Ta \end{bmatrix}$ The second	327.4 Dh 271.3 Ri 254 Do (302) At (-71) Dh
1.90 Costum Barting Lathaum Hafting Tantakan Walter Walter Dealur 22.5 Cost 22.5 ii 21.4 i 19.3 A 4 13.6 ii 9 11.85 ii 1	
Cestain bandin cantantan namun inamun kenangan yang kenangan kenangan Kenangan kenangan ke	Lead Bismuth Polonium Astatine Radon
87 (223) 88 (226) 89 (227) 104 = 105 -	
Franclum Radium (Rutherfordium) Hahnium	

*	58 140.12 3468 3.4 795 Ce 6.67	59 140.907 3127 3,4 935 Pr 6.77 Preseodumium	60 144.24 3027 3 1024 7.00 Noodumium	61 (147) - 3 (1027) Pm	62 150.35 1900 3,2 1072 Sm 7.54	63 151.96 1439 826 5.26 EU 5.26	64 157.25 3000 3 1312 Gd	65 158.924 2800 3.4 1356 Tb 8.27	66 162.50 2600 3 1407 Dy 8.54	67 164.930 2600 1461 8.80 HO	68 167.26 2900 1497 9.05 Er	69 168.934 1727 3 1545 Tm 9.33	70 173.04 1427 3,2 824 Yb 6.98	71 174.97 3327 3 1652 LU 9.84	
* *	90 232.04 3850 Th	91 (231) - 5,4 (1230) Pa	92 238.036 3818 6.5.4.3 1132 U	93 (237) 6,5,4,3 637 ND	94 (242) 3235 6,5,4,3 640 Pu	95 (243) - 6,5,4,3 - Åm	96 (247) - Cm	97 (247) - Bk	98 (249) - Cf	99 (254) - Es	100 (253) - Fm	101 (256) - Md	102 (254) - NO	103 (257) 	~
	Thorium	15.4 Protactinium	19.07 Uranium	19.5 Neptunium	- Plutonium	11.7 Americium	- Curium	- Berkelium	- Californium	– Einsteinium	- Fermium	- Mendelevium	- Nobelium	- Lawrencium	

CHEMICAL PRECIPITATION

- Degree of clarification obtained depends on the amount of chemical used.
- From 80 to 90% of total suspended matter, 40 to 70% of BOD5, 30 to 60% of COD, and 80 to 90% of the bacteria can be removed.
- Plain sedimentation can only removed 50 to 70% total suspended matter and 30 to 40% organic mater.

Alum

 When alum is added to wastewater containing calcium and magnesium bicarbonate alkalinity the reaction occurs may be illustrated as follows:

 $Al_2 (SO_4)_3.18H_2O + 3Ca (HCO_3)_2 \Leftrightarrow 3CaSO_4 + 2Al(OH)_3 + 6CO_2 + 18H_2O$

Alum

 $Al_2 (SO_4)_3.18H_2O + 3Ca (HCO_3)_2 \Leftrightarrow 3CaSO_4 + 2Al(OH)_3 + 6CO_2 + 18H_2O$ (1)

- i.e.
- Aluminium sulphate (666.7) + calcium bicarbonate (3x100 as CaCO3)⇔ calcium sulphate (3x136)+aluminium hydroxide (2x78) +carbon dioxide(6x44) + water (18x18).
- The insoluble aluminium hydroxide is a gelatinous floc that settles slowly through the wastewater.

Alum

 Because alkalinity is reported as CaCO3, the molecular weight of which is 100, the quantity of alkalinity required to react with 10 mg/L of alum is:

$$10.0 \text{ mg/Lx} \frac{3 \text{ x } 100 \text{g/mol}}{666.7 \text{ g/mol}} = 4.5 \text{ mg/L}$$

• If less than this amount of alkalinity then lime is added but seldom required in the chemical treatment of wastewater.

Ferric chloride The reactions for ferric chloride are:

162.1 Fecl ₃	3x18 +3H ₂ O \Leftrightarrow	106.9 Fe(OH) ₃	$+3H^+$	+3Cl ⁻	(2)
Ferric Chloride	water	Ferric hydroxide			
3H ⁺	+3HCO ₃ ⁻ ⇔ bicarbonate	3H ₂ CO ₃ carbonic acid			

Ferric sulphate and lime



Ferrous sulphate and Lime

 In most cases ferrous sulphate cannot be used alone as a precipitant because lime must be added at the same time to form a precipitate.

The reaction of ferrous sulphate alone is shown below:

100 as CaCO₃ 178 278 136 7x18 $FeSO_4.7H_2O + Ca(HCO_3)_2 \Leftrightarrow$ $Fe(HCO_3)_2 + CaSO_4$ $7H_2O$ (4)+Calcium Calcium Ferrous Ferrous sulphate sulphate bicarbonate bicarbonate

If lime in the form of Ca (OH)2 is now added, the reaction:

2x56 as Ca O 178 89.9 2x1002x18 $Fe(HCO_3)_2 + 2Ca(OH)_2 \Leftrightarrow$ (5) $2CaCO_3$ $2H_2O$ $Fe(OH)_2$ ++Calcium Calcium Ferrous Ferrous bicarbonate hydroxide hydroxide carbonate

Ferrous hydroxide is next oxidised to ferric hydroxide, the final form desired, by oxygen dissolved in the waste water:

4x89.9322x184x106.9 $4Fe(OH)_2$ + O_2 + $2H_2O$ \Leftrightarrow $4Fe(OH)_3$ (6)FerrousOxygenFerric hydroxideFerric hydroxide

The insoluble ferric is formed as a bulky, gelatinous floc similar to the alum floc.

The alkalinity required for a 10 mg/L dosage ferrous sulphate is

$$10.0 \text{ mg/L x} \frac{100 \text{ g/mole}}{278 \text{ g/mole}} = 3.6 \text{ mg/L}$$

Lime required is $10.0 \text{ mg/L x} \frac{2 \text{ x 56 g/mole}}{278 \text{ g/mole}} = 4.0 \text{ mg/L}$

The oxygen required is $10.0 \text{ mg/L} = \frac{32 \text{ g/mole}}{4 \text{ x } 278 \text{ g/mole}} = 0.29 \text{ mg/L}$

Ferrous sulphate and Lime

- Because the formation of ferric hydroxide is dependent on the presence of dissolved oxygen, the reaction in equation 6 cannot be completed with septic waste water or industrial wastes.
- Ferric sulphate may take the place of ferrous sulphate, and its use often avoids the addition of lime and the requirement of dissolved oxygen.

Ferric chloride and Lime

The reactions for ferric chloride and lime are:

$\begin{array}{l} 2x162 \\ Fe_2 Cl_3 \end{array} + $	$3x56 \text{ as Ca O} 3Ca(OH)_2 \Leftrightarrow $	$\begin{array}{l} 3x111\\ 3CaCl_2 \end{array} + \end{array}$	2x106.9 $2Fe(OH)_3$ (7)
Ferric chloride	Calcium hydroxide	Calcium sulphate	Ferric hydroxide

Chemical Precipitation for Phosphate Removal

- Removal of phosphorous from wastewater involves the incorporation of phosphate into the suspended solids and the subsequent removal of those solids.
- Phosphorous can be incorporated into biological or chemical precipitates.
- Metal salts such as alum and ferric chloride are commonly used for the removal.
- Polymers have been also used in conjunction with iron salts and alum.

Chemical Precipitation for Phosphate Removal

- Lime is rarely used due to the problem of substantial increase in the mass of the sludge compared with metal salts. The problems are operating and maintenance associated with handling, storage and feeding of lime.
- Ferrous sulphate and ferrous chloride are also used.
- The precipitation of phosphorous from wastewater can occur in a number of different locations within the following process flow diagram.

Chemistry of Phosphate Removal:

- The quantity of lime required will depend primarily on the alkalinity of the wastewater.
- The quantity of lime required to precipitate the phosphorus in wastewater is typically about 1.4 to 1.5 times the total alkalinity expressed as CaCO3.
- High pH value is required to precipitate phosphate, co-precipitation is usually not feasible.
- Re-carbonation with carbon dioxide (CO2) is used to lower the pH value.

SENDIMENTATION BASIN

- Sedimentation the removal of solids particles from a suspension by settling under gravity.
- Clarification is a similar term which refers specifically to the function of a sedimentation tank in removing suspended matter from the water to give a clarified effluent.

- Thickening the settled impurities are concentrated & compacted on the floors of the tank & in the sludge-collecting hoppers.
- Sludge concentrated impurities withdrawn from the bottom of sedimentation tanks.
- Scum material that floats to the top of the tank.

Application of Sedimentation

- In water treatment to remove impurities that have been rendered settleable by coagulation and flocculation.
- In municipal wastewater treatment:
- main process in primary treatment.
- for removing 50 to 70% of the suspended solids.
- to remove the biological flocs produced by micro-organisms after biological treatment.

Classification of Settling Behaviour:

- is distinguished on the basis of the nature of particles to be removed and their concentration.
- Class I unhindered settling of discrete particles.
- Class II settling of dilute suspensions of flocculent particles.
- Class III hindered settling & zone settling.
- Class IV compression settling (compaction).

Ideal Settling Behaviour – Sedimentation Class I

- involving discrete particles.
- Each particles settles freely without interference from adjacent particles.
- As a particle settles, it accelerates until the drag force due to its motion is equal to the submerged weight of the particle. At this juncture the particle will have reached its terminal velocity Vp. Thus gravitational force= drag force

- Settling Particles:
- Density, $D = \rho_p$
- Volume = \forall_p
- Projected Area = A_p
- Vp = settling velocity
- W = submerged wt.
- The drag force on a particle:

•
$$D = C_D \rho_f A_p \frac{V_p^2}{2}$$
 (1)



Submerged weight of the particle:

$$W = \left(\rho_p - \rho_f \right) g \, \forall_p \tag{2}$$

Where $\rho_f =$ fluid density

By equating the above expression and after substituting for $A_p \& \forall p$;

$$V_{p} = \sqrt{\frac{4}{3} \frac{\left(\rho_{p} - \rho_{f}\right)}{\rho_{f}} \frac{gd}{C_{D}}}$$
(3)

In practice C_D is a function of Reynolds Number, R_e.
For spherical particles, it can be represented by the following eqn:

For Re<1,
$$C_D = \frac{24}{R_e}$$

For 14, $C_D = \frac{24}{R_e^{\frac{1}{2}}} + 0.34$
For 10³5, $C_D \approx 0.4$

Substituting equation (3) for Re<1, where
$$R_e = \frac{\rho_f V_p d}{\mu}$$
,

$$\therefore V_{p} = \frac{g}{18} \frac{\left(\rho - \rho_{f}\right)}{\mu} d^{2}$$
(4)

.

This is known as Stoke's Law.

In reality particle diameter & density not normally known. Particles also have irregular shape

Sendimentation Class II – Settlement of dilute suspensions of flocculation particles.

- involves a variety of particle sizes settling at different rates.
- colliding particles may coalesce to form larger aggregates with an increase settling velocity, so typical path followed by flocculent particles is curved.
- the depth should be great enough to provide the opportunity for particle agglomeration to occur.
- effects of depth & detention time on solids concentration for flocculating particles can be determined by settling tests.

Sendimentation Class III – Hindered settling and sludge blanket clarifiers.

- As the concentration of particles in suspension increased, a point is reached where particles are so close together that they no longer settle independently of one another.
- The velocity fields of the fluid displaced by settling particles
- When particles concentration is so high, the whole suspension tends to settle as a 'blanket'. This is termed as zone settling. E.g. activated sludge.



Sedimentation Class IV – Compression settling.

- As the settling particles approach the floor of the sedimentation tank, particles concentration are very high and are in contact with adjacent particles.
- Settling can only occur by adjustments within the matrix and takes place at a reducing matrix.
- Occurs as the settled solids are compressed under the weighr of overlying solids; voids are gradually diminished.
- E.g. activated sludge final settling tank.

Design Procedures:

- Determine flow rate, Q.

- Select surface loading,
$$V_s = \frac{Q}{A} = \frac{Q}{BL}$$

- For detention time,
$$t = \frac{LBD}{Q} = \frac{D}{V_s} = \frac{L}{Q/BD}$$

- For depth D;
$$D = V_s \cdot t = V_s \cdot \frac{LBD}{Q}$$

- For rectangular tank in practice length: breaddth ratio normally 4:1 or 3:1.

Guidelines for surface loading.

Types of Installation	Surface Loading (m/day)	Detention time (hr)
Small	20 to 30	3 to 4
New Technologies	30 to 40	$2\frac{1}{2}$ to $3\frac{1}{2}$
New Tech. + good operation	35 to 45	2 to 3
Large + excellent operation	40 to 60	$1\frac{1}{2}$ to $2\frac{1}{2}$

Outlet Arrangement

- IRC Technical Report No.18(1981) suggested: $L = 0.2Q/HV_s$

Where:

L = combined weir length (m)

Q =flow rate (m²/day)

H = depth of tank (m)

 V_s = settling velocity (m/day)

Adjustable V-notched weirs are convenient to ensure uniform flow throughout the collecting trough, especially low overflow rates are used



Sedimentation tank







Fig 13.4 An early type of rectangular flocculation and settling unit featuring hydraulic flocculation, horizontal flow sedimentation and intermittent sludge removal from an emptied tank



FIGURE 9-12 Rectangular clarifier. (Courtesy Envirex, a Rexnord Company.)

Multi-storey horizontal tank



FIGURE 7.18 Multistory horizontal tank with parallel flow on three levels. (*Source:* Courtesy of OTV, Paris, France, and Kubota Construction Co., Ltd., Tokyo, Japan.)



Radial flow circular Sedimentation tank



FIGURE 7.19 Circular radial-flow clarifier. (*Source:* Courtesy of Baker Process Equipment Co., Salt Lake City, Utah.)



Fig. 7.6 Lamella separator and thickener. (Passavant Ltd.)



Fig. 7.7 The 'Super Pulsator' clarifier. (Degremont Laing Ltd.)



Fig 13.5 Arrangement of hopper-bottomed 'sludge blanket' clarifier with separate sludge concentrator



Fig 13.9 Solids contact clarifier with mechanical circulation, gravity sludge return and peripheral sludge concentrator (Courtesy: Degremont)



Fig 13.8 Solids contact clarifier with mechanical circulation and sludge scraper, with central sludge withdrawal (Courtesy: William Boby - Reactivator)

Pulsator



Fig 13.7 Pulse control in a pulsator

DISSOLVED AIR FLOTATION

- Flotation processes have been used in the mining and chemical processing industries for over 100 years
- The ancient Greeks used this process to separate minerals from the gangue over 2000 years ago
- Dissolved-air flotation was patented in 1924 to Niels Peterson and Carl Sveen in Scandinavia.

DISSOLVED AIR FLOTATION

- The use of DAF in the treatment of waste water and potable water began in the late 1960's
- dissolved air flotation has been used for water clarification in Europe especially in the Scandinavian countries for more than 20 years.

DISSOLVED AIR FLOTATION

- In the United Kingdom, the first full-scale water treatment plant using this process was commissioned in 1976 at the Glendye Treatment Works of the Grampian Regional Council, Scotland
- Flotation may be defined as the transfer of a solid from the body of a liquid to the surface by means of bubble attachment
- Different methods of bubble generation give rise to different types of flotation processes.

There are three types of flotation process:

- 1. Electro- or electrolytic flotation.
- 2. Dispersed air flotation.
- 3. Dissolved air flotation.

Electro-flotation

- bubbles (oxygen and hydrogen) are generated by passing an electric current between two electrodes in a dilute aqueous solution
- The anodes are prone to corrosion while the cathodes are subject to scaling by carbonate deposition
- The bubbles produced by the electro-flotation process are normally small
- This process is suitable for effluent treatment (Ho and Chan, 1986), sludge thickening and water treatment installations of 10 to 20 m3/hour

Electro-flotation



Dispersed Air Flotation

- has two different systems to generate bubbles namely, foam flotation and froth flotation.
- In the foam flotation system, bubbles are generated by forcing the air through a porous media made out of ceramic, plastic or sintered metal
- In the froth flotation system (as shown in Figure 2.3) a high speed impeller or turbine blade rotating in the solution is used to produce air bubbles

Dispersed Air Flotation

- This process produces large air bubbles of more than 1mm in diameter
- This process has been assessed for potable water treatment but was not suitable, mainly for the separation of minerals and removal of hydrophobic materials such as fat emulsions in selected waste water treatment

Foam flotation



Froth flotation



Dissolved Air Flotation (DAF)

- Three main types of dissolved air flotation
- 1. Vacuum flotation
- 2. Micro-flotation
- 3. Pressure flotation

Vacuum Flotation

- It is used for the recovery of fibres in the pulp and paper industry
- In this process, water is saturated with air under atmospheric pressure and then a vacuum is applied to the flotation tank.

Micro-flotation

- for the treatment of domestic sewage and industrial effluents
- In the micro-flotation process, water is passed down and up a shaft of approximately 10 metres deep. The whole water column will be subjected to hydrostatic pressure. Water will be aerated as it passes the down-flow section and air dissolves in the water due to an increase in hydrostatic pressure

Pressure Flotation

- Pressure flotation is the most common process used in dissolved air flotation.
 There are three basic types of pressure dissolved air flotation processes:
- Full-flow pressure flotation
- Split-flow pressure flotation
- Recycle-flow pressure flotation

Full-flow pressure flotation (Source: Zabel, 1980)



Split-flow pressure flotation (Source: Zabel, 1980)



Recycle-flow pressure flotation (Source: Zabel, 1980)


Dissolved Air Flotation



Dissolved Air Flotation



Saturator



Typical design and operation parameters for DAF

PARAMETER	DESIGN VALUES	DESIGN VALUES
	RANGE	TYPICAL
CHEMICAL PRETREATMENT		
Coagulation dose	Determine from jar test	
Flocculation time (minutes)	5 - 30	20
G value (per second)	10 - 150	70

Typical design and operation parameters for DAF -

FLOTATION TANK DESIGN:

PARAMETER	DESIGN VALUES	DESIGN VALUES
	RANGE	TYPICAL
Detention time (minutes)	5 - 15	10
Depth (m)	1.0 - 3.2	2.4
Overflow rate (m/hr.)	5 - 15	8
Freeboard (m)	0.1 - 0.4	0.3

Typical design and operation parameters for DAF - AIR

SATURATION SYSTEM AND **SLUDGE**:

PARAMETER	DESIGN VALUES	DESIGN VALUES	
	RANGE	TYPICAL	
Operating pressure (KPa)	350 – 620 (50 - 90psi)	485 (70psi)	
Recycle ratio (%)	6 - 30	6 - 12	
Bubble size (µm)	10 - 120	40 - 50	
Saturator efficiency, packed (%)		90	
Saturator efficiency, unpacked (%)		70	
SLUDGE:			
Percent solid (%)	0.2 - 6	3	

DAF performance related to water quality characteristics

PARAMETERS **FINDINGS** 1.Algae Removal (a) Better than sedimentation (b) Alum performed better than ferric sulphate (c) 95% removal compared with 65% from sedimentation 2 (a) DAF performed same as sedimentation Colour and turbidity removal (b) Colour of 45^o Hazen and turbidity less than 100 FTU is limiting point for DAF (c) DAF removed 70% of turbidity and 79% of suspended solids (d) Removal of true colour at the same pH and coagulant dose, DAF performed same as sedimentation (e) Not much different in result using ferric sulphate or alum (f) Turbidity removal not effective in extreme winter conditions. (g) DAF removed 30 to 45.7% of turbidity and 40.8 to 50.5% of colour 3. Residual coagulant (a) DAF performed the same as sedimentation concentration 4. Bacteria (a) 99% could be removed 5. Protozoa (a) Cryptosporidium removal more effective using DAF compared with sedimentation

DAF performance related to process parameters

PARAMETERS	FINDINGS
1. Performance at	(a) Dependent on the efficiency of flocculation
flotation stage	
2. Velocity gradient in	(a) Non-tapering:
flocculation	(i) 70 per sec.
	(ii) 60 to 80 per sec.
	(b) Tapered (25 to 80 per sec.)
3. Retention times	(a) In flash mixer:
	1 1/2 minutes
	(b) In flocculation tank:
	7 min.;9 min.;12 min.;8 min.;20 to 127 min.;5 to 120 min.
	(c) In flotation tank
	20 min.,5 1/2 min.
4. Overflow rate	(a) 2 to 4 mm/sec by flotation and 0.3 to 1.3 mm/sec by sedimentation
	(b) DAF:Sedimentation by the ratio 6:1 to 12:1

Theory of Flotation

• The principles of flotation based on bubble generation, bubble attachment and solid separation.

For bubble-particle attachment several theories were forwarded by various researchers:

- Bubbles grow by precipitation from a supersaturated solution on the surface of the particles (Taggart, 1945)
- Bubble-particle attachment involves chemical and physical aspects such as surface energy, surface tension, adsorption, contact angle, polarity, surface reactivity, surface condition and adding air-adhering agents to some minerals (Gaudin, 1939)

-----for bubble-particle attachment several theories were forwarded by various researchers:

- An 'induction time' is required to allow the thin liquid layer between bubble and particles to drain away so that coalescence can occur (Sutherland, 1948)
- If the surface conditions of the particles are appropriate, the bubbles will collide and coalesce with the particles (Klassen and Mokrousov, 1963)
- The particle follows its trajectory as it approaches a bubble and this trajectory will be dependent on viscous effects and Reynolds number (Flint and Howarth, 1971)

The following mechanisms are important for the bubble attachment process to take place:

- Adhesion of gas bubbles with the suspended phase as a result of collisions or by nucleation.
- 2. The trapping of gas bubbles in a floc structure
- 3. The absorption of gas bubbles into a floc structure as it is formed

Solubility of air in water (Henry's Law)

SOLUBILITY OF AIR IN WATER



Filtration

- To remove fine inorganic & organic particles from the water.
- Sand filter for water supply.
- Vacuum filters & filter press are used for removing water from sludge.
- Diatomaceous earth filters for municipal water supplies.

Mechanical Straining

- Granular filter remove particles much smaller than the dimensions of the interstices between their grains.
- When larger particles of impurities lodged in a space between two grains, the portion of opening not occupied by the particles will provide smaller opening to trap smaller particles. If this is the case, the surface layers would rapidly become clogged and fine particles would be likely to pass through the remaining depth without being caught or detained.
- However granular filter capable of capturing very fine particles.
- Mechanical straining account for only a minor part.

Adsorption

- Three essential parts in filtration:
- the bed fixed solid.
- the water passing through the interstices between the fixed solid.
- the solid impurities suspended in the water.

- -Adsorption of particles of impurities onto the fixed bed is a major factor in a successful filtration through porous media.
- -Efficiency depends on surface properties of both adsorbing matrix & small particles.
- Small particle in close proximity to a solid surface is subject to electrical attraction or repulsion & to the attraction caused by Van der Waals forces.
- -it is also subject to hydraulic forces resulting from the movement of water.

Transport

- Flow is laminar
- Three mechanism by which particles are transported into contact with the filter matrix:
- 1. Interception
- 2. Sedimentation
- 3. Diffusion



Interception – particle A streamline for A to A'. Sedimentation – particle B is deflected from streamline by gravity. Diffusion – particle C is buffeted by Brownian motion.

Interception & sedimentation is effective as the particle size increases. Effectiveness of diffusion increases with a decrease in size.

Types of filters

- Slow sand filter
- Rapid sand filter
- Pressure filter
- Roughing filter used at preliminary stage

Slow sand filter



Fig 13.14 Diagrammatic section of slow sand filter

History for slow sand filter

- 1804 filter was tested for industrial used in Scotland
- 1829 first filtration for public water supply in London
- 1852 filtration was required in accordance to water act in London for surface water supplies.
- 1892 cholera epidemic in Hamburg, 7582 of the 580,000 population died. Water supply for Hamburg was abstracted from River Elbe, but treatment was only by sedimentation. Nearest town called Altona using the same source of water but was filtered before supplied to the consumers and not record of death.

River Elbe, Dresden



 Altona is part of Hamburg, N Germany, a port on the Elbe River. Its manufactures include chemicals, textiles, and tobacco products. There are fisheries, and the district is a rail center. Founded as a fishing village in the 16th cent. and later one of the first free ports in N Europe, Altona was incorporated into Hamburg in 1937





Mechanisms of slow sand filtration

- Mechanical straining
- Biological activity
- Although gravitational settling and adsorption are of importance. Other mechanisms such as inertial and centrifugal forces, mass attraction, electrostatic are of lesser importance.

 As a clean filter is returned to operation following cleaning the alluvial mud, algal, planktonic and protozoan matters are strained out to form a layer on top of the sand and also intimately mixed with the topmost grains. This is the filter skin or 'schmutzdecke' which acts both as a very fine filter coat and as an active biological zone. In this schmutzdecke, and at the top of the sand layer, most of the suspended solids in the water are filtered out.

 Pathogenic bacteria adsorbed on to the sand or trapped in the schmutzdecke are a prev adverse conditions and to predators ranging from bacteriophages to protozoa. Very few pathogenic bacteria escape through a well designed and operated slow sand filter and removal can be expected to be to greater than 99.9%. Below the 'schmutzdecke' bacterial activity continues for about 400mm

Comparison of slow sand filter with rapid sand filter

Parameter	Slow sand filter	Rapid sand filter	
Filtration rate (m/hr)	0.1 - 0.4	5 - 20	
Effective size of sand	0.15 - 0.4	0.6 - 1.0	
(mm)			
Non-uniformity coefficient	1.7 - 3.0	1.3 - 1.5	
Penetration of suspended	Superficial	Deep	
solids			
Method of cleaning	Scraping off surface layer	Upward flow or	
		backwashing fluidised the	
		bed	
Time between cleaning	30 - 150	1 - 3	
(day)			
Inlet water requirement	Low turbidity < 50NTU	Coagulation, flocculation	
		& sedimentation	
Basic process	biologi	fizikal	
Economical aspect	Unskilled labours	Complicated design with	
		complex equipment	
Water loss	Water wastage during	1 to 4 to 6% of filtered	
	maturation	water	

Design criteria for rectangular slow sand filter unit

- Total depth 2.5 to 3.0 m.
- constructed from reinforced concrete, 'ferrocement', stones or bricks and must not be cracked or leaked
- Plan area up to 5000 m².
- At least 2 filters in parallel so that filter can be operated if the other needs to be cleaned.
- Depth of chamber will be as follows ukur (Schulz and Okun, 1984):

 Free board 		0.20 m
 supernatant water 		1.00 m
 Initial filter media 		1.00 m
 Supporting gravel 		0.30 m
 Underdrained brickworks 		<u>0.20 m</u>
_	Total	2.70 m

Slow sand filter



A= inlet valve, B= supernatant drain valve, C= backfilling valve, D= drained off valve, E=regulating valve, F=drain valve, G= treated water control valve, H= Outlet weir and I= flow calibration system.

Rapid sand filter



Fig. 7.8 Rapid gravity filter. (Paterson Candy International Ltd.)

Rapid sand filter



Fig 13.15 General arrangement of a typical rapid sand filter

Rapid sand filter

- Higher filtration rate of 3 to 6 m/hour compared to slow sand filter
- Filter media consist of:
- Upper sand layer from 0.45 0.95 mm in size, with depth between 0.6 – 1.0 m
- Lower layer of gravel with from 3 50 mm with depth between 0.1 – 0.45 m

Properties required for medium

- From "Specification for Granular Filter Media" by K J Ives in Effluent and Water Treatment Journal, June 1975 pp 296-305.
- (a) DURABILITY resistance to abrasion during transport, laying and cleaning. 1% loss per year is a suggested criterion.
- (b) INSOLUBILITY:
- (i) Material should not be lost by dissolving.
- (ii) Toxic substance must not be leached out of medium.
- Test for (i) by immersing in acid and for (ii) by soaking in distilled water and testing for any likely dissolved contaminants.
- c) DENSITY Is there a tendency to float?
- Can the material be fluidised for back-washing?
- d) SHAPE, sphericity $\psi = d_h/d_s$
- Where d_h is the hydraulic diameter, being the size of a spherenof equal settling velocity to the grain and ds is the mean sieve size.
- Angular particles (low sphericity) are most effective than spherical, but particles for which ψ <0.6 are usually too flat, and cause high headlosses.

•	Value of ψ : sphere	1.00
•	British filter sand	0.85
•	British anthacite	0.7

- e) PREPARATION Media should be washed and be sieved to a fairly uniform size.
- f) ABILITY TO RETAIN SOLIDS. The surface chemistry should be tested with a small labratory filter, using the water that the full-scale plant is required to treat.

- Washing will be made when head loss more than 2.5 – 3 m
- Water for backwash is obtained from filtered water.
- Washing time is between 10 -15 minutes for every 24 – 48 hours.
- 3 types of backwashing: using water only, air followed with water, and finally combination of air and water.
- For backwashing air followed with water, the air flow rate used is at 27 – 35 m³/hr/m² for 4 minutes and water flow rate is at 45 m³/hr/m².

Pressure filter

- Operate at pressure higher than atmospheric thus may required water to be pumped under pressure into the filter if gravitational pressure is low.
- Normally in cylindrical form.
- For practical purpose, maximum diameter for vertical filter is limited to 2.75m and for a horizontal filter with the same diameter with a length of 12m.
- Filtration rate up to 15 m3/m2 per hour.
- Pressure acting on the filter normally less than 80 m head of water.

Pressure filter



Fig. 7.10 Sectional view of a pressure filter. (Paterson Candy International Ltd.)

ADSORPTION

- Process of collecting soluble substances that are in solution on a suitable interface, between the liquid and a gas, a solid, or another liquid.
- Adsorption is used at the air-liquid interface in the flotation process.
- Only the case of adsorption at the liquidsolid interface will be considered in this discussion.

-----ADSORPTION

- Activated-carbon treatment of wastewater is usually thought of as a polishing process for water that has already received normal biological treatment.
- Carbon is used to remove a portion of the remaining dissolved organic matter.

Activated-Carbon and Its Use

- Prepared by first making a char from materials such as almond, coconut, and walnut hulls, other woods, and coal.
- The char is produced by heating the material to a red heat in a retort to drive off the hydrocarbons but with an insufficient supply of air to sustain combustion.
- The char particle is then activated by exposure to an oxidizing gas at high temperature. This gas develops a porous structure in the char and thus creates a large internal surface area. The surface properties that result are a function of both the initial material used and the exact preparation procedure.

-----Activated-Carbon and Its Use

- After activation, the carbon can be separated into or prepared in different sizes with different adsorption capacities.
- The two size classifications are powdered, which has a diameter of less than 200 mesh, and granular, which has a diameter greater than 0.1 mm.
- A fixed-bed column is often used as a means of contacting wastewater with Granular activated carbon (GAC).
- The water is applied to the top of the column and withdrawn at the bottom. The carbon is held in place with an under-drain system at the bottom of the column.

fixed-bed column



-----Activated-Carbon and Its Use

- Provision for backwashing and surface washing is usually necessary to limit the head loss buildup due to the removal of particulate material within the carbon column. Fixed-bed columns can be operated singly, in series, or in parallel.
- Expanded-bed and moving-bed carbon contactors have also been developed to overcome the problems associated with head loss. The influent is introduced at the bottom of the column and is allowed to expand, much as a filter bed expands during backwash.

-----Activated-Carbon and Its Use

 An alternative means of application is that of adding powdered activated carbon (PAC). Powdered activated carbon has been added to the effluent from biological treatment processes, directly to the various biological treatment processes, and in physical-chemical treatment process-flow diagrams. The addition of PAC directly in the aeration basin of an activatedsludge treatment process has proved to be effective in the removal of a number of soluble refractory organics.

Analysis of the Adsorption Process

- Takes place in three steps: macro-transport, micro-transport, and sorption.
- Macro-transport involves the movement of the organic material through the water to the liquid-solid interface by advection and diffusion.
- Micro-transport involves the diffusion of the organic material through the macro-pore system of the GAC to the adsorption sites in the micropores and submicropores of the GAC granule. Adsorption occurs on the surface of the granule and in the macropores and mesopores.

-----Analysis of the Adsorption Process

- Sorption is the term used to describe the attachment of the organic material to the GAC. The term *sorption* is used because it is difficult to differentiate between chemical and physical adsorption.
- The theoretical adsorption capacity of the carbon for a particular contaminant can be determined by calculating its adsorption isotherm.
- The quantity of adsorbate that can be taken up by an adsorbent is a function of both the characteristics and concentration of adsorbate and the temperature.

-----Analysis of the Adsorption Process

- Generally, the amount of material adsorbed is determined as a function of the concentration at a constant temperature, and the resulting function is called an *adsorption isotherm*.
- Equations that are often used to describe the experimental isotherm data were developed by Freundlich, by Langmuir, and by Brunauer, Emmet, and Teller (BET isotherm).
- Of the three, the **Freundlich isotherm** is used most commonly to describe the adsorption characteristics of the activated carbon used in water and wastewater treatment.

Freundlich Isotherm. The empirically derived Freundlich isotherm is defined as follows:

$$\frac{x}{m} = K_f C_e^{1/n} \tag{1}$$

where, x/m = amount adsorbate adsorbed per unit weight of absorbent (carbon)

Ce =equilibrium concentration of adsorbate in solution after adsorption

 K_f , n = empirical constants.

-----Freundlich Isotherm

The constants in the Freundlich isotherm can be determined by plotting (x/m) versus *C* and making use of Eq. 1 rewritten as:

$$\log\left(\frac{x}{m}\right) = \log K_f + \frac{1}{n}\log C_e \tag{2}$$

Langmuir Isotherm. Derived from rational considerations, the Langmuir adsorption isotherm is defined as:

$$\frac{x}{m} = \frac{abC_e}{1+bC_e}$$
(3)

where x/m = amount adsorbed per unit weight of adsorbent (carbon) a, b = empirical constants Ce = equilibrium concentration of adsorbate in solution after adsorption

-----Langmuir Isotherm

- Isotherm was developed by assuming that (I) a fixed number of accessible sites are available on the adsorbent surface, all of which have the same energy and that (2) adsorption is reversible.
- Equilibrium is reached when the rate of adsorption of molecules onto the surface is the same as the rate of desorption of molecules from the surface.

-----Langmuir Isotherm

- The rate at which adsorption proceeds, is proportional to the driving force, which is the difference between the amount adsorbed at a particular concentration and the amount that can be adsorbed at that concentration. At the equilibrium concentration, this difference is zero.
- Correspondence of experimental data to the Langmuir equation does not mean that the stated assumptions are valid for the particular system being studied because departures from the assumptions can have a canceling effect.

-----Langmuir Isotherm

The constants in the Langmuir isotherm can be determined by plotting C/(x/m) versus C and making use of Eq. 3 rewritten as

$$\frac{C_e}{\left(x/m\right)} = \frac{1}{ab} + \frac{1}{a}C_e \tag{4}$$

Just Say No To Bottled Water!

Bottled water is over priced, over 95% of cost is bottle, label, lid and transportation.

Bottled water is virtually unregulated, industry lobbyists fight against Federal Purity Regulations every year in Washington... Why?

Bottled water is bad for the environment, Millions of plastic bottles pollute our landfills daily.

Any time water is stored for a prolonged period of time in a plastic bottle, it will take on traces of the chemicals used in the plastic such as Pthylate.

According to the FDA, "Companies that market bottled water as being safer than tap water are defrauding the American public..."

Home water filtration offers significantly higher quality water, at a fraction of the cost and far more convenient than bottled water.

Thank you

