



Waste Water Treatment Plant And Leachate Chemical Effect

Ahmad Samadi, Engineering Building & Infrastructure, contact@ebni.com.au
Dep. of Civil and Environmental Engineering, Western Sydney University,
Hawkesbury, SYD, NSW, 2753. 21/08/2025

Abstract

A wastewater treatment plant comprises three treatment cycles. Although it may appear complex, it essentially involves separating solids from liquids through primary treatment, which includes a step screen to remove toilet paper, followed by a vortex grit chamber to extract heavy grits in order to achieve an adequate flow splitter. Human waste proceeds to a sedimentation tank where heavy solids are transferred to tertiary treatment and liquids to the bioreactor and clarifier, as well as IDAL (Intermittently decanted aerated lagoons) for further clarification of the water, allowing solids to settle in the equalisation basin before passing through a mixing chamber and dual media filter for purification and subsequent disinfection via chlorination. The tertiary treatment of dissolved air flotation is more aerated through diffusion than the IDAL in order to sediment the solids, in contrast to the sludge cake, which is collected from the top of the dissolved air flotation tank. The sludge return is utilised in the aerobic digester as it is essential for the chemical reaction to remove or reduce the biomass to dry centrifuge the solids for safe fertilisers. The following research acts as constructability advice for wastewater treatment utility. However, the liquid substances that drain from heaped debris or land are known as leachate in the context of a landfill. Derivative from the material it has traversed, leachate contains elevated concentrations of offensive components. As secondary products of the solid organic decomposition processes, landfill leachates are also generated. The four primary categories of constituents that are the most hazardous products of this leachate are soluble organic matters, inorganic components, heavy metals, and xenobiotic organic compounds.

1. Introduction

Professionals who generate effluent of superior quality currently operate an extensive array of treatment facilities worldwide. There are four fundamental categories of highly effective therapeutic technologies that are presented: (Metcalf et al. 1991) demonstrated the recycling and integrated treatment with household wastewater to transfer leachate, (Serdarevic 2018) proved the biodegradation using aerobic and anaerobic methods, and (Abdel-Shafy & Mansour 2022) physical and chemical processes to transfer contaminants from municipal solid waste to water. Together with the compacted waste layers, the organic

biodegradable components in MSWs create an anaerobic environment in the landfill researched by (Abdel-Shafy & Mansour 2022). Consequently, the leachates produced by various landfills have compositions that are virtually identical to one another (Serdarevic 2016). Most landfills receive a large volume of mixed industrial, commercial, and municipal waste that is all dumped in one location (Abdel-Shafy & Mansour 2018). Each one of them contributes to the formation of substantial quantities of leachates and the contamination of groundwater. There will be a fluctuating plume of contaminants if groundwater is contaminated by a landfill. Wells in such plumes will undoubtedly be contaminated. As a result, other wells that are outside the plume—including those near the landfill—might not be impacted. Prolonged droughts and intermittent heavy rainfall promote the dispersion of leachate in nearby areas, resulting in significant surface and subterranean contamination. Consequently, membrane technologies, air stripping, adsorption, coagulation/flocculation, sedimentation/flocculation, chemical oxidation, and chemical precipitation all affect groundwater quality. It is likely that heavy precipitation recharge improves the leachate contamination potential during the post-monsoon season. An essential environmental issue is the production of leachate. Numerous factors influence the formation of varying quantities and qualities of leachate from landfills. The parameters include annual precipitation, runoff, infiltration, evaporation, transpiration, ambient temperature, trash composition and density, starting moisture content, and landfill depth. (Abdel-Shafy et al. 2024) asserted the stabilisation of solid waste in a sanitary landfill and the quality of leachate are primarily the outcomes of physical, chemical, and biological processes. Both soluble mineral compounds and a very high organic component are among the highly heterogeneous variety of materials typically found in landfills. Applying the proper wastewater technology is necessary based on the substances present. Prior to contemplating the design of any leachate management system, it is essential to identify the objectives to be accomplished. Examples include the regulation of the type and quantity of waste materials. The purpose of this research is to provide an overview and shed light on the waste water treatment plant stages and the effect of the collected solid waste and its composition of landfill leachate, which is produced from nonhazardous commercial, industrial, and residential solid waste. The features, stabilisation, and environmental impact of landfill leachate are further topics for discussion. Land fill leachate treatment is also included in the review, including various membrane technologies, physical, chemical, and adsorption treatment systems, biological treatment, and conventional

treatment

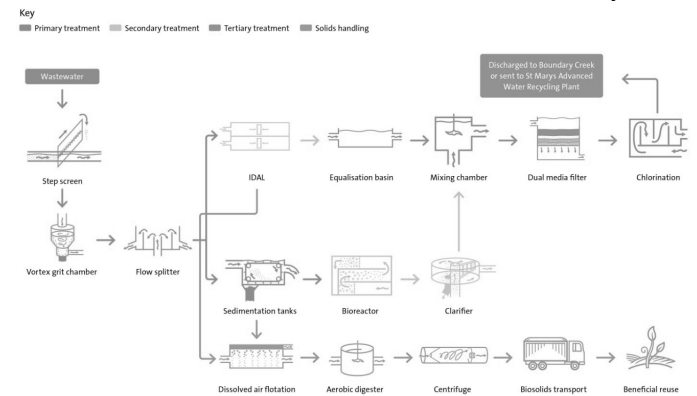


Figure 1: Steps illustrating wastewater treatment process(Source: [Sydney Water](#)).

2. Step Screening

High separation efficiencies to remove non degradable objects floating from sewerage inlets at certain wastewater flow rate achieved with economical screening separation and transportation, the screens mechanically raises and collects non-biodegradable waste about 10 mm in diameter objects such as toilet paper.

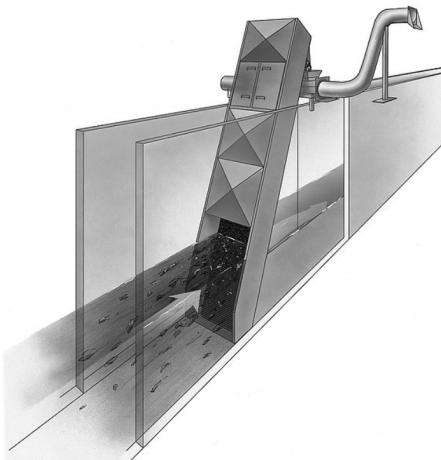


Figure 2: Simplified diagram of step screen

systems.

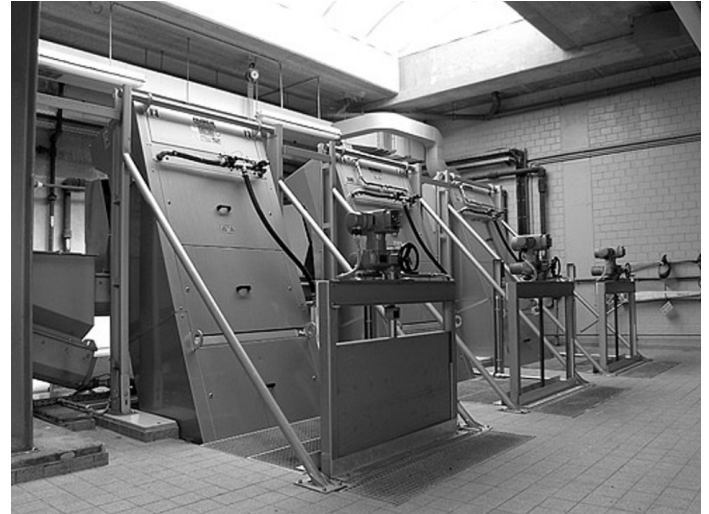


Figure 3:Actual step screen(Anon n.d.)

The step system's functionality, user-friendly operation, self-cleaning effect based on countercurrent principle, ease of maintenance, capacity for extremely high screening volumes, and dependability under operation are all factors in its success and widespread adoption. These days, a large number of treatment facilities worldwide are run by professionals who produce high-quality wastewater. A number of highly effective therapeutic technologies are presented, including four fundamental groups: (1) Transfer of leachate: based on recycling and integrated treatment with household wastewater, (2) biological treatment: biodegradation using both aerobic and anaerobic methods; (3) physical and chemical processes: coagulation/flocculation, sedimentation/flocculation, air stripping, adsorption, chemical oxidation, chemical precipitation, and membrane technologies.

2.1 Landfill leachate due to the effect of collected waste

The liquid that transports suspended solids and any other pollutants through the environment is known as leachate. It travels through materials, matters, extracts, or soluble liquids. In engineering and environmental sciences, the term "leachate" is frequently used (Abdel-Shafy & Mansour 2018; Ahn et al. 2002).

Leachate is an extremely repulsive dark liquid that is created when solid waste is disposed of in an aerobic setting (Abdel-Shafy & Mansour 2018; Rettenberg 2006). The term is defined as hazardous dissolved substances that are buried and circulate in the soil of the environment. Landfilling of putrescible, industrial, or municipal solid waste is where the term is most often employed, though. As a result, a leachate is a liquid that has the capacity to travel through soil, geological formations, matter, and groundwater in the form of liquid or extracts. It consists of soluble, suspended solids, and/or any other component of the material it has traversed.

The liquid materials that drain from accumulated material or land are referred to as leachate in the context of the landfill environment (Costa et al. 2019). As a result, the disagreeable, unwanted material that leachate includes in substantially higher proportions is primarily derived from the material it came into contact with and passed through (Costa et al. 2019). In the limited environmental context, the term "leachate" refers to liquid materials or substances that are ejected, drained, or leached from stockpiled material or land. Consequently, the leachates in question are distinguished by their considerably elevated concentrations of offensive, undesirable materials that are either derived from the stockpiled material, soil, or have passed through (Teng et al. 2021).

The primary source of liquid leachate is the municipal solid waste landfills (MSWLFs), which generate substantial quantities of this highly contaminated, black, and offensive liquid. Nonetheless, certain leachates arise directly from the solid waste that has been disposed of, attributed to the presence of moisture, other decomposing materials, and the breakdown of municipal solid refuse. But a lot of it could come from various surface water and/or runoff that seeps through the landfill and percolates down to the solid waste materials (Chen et al. 2019). When waste materials come into direct contact with surface water and/or runoff, the resulting leachate becomes severely and heavily contaminated. The latter seeps out of the landfill, where extensive environmental contamination takes place, and/or migrates concurrently into groundwater (Costa et al. 2019; Abdel-Shafy & Kamel 2016). As a result, environmental contaminants are released into the environment in greater quantities as water passes through landfill sites. The characteristics and quality of the leachate from the MSWLF exhibit significant variability. Compared to industrial wastes or raw sewage water, it typically has more pollutants (Alfaia et al. 2017).

2.2 Characteristics of landfill leachate

Heavy metals, dissolved organic matter, inorganic macrocomponents, and xenobiotic organic compounds are the four primary contaminants found in landfill leachate, which is defined as a potent wastewater ((Ed.). 2012). Groundwater and surface water contamination is the landfill leachate's biggest environmental hazard. The poisoning of groundwater by landfill leachate represents the most significant environmental consequence. An essential component contributing to this pollution is rainfall, which disperses the contaminated water both vertically and laterally, as illustrated in the figure below. This is mostly because leachate collection systems and/or well-defined geometric liners were not included in the design and construction of the majority of landfills (Abdel-Shafy & Kamel 2016).

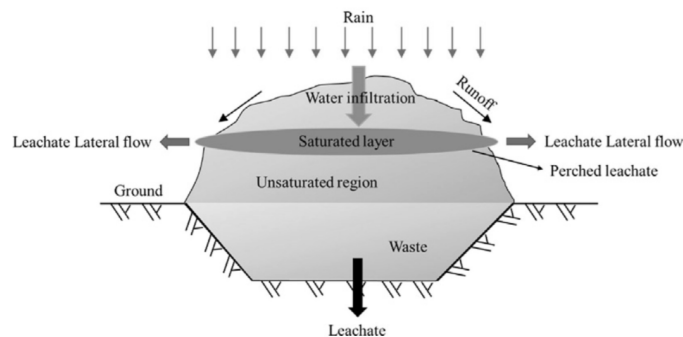


Figure 3: Effect of rainfall on landfill contamination.

Recently, numerous countries enacted regulations mandating the implementation of appropriate engineering designs for the installation of liners as a leachate collection system, together with a plan for leachate treatment. Landfill leachate has been documented to contaminate surface water and groundwater ((Ed.). 2012). The oxygen depletion of certain bodies is one of the main consequences of leachates on surface water. As a result, significant alterations occurred, including the composition of benthic fauna and flora, along with the toxicity levels of ammonia (Abdel-Shafy & Aly 2002; Abdel-Shafy & Mansour 2013).

In order to make long-term predictions about the environmental effects of landfills, it is critical to comprehend the composition of the leachate (Wang et al. 2002). If a landfill ceases to accept additional garbage and a final cover is applied, the waste will persist in decomposing and leachate will continue to be produced. However, following the final cover placement on the landfill, there is a considerable drop in leachate output. In this regard, little information is available regarding leachate production over the long period (Singh et al. 2007). The environmental impact will also be lessened by a landfill cover's long-term durability. In the United States, landfills are subject to a monitoring period of 30 years following their closure. It is anticipated that the landfill would be stable after this 30-year period. There is no longer a necessity for intensive monitoring. Nonetheless, in certain instances, the suggested 30-year monitoring duration for lined landfills may be inadequate for long-term assessment. It is essential to predict the composition of leachate that will be emitted into the surrounding environment (Serdarevic 2016). In particular, the landfill that accepts a mixture of municipal garbage together with the non-hazardous of both commercial and industrial wastes must assess the long-term composition of a specific leachate following the complete degradation of the refuse (Abdel-Shafy & Mansour 2018). under addition to the landfill that accepts certain burned residues along with the other inorganic wastes, leachates are also discharged from landfills that are kept under aerobic conditions.

3. Grit Removal

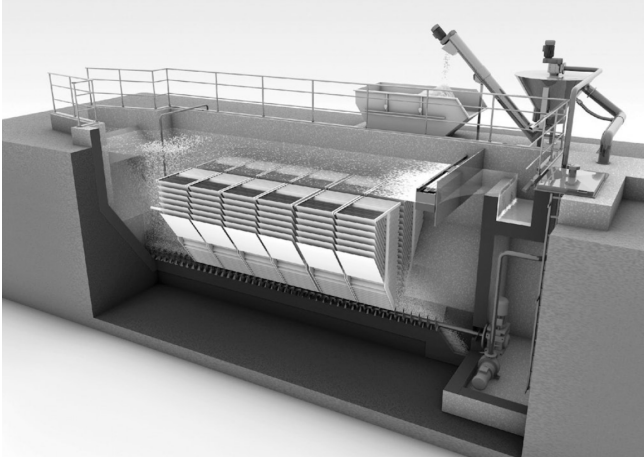


Figure 4: schematic sketch of grit collector (Operator 2019)

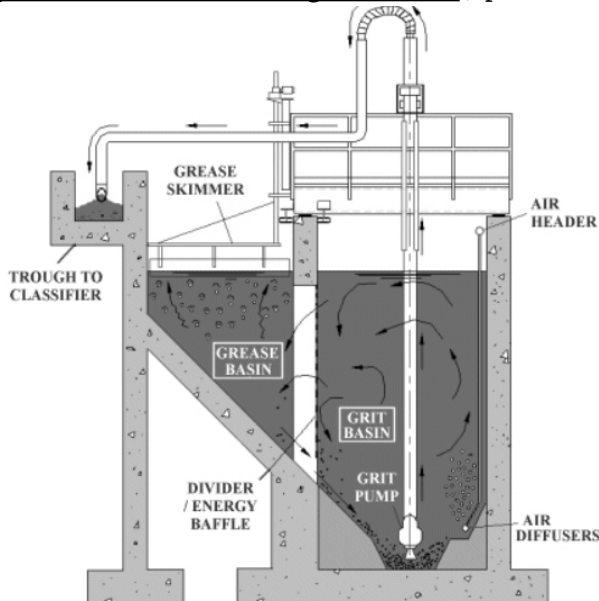


Figure 5: Cross section of grit collector through aerated chamber (Anon 2016)

An aerated chamber collects the Grit catch in a system, whereas an unaerated chamber uses a laminal separator to catch grit.

The design of vortex grit basins involves multiple parameters as shown in figure 6 that directly influence performance and operational efficiency. Key factors include floor slope, rotating impellers, grit hopper configuration, and inlet/outlet design. Each parameter contributes differently to the flow dynamics, grit removal efficiency, and maintenance requirements.

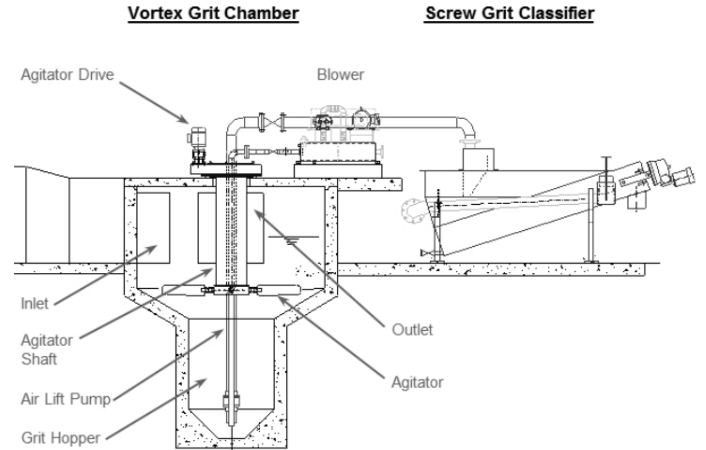


Figure 6 : Drawing of the vortex grit chamber (Anon n.d.).

3.1. Floor Slope

Floor slope plays a crucial role in determining the flow regime within the basin. A flat floor can induce a toroidal (doughnut-shaped) flow pattern, characterised by downward flow along the outer edges, inward movement along the floor, and upward flow at the centre. This pattern helps direct grit toward the centre for collection in the hopper. However, higher upward velocities associated with the toroidal flow increase the risk of grit particles being carried into the effluent, reducing removal efficiency. Recent CFD (Computational Fluid Dynamics) analyses and field observations, such as those by Chien et al. (2010), show no consistent evidence of toroidal flow patterns in full-scale basins, especially with flat-floor designs. In fact, flat floors often lead to the accumulation of grit, particularly under low-flow conditions, which can later resuspend during peak flows and overload grit-handling systems. To address this, earlier experiments (e.g., APWA, 1974) involved converting flat floors to sloped floors to prevent such accumulation.

3.2. Rotating Impeller

Rotating impellers are conventionally used to maintain heavy organics in suspension while allowing finer grit to settle. However, settling velocities for organic solids and fine grit often overlap, which complicates the optimization of impeller speed. One solution is to design the basin for optimal removal of grit based on particle size and density, supplemented by grit washing equipment to handle organics captured in the underflow. CFD studies by (Chien et al. 2010) suggest that impellers may induce a reverse toroidal flow pattern in lab-scale basins, where flow moves outward along the floor and upward along the walls. This reverse pattern could inhibit effective grit capture. However, the impact of impellers on flow patterns diminishes in full-scale basins, where other design elements dominate flow behaviour.

3.3. Grit Hopper Design

A grit hopper provides a dedicated zone for settled grit to accumulate, isolated from the influent flow and impeller influences. However, grit accumulation in the hopper can cause blockages in the suction line of the grit pump, especially if intermittent pumping is employed. Proper design and operational strategies are essential to minimise these risks and maintain system reliability.

These design considerations highlight the complexity of vortex grit basin performance, emphasising the need for a balance between flow dynamics, grit removal efficiency, and maintenance. Analytical tools like CFD and empirical field studies provide valuable insights for optimising these systems.

Through the introduction of air along one side of the grit chamber, a spiral velocity pattern that is perpendicular to the flow through the tank is created. Lighter organic particles are suspended and eventually carried out of the tank, whereas heavier particles accelerate, diverge from the streamlines, and sink to the bottom of the tank. For the Removal of Grit While effluent exits the tank at the top, grit settles by gravity into the bottom of the tank (in a grit feeder). A grit pump or an air lift pump can be used to remove the grit that collects in the grit hopper. The procedure used to eliminate grit, silt, and sand from wastewater is called grit removal.

3.4. Numerical review over vortex operation

A study was conducted by (Pretorius 2012) and it mentioned that solid information on removal of grit in vortex grit basins is surprisingly rare owing to the difficulties in sampling and the low priority frequently set on the task. To find a common separation process in vortex grit basins, a review of full-scale and experimental plant data, mathematical models, and CFD model findings were conducted. The development of a model or equation for vortex grit basin design would result from the discovery of such a mechanism. Grit basin analysis demonstrates the absence of complex flow patterns and the insignificance of centrifugal forces. In vortex grit basins, the primary process for grit removal is sedimentation under the effect of gravity.

Table 1: Specific gravity of material that makes up grit

| Material | SG | Source |
|----------------------|----------|---------------------------|
| Sand | 1.52 | (Lindeburg 2005) |
| Gravel | 2.65 | (Schmidt et al. 1997) |
| Quartz | 2.64 | (Incropera & DeWitt 1990) |
| Concrete | 2.30 | (Incropera & DeWitt 1990) |
| Cement | 3.13 | (Lindeburg 1999) |
| Aggregate | 2.64 | (Lindeburg 1999) |
| Eggshells | 2.53 | (Tsai et al. 2006) |
| Bone, rat | 2.0-2.25 | (Repo et al. 1988) |
| Coffee grounds (dry) | 0.65 | (Horio et al. 2009) |

(Wilson et al. 2007) conducted a comparison between the theoretical velocities for $SG = 2.65$ and the settling velocities of collected grit. It was demonstrated that the larger grit particles (those larger than $\sim 125 \mu m$) sank at a nearly constant speed. The suggestion is that grit particles with a higher specific gravity, which settle more quickly than $125 \mu m$ "sand" particles ($SG = 2.65$), are not found in the effluent direction of wastewater treatment plants. Instead, they build up in the collecting system and are eventually removed at a peak flow event.

Since the early 20th century, vortex grit basins have been employed for removing grit from wastewater (Pretorius 2012). These basins serve dual purposes: as grit removal systems in wastewater treatment and as devices for separating suspended solids in combined sewer overflows. The consequences of inadequate grit removal include damage to downstream equipment and operational interruptions in downstream processing units, such as aeration basins and digesters.

Therefore, accurately evaluating the capacity of vortex grit basins is crucial for efficient plant operations. Vortex grit basins offer several benefits, including minimal power consumption, reduced odour potential (particularly when compared to aerated grit chambers), low head loss, and space efficiency. Consequently, these basins are often the preferred choice, especially for large-scale installations.

To calculate the terminal settling velocity of a grit particle, Stokes' Law is often used:

$$(1) u_p = d_p^2 g (\rho_p - \rho) / (18\mu)$$

u_p = terminal settling velocity of grid plate (m/s)

d_p = particle diameter (m)

g = acceleration due to gravity (9.8 m/s^2)

ρ_p = the particle density (Kg/m^3) ρ = fluid density (Kg/m^3)

μ = fluid viscosity (Kg/m.s). (2) $Re_p = \rho \cdot u_p \cdot d_p / \mu$ since

Re_p = Reynolds number < 0.2 (Richardson et al. 2014)

For Re_p = Reynolds number > 0.2 and $1,000$ which applies

for most grit particles.

$$(3) u_p = d_p^2 g (\rho_p - \rho) / \left(18\mu \left[1 + Re_p^{0.687} \right] \right)$$

(4) $\eta = f(Q/d^n)$ η = removal efficiency

Q = flow rate m^3/s or mgd d = basin diameter in m.

(5) $Sf = s/SSf$ = sphericity

s = surface area of a sphere having the same volume as the particle (m^2)

S = Surface area of the particle (m^2) (6) $\eta = \frac{1 - \exp[\alpha(\phi - 1)]}{\frac{1}{\phi} - \exp[\alpha(\phi - 1)]}$

$\phi = u_p / \text{SOR}$ the ratio of terminal grit particle settling velocity to surface overflow rate

R = Volumetric ratio of underflow to influent flow

$\alpha = 54$ *pelect number* the tangential velocity at the edge of the basin, which seems to be observed and differs from the inflow velocity. A non-measurable approach to α prediction would be necessary for a designer. The system is highly mixed and turbulent when $\alpha = 0$, which leads to a high removal efficiency, find equation (7). (7) $\eta = \phi(\phi + 1)$. (8) $\eta = \phi \ln$ a vortex basin, the two extremes are represented by equations 7 and 8. When analysing main clarifiers and taking discrete particle settling into account, equation 8 is occasionally utilised.

Tabel 2: Reference of η values for calculation

Values of η
to obtain

2 This suggests that the primary mechanism for removal in a vortex grit basin is gravity sedimentation. The ratio Q/d^2 would be analogous to the surface overflow rate (SOR), which would serve as a crucial design parameter.

2.5 Based on their consistent Froude number, early workers seem to have embraced this as the proper value.Applied to free-surface flows, the primary effect is Froude number, which also guides similarity rules for such flows (White 1994)The Froude number can be defined as the ratio of gravity to inertia, to put it simply. It has been used since the beginning to analyse vortex grit basins (Pretorius 2012).

3 Unit volume and HRT would be equivalent to d^3 . HRT is mentioned in certain removal efficiency theories for sedimentation basins, notably in relation to rectangular basins. This would be an unexpected outcome.

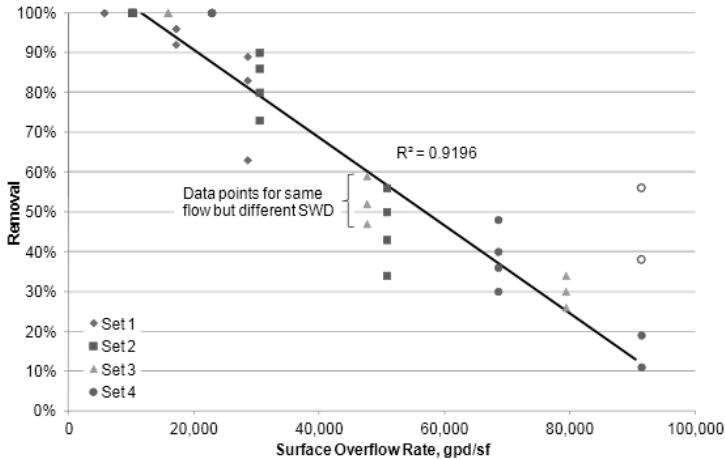


Figure 7 : Removal with respect to SOR (Sullivan et al. 1974)

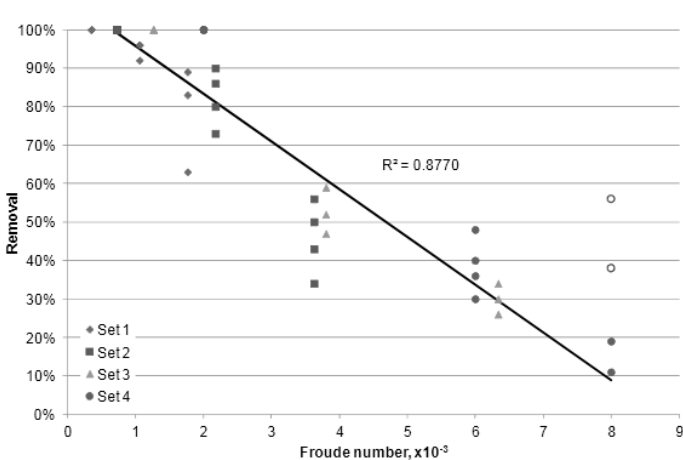


Figure 8 : Removal with respect to Fr (Sullivan et al. 1974)

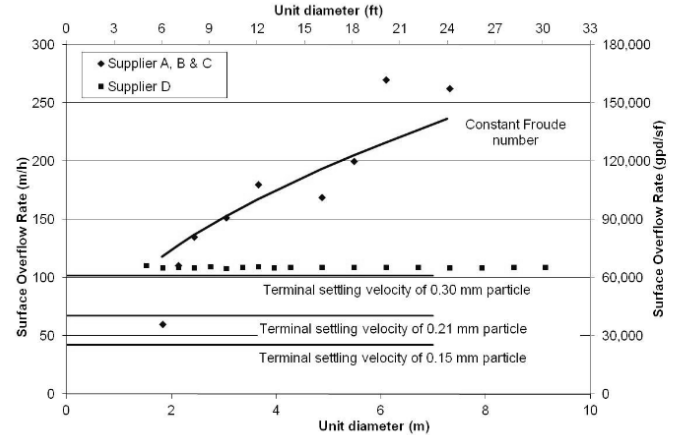


Figure 9: Particle Terminal Settling Velocities and Vortex Basin Design Capabilities

The relationship between removals and surface overflow rate (SOR) is clearly demonstrated in (Figure 9), exhibiting a strong correlation. Variations in removal at each tested flow (SOR) are observed due to differing liquid depths, suggesting that deeper liquid levels may enhance removal within the tested range. The ratio of SWD to liquid depth ranged from 0.3 to 0.9, whilst in current full-scale designs, this ratio typically falls between 0.4 and 0.65.

Given the fixed diameter of the test unit, it was not possible to determine whether liquid depth or hydraulic retention time (HRT) had a greater influence on removal, as these parameters were directly proportional in this setup. All experiments were conducted in a unit with a sloped floor, precluding verification of any benefits from the cone volume.The unexpectedly high removals at elevated SOR, represented by two hollow data points on the right, indicate that the unit may have achieved increased removal through centrifugal separation. However, this phenomenon is unlikely to occur in full-scale vortex units due to low inlet

velocity (and head loss), larger unit radius, and the use of peripheral outlets.

The researchers propose that the optimal design could have an α value as high as 1.5. Utilising this figure 9, the safety factor for 90% removal is 1.7, whilst for 95% it is 2.1. This indicates that a more precise estimation of α is crucial for practical designs. Two approaches could achieve this: firstly, developing an analytical solution that expresses α in terms of variables known to the designer; secondly, determining α experimentally and applying it to similar designs.

The researchers seem to suggest that α would remain constant if the ratio between surface overflow rate and inlet velocity is maintained. Current vortex grit basin designs typically aim to keep inlet velocity constant, based on the estimated velocity needed to suspend coarse grit. Consequently, sizing vortex grit basins to maintain a constant surface overflow rate would preserve a constant α value, ensuring consistent performance across differently sized basins with constant inlet velocity and SOR.

The researchers' modelling and experimental testing revealed that tank diameter (surface overflow rate), inlet diameter (inlet velocity), and outlet geometry were the three most influential parameters on performance. The impact of the latter two will be discussed subsequently.

A study conducted by (Chien et al. 2010) examined a vortex grit basin's failure to meet specified removal efficiency. Their CFD model indicated that halving the flow through the basin from 70 to 35 mgd, thus reducing surface overflow rate by 50%, would decrease the maximum velocity from 6 to 2 ft/s. Furthermore, the model predicted an increase in coarse grit (diameter = 649 μm ; SG = 1.4) removal from 89% to 100%, and medium grit (diameter = 254 μm ; SG = 1.4) removal from 7% to 72%. However, within two years of commissioning the vortex grit basins, the plant had to halt downstream aeration basins to remove grit for the first time ever. Grit was also discovered in secondary clarifiers and chlorine contact basins.

The (figure 9) also shows the typical settling velocities for different particle sizes that are utilised in grit removal standards. It should be noted that the terminal settling velocities were calculated with the optimistic assumption of an SG of 2.65, as supported by the data in Table 1.

The figure indicates that, under the most advantageous set of assumptions, i.e., SG = 2.65 and that equation 8 is correct, even the most conservative of the suppliers would only achieve partial removal of the three particle sizes indicated. According to the statistics, most suppliers of big vortex grit basins are overly enthusiastic about their capability.

Refuse components undergo a number of intricate chemical and biological reactions when they are deposited in a

landfill, along with solid wastes (Dev 2007). (Rettenberg 2006) observed that this landfill undergoes four phases of decomposition: (1) the initiation of an aerobic phase, (2) the formation of an anaerobic acidic phase, (3) the establishment of a methanogenic phase, and (4) the emergence of a stable methanogenic phase. According to subsequent reports, an aerobic phase or additional humic decomposition phase was suggested (Serdarevic 2018). When garbage or solid waste is thoroughly digested, the diffusion rate of oxygen to a landfill may exceed the rate of microbial oxygen depletion a research from (Ireland 2000). This could result in the anaerobic ecosystem of the landfill transitioning to an anaerobic environment over time. Given that refuse and solid waste are repeatedly buried in landfills in multiple layers, lifts, and cells over an extended period of time, it is not uncommon for various parts or layers of the landfill to undergo varying degrees of decomposition (Ahn et al. 2002). The features of landfill leachate mostly depend on the decomposition condition of trash and the related components. Consequently, the properties and composition of leachate might vary significantly across different landfills (Ahn et al. 2002).

3.4. Landfill stabilization

For more than 50 years, landfills have been seen all across the world. This prolonged duration culminated in the conclusion that it encompasses intricate sequences of biological and chemical reactions triggered by the entombment of solid waste in a landfill.

As the municipal and industrial solid waste broke down, the investigation found leachate and gases emitted from various materials and compositions (Alfaia et al. 2017). The initial four phases are the aerobic phase, the anaerobic acid phase, the beginning methanogenic phase, and the stable methanogenic phase. Breakdown

Because there is no air in the atmosphere during these phases, waste cells start to change into aerobic cells (Kjeldsen et al. 2002). This mostly pertains to the fact that numerous meticulously monitored landfills are aged between 30 and 50 years, indicating that they remain in the stable ethanogenic phase. Freshly buried waste undergoes an initial aerobic stage during which the oxygen in the empty spaces is quickly absorbed, producing CO₂ and perhaps raising the temperature (Çeçen & Aktaş 2004). The phase for a specific dump lasts only a few days, as oxygen is not replenished after the waste is covered.

Additionally, during this aerobic period, the wastes are typically not at field capacity. The majority of leachate generated during this phase results from the discharge of liquid moisture during both compaction and precipitation of the buried solid waste refuse (Çeçen & Aktaş 2004).

Waste rejects turn anaerobic to support fermentation reactions after the oxygen sources are depleted on schedule (Abdel-Shafy & Mansour 2018). The total dry weight of municipal solid wastes (MSW) is approximately 45 to 60% due to the presence of hemicelluloses and cellulose

materials ((Ed.). 2012). They constitute the primary biodegradable component of solid waste . Biodegradation of solid waste in landfills under anaerobic circumstances produces methane and carbon dioxide. The biodegradation of cellulose and hemicellulose is accomplished by three principal groups of bacteria: (1) hydrolytic and fermentative bacteria, which hydrolyse polymers and ferment the resultant monosaccharides into carboxylic acids and alcohols; (2) acetogenic bacteria, which convert these acids and alcohols into hydrogen, acetate, and carbon dioxide (CO₂); and (3) methanogens, which transform the end products of acetogenic reactions into methane and carbon dioxide. Under narrow pH values and in the vicinity of neutral, these conversion and biodegradation processes operate effectively.

The pH value decreases during the second phase of anaerobic biodegradation due to the accumulation of carboxylic acids and the dominance of fermentative, hydrolytic, and acetogenic bacteria (Abdel-Shafy 1996a). This stage allows for the measurement of the leachate's greatest COD and BOD concentrations. In such an acid phase, the BOD:COD ratio was reported to exceed 0.4. The acidic pH of the fermented system renders the acid phase leachate chemically aggressive, capable of dissolving various substances (Abdel-Shafy 1996a). The initial methanogenic phase occurs when detectable quantities of methane are produced . The initiation of the methanogenic phase is primarily defined by the refuse's initial pH level being adequately neutral for the duration of methanogenic bacterial proliferation. These bacteria transform the accumulated acids into both methane and carbon dioxide; thus, the production rate of methane gas will be enhanced by these methanogenic bacteria [30]. In the interim, the decomposition of cellulose and hemicellulose will commence. As a result, the concentrations of BOD and COD diminish while the pH level rises due to the consumption of acids. Consequently, the ratios of BOD to COD will diminish when the carboxylic acids are utilised (Abdel-Shafy & Mansour 2018; Abdel-Shafy & Kamel 2016). The rate of methane synthesis will peak during a stable methanogenic phase and then fall as the pool of soluble substrate, specifically carboxylic acids, diminishes (Abdel-Shafy & Mansour 2014). The pace at which hemicellulose and cellulose hydrolyse during this phase determines the rate at which methane CH₄ is produced. A regular steady state concentration of a few mg.L⁻¹ is maintained as the pH continues to rise. The leachate contains a certain amount of COD and is primarily composed of resistant substances like fulvic and humic acids. Due to the quick consumption of generated carboxylic acids, the BOD to COD ratio will typically gradually drop below 0.1 during this phase (Abdel-Shafy 1996b; Abdel-Shafy & Mansour 2014).

4. Flow Splitter

Wastewater is sent to various treatment stages using flow splitters, which maximises treatment efficiency and ensures proper loading on each unit. The research topic will investigate the hydrodynamic performance of hydraulic structures, focusing on flow-induced vibrations, sediment transport, and water level dynamics. (Steven J. Wright, Daniel B. Schlapfer, Razik Al-Saigh 1988) study will analyse the negligible impact of vortex shedding on vane structures, with vibration frequencies exceeding 65 Hz compared to a maximum shedding frequency of 0.15 Hz. Additionally, it will assess sedimentation patterns in splitter chambers under varying flow regimes and grit characteristics, identifying the impact of increased plant flows on sediment deposition. The study will also examine water surface level variations across overflow weirs, correlating them with flow distribution patterns to optimise sensor placement and minimise deposition zones.

4.1. Numerical review of split chamber

Design practice example (Caltrans Division of Design 2020)

This design example outlines the installation of an Upstream Flow Splitter (UFS) upstream of a Best Management Practice (BMP) Infiltration Basin at a site in Orange County. The basin's bottom elevation is set at 52.43 metres to maintain a 3.05-meter clearance above the seasonally high groundwater level of 49.38 metres. With an infiltration rate of 25.4 mm per hour for the sandy loam soil and a safety factor of 2, the water quality volume (WQV) depth in the basin is limited to 0.61 metres to achieve a 48-hour drawdown period. As a result, the water surface elevation for the WQV (WSWQ) is set at 53.04 metres.

The basin operates "offline" by using a UFS to regulate inflow, ensuring that the maximum 25-year water surface elevation in a surcharged condition (WSSUR) stays at least 0.21 metres below the roadway subgrade. The allowable WSSUR elevation is limited to 53.09 metres. The UFS control depends on the storage volume within the downstream BMP. Figures 10, 11, and 9 provide schematics and profiles of the storm drain, UFS, and BMP Infiltration Basin. The discharge from the BMP will flow into Orange County Creek, which runs adjacent to the project site.

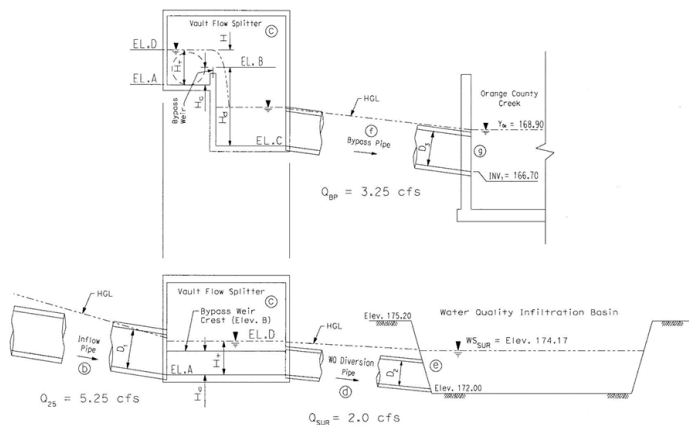


Figure 10: System Profile Showing Flow in the Surcharged Condition

This design example in figure 10 outlines the installation of an Upstream Flow Splitter (UFS) upstream of a BMP Infiltration Basin at a site in Orange County. The bottom of the basin is set at 52.43 metres to maintain a 3.05-metre clearance above the seasonally high groundwater level of 49.38 metres. The infiltration rate of the sandy loam soil is 25.4 mm per hour. With a safety factor of 2, the depth of the Water Quality Volume (WQV) in the basin is limited to 0.61 metres to achieve a 48-hour drawdown. As a result, the WQV water surface elevation (WSWQ) is set at 53.04 metres.

The basin is designed "offline," with a UFS regulating inflow to ensure the maximum 25-year water surface elevation under surcharged conditions (WSSUR) remains at least 0.21 metres below the roadway subgrade. The allowable WSSUR elevation is set to 53.09 metres. UFS control is based on the volume stored in the downstream BMP. Figures 8-3, 8-4, and 8-2 provide schematics and profiles of the storm drain, UFS, and BMP. The BMP will discharge into Orange County Creek, which flows adjacent to the site.

Given Data

The following hydrologic data and design parameters inform the UFS design:

- $Q_{25} = 0.149 \text{ m}^3/\text{s}$ (Peak design flow rate for the storm drain system)
- $TC = 15$ minutes
- $WQV = 481 \text{ m}^3$ (Water quality volume for treatment)
- $WSWQ = 53.04 \text{ m}$
- $WSSUR = 53.09 \text{ m}$ (Maximum water surface elevation under 25-year peak flow)
- $QSUR = 0.057 \text{ m}^3/\text{s}$ (Maximum discharge from the overflow structure)
- $VX = 0.054 \text{ acre-ft} = 66 \text{ m}^3$ (Flow volume at 50% of Q_{25} over 15 minutes)

- $WSX = 52.58 \text{ m}$ (Pool elevation for VX)

Additional Design Parameters

- BP Pipe Invert Elevation at Orange Creek (INV1) = 50.82 m
- Bottom Elevation of BMP Infiltration Basin (INV2) = 52.43 m
- TOG Elevation on Drainage Inlet Upstream of UFS = 54.48 m
- IN Pipe Diameter (D1) = 0.61 m

Step 1: Select the Diameter of the WQ Diversion Pipe and Set the Bypass Control Elevation (BCE)

The proposed UFS location is selected with a WQ diversion pipe length of approximately 9.14 metres. The BCE is set at 53.04 metres, matching the WSWQ.

- $BCE = WSWQ = 53.04 \text{ m}$
- Initial Invert Elevation of WQ Diversion Pipe (Elevation A) = 52.58 m

To minimise headwater depth on the WQ diversion pipe at Q_{25} , the largest standard pipe diameter of 0.46 metres is chosen. Hydraulic analysis using HY-8 (FHWA, 2016) confirms that the selected 0.46-meter pipe will maintain the headwater depth below the BCE at Q_{25} .

Hydraulic Analysis Parameters

- $Q_{25} = 0.149 \text{ m}^3/\text{s}$
- Pipe Diameter = 0.46 m
- Pipe Length = 9.14 m
- Upstream Invert Elevation = 52.58 m
- Downstream Invert Elevation = 52.43 m
- Slope = 0.0167 m/m
- Manning's "n" = 0.012 (Precast RCP)

The hydraulic analysis, summarised in Table , indicates that the peak flow rate of $0.149 \text{ m}^3/\text{s}$ is low relative to the UFC's cross-sectional area, resulting in a flow velocity below 0.3 m/s. As the velocity head is negligible, it is not included in the calculation of headwater depth.

Table 3: Rating table for WQ diversion pipe

| Flow rate m^3/s | WQ Diversion Pipe Headwater depth (m) |
|---------------------------------|---------------------------------------|
| 0 | 0 |
| 0.0014 | 0.0182 |
| 0.0297 | 0.1584 |
| 0.0447 | 0.1851 |
| 0.0594 | 0.2286 |
| 0.0891 | 0.2956 |

| | |
|--------|--------|
| 0.1042 | 0.3231 |
| 0.1189 | 0.3535 |
| 0.1339 | 0.3779 |
| 0.1486 | 0.4053 |

The velocity for half-full flow and an empty basin is checked to ensure a minimum self-cleansing velocity of 0.91 m/s during smaller storm events. The minimum pipe slopes required to achieve this velocity are provided in Table 8-2 and must meet the standards outlined in HDM Topic 838.4(3).

- Manning's "n" = 0.012 (for precast RCP, per HDM Table 851.2)
- Minimum Velocity (v) for half-full flow = 0.91 m/s

Table 4: Half Full Flow Hydraulic Analysis of WQ Diversion Pipe

| Pipe Diameter (D)(m) | Pipe Area $A = \frac{\pi D^2}{4}$ | Half Full Area $a=A/2$ | Half Full discharge $Q=Va=3a$ | Wetted Perimeter $r, p = \frac{\pi D}{2}$ | Hydraulic Radius $R = a/p$ | Minimum Allowable slope S |
|----------------------|-----------------------------------|------------------------|-------------------------------|---|----------------------------|---------------------------|
| 0.457 | 0.538 | 0.269 | 0.075 | 0.718 | 0.114 | 0.0021 |

The velocity for half-full flow and an empty basin is checked to ensure a minimum self-cleansing velocity of 0.91 m/s during smaller storm events. The minimum pipe slopes required to maintain this velocity are provided in Table 3. These slopes must meet the requirements of the HDM

- Manning's "n" = 0.012 (for precast RCP, per HDM Table 851.2)
- Minimum Velocity (v) for half-full flow = 0.91 m/s

The slope of the WQ diversion pipe (0.0167 m/m) exceeds the minimum required for half-full flow, ensuring the pipe is self-cleansing, and the proposed pipe profile is acceptable.

Step 2: Calculate Dimension "Hc"

Dimension Hc is calculated as:

$$(9) H_c = \text{Elevation B} - \text{Elevation A} = 53.04 - 52.58 = 0.46 \text{ m}$$

The hydraulic profile for the QHDM and a tailwater at WSX is shown in Figure 11.

Step 3: Calculate the Maximum Allowable Water Surface in the UFS During Bypass and the Corresponding Hydraulic Head "H"

The maximum allowable water surface elevation in the Infiltration Basin during surcharge (WSSUR) is 53.09 m, with a surcharge flow rate (QSUR) of 0.057 m³/s. The head loss through the WQ diversion pipe for QSUR is added to WSSUR to determine the maximum allowable water surface elevation in the UFS and the corresponding hydraulic head H for the bypass component (see Table 8-3

Table 5: Full Flow Hydraulic Analysis of WQ Diversion Pipe (QSUR)

| Pipe Diameter (D)(m) | Pipe Area $A = \frac{\pi D^2}{4}$ | Velocity $V=Q/A$ (m/s) | Velocity Head $V_h = \frac{V^2}{2g}$ | Entrance Loss $0.2V_h$ | Exit Loss V_h | Friction Loss SF (m/m) | Friction Head Loss HT = SFL | Total Head Loss HT (m) |
|----------------------|-----------------------------------|------------------------|--------------------------------------|------------------------|-----------------|------------------------|-----------------------------|------------------------|
| 0.457 | 0.538 | 0.344 | 0.269 | 0.075 | 0.718 | 0.114 | 0.0021 | 0.0101 |

Entrance loss coefficient (Ke) = 0.2, for rounded headwall entrance (FHWA 2009). Exit loss coefficient (Ke) = 1.0, for exit with D2/D1 > 10 and V < 2.0 fps (FHWA 2009). Sf = [Qn/(KQD2.67)]², KQ = 0.46 in English units (FHWA 2009).

Step 4: Calculate Elevation D and the Hydraulic Head H

$$(10) \text{Elevation D} = WS_{sur} + H_T = 53.09 + 0.0101 = 53.1 \text{ m}$$

$$(11) H = \text{Elevation D} - \text{Elevation B} = 53.1 - 53.04 = 0.06$$

The hydraulic profile for the bypass/surcharge condition is illustrated in Figure 9.

Step 5: Determine the Size of the Bypass (Weir Length or Pipe Diameter) and Select the Type of UFS

The bypass flow rate for the UFS design is calculated by subtracting the surcharge flow through the Infiltration Basin (which passes over the basin's overflow structure) from the peak flow rate:

$$(12) Q_{BP} = Q_{HDM} - Q_{SUR} = 0.149 - 0.057 = 0.092 \text{ m}^3/\text{s}$$

If a **Type 4 UFS** is used, the BP pipe must convey 0.092 m³/s with only 0.06 m of headwater. Rating curves for various BP pipe diameters are shown in Figure 8-1.

It is observed that a headwater depth of **0.06 m** on a Type 4 BP pipe, with diameters of **600 mm** or **900 mm**, results in a bypass capacity of less than **0.0085 m³/s**, which is far below the required **0.092 m³/s**. Achieving the necessary bypass flow would require raising the headwater elevation in the UFS to more than **53.29 m**, exceeding both the QSUR and WSSUR limits in the Infiltration Basin. As this conflicts with the BMP design criteria for the site, the Type 4 UFS is unsuitable, and a weir-type UFS is proposed.

Weir Length Calculation:

The weir equation is used to determine the required weir length. The coefficient for an submerged sharp-crested weir (CSCW) is calculated as:

$$(13) C_{scw} = 3.27 + 0.4(H/H_c) = 3.27 + 0.4(0.06/0.46) \\ = 3.32$$

The required weir length (L_w) for a sharp-crested weir with end contractions is calculated for Q_w = QBP:

$$(14) L_w = \left[\frac{Q_w}{C_{scw} \times H^{1.5}} \right] + 0.2H = \left[\frac{0.092}{3.32 \times 0.06^{1.5}} \right] + 0.02(0.06) \\ = 3.35 \text{ m}$$

To accommodate this weir length, a **Type 1 UFS** is selected for the site.

Step 6: Set the Invert Elevation for the BP Pipe and Calculate Hydraulic Head (Hd)

Since the flow rate over the weir is **0.092 m³/s**, a hydraulic analysis of the BP pipe is performed using the given tailwater conditions to finalise the invert elevation and hydraulic head.

The elevation at Orange County Creek is 51.48 m, and the invert of the BP pipe where it connects to the creek is set at 50.81 m. For the initial analysis, the invert elevation of the BP pipe (Elevation C in Figure 8-6) is set at 52.27 m. The hydraulic analysis must confirm that the pipe profile provides a minimum 250 mm of freeboard to the weir crest (BCE).

The analysis is performed using HY-8 and is based on the following parameters:

- Flow rate (QBP) = 0.092 m³/s
- Diameter = 600 mm (Same as IN pipe diameter)
- Pipe length = 25.91 m
- Upstream invert elevation (Assumed) = 52.27 m
- Downstream invert elevation = 50.81 m
- Slope = 0.0565 m/m
- Manning's n = 0.012 (for precast RCP)

The results of the hydraulic analysis are summarised in Table 8-4. This analysis ensures the proposed pipe profile can convey the required flow while maintaining adequate freeboard and preventing overflow.

Table 6: Rating Table for BP Pipe

| Flow rate m ³ /s | WQ Diversion Pipe Headwater depth (m) |
|-----------------------------|---------------------------------------|
| 0 | 0 |
| 0.0014 | 0.015 |
| 0.0184 | 0.106 |
| 0.0277 | 0.134 |
| 0.0036 | 0.155 |
| 0.046 | 0.176 |
| 0.055 | 0.192 |
| 0.064 | 0.207 |
| 0.073 | 0.225 |
| 0.082 | 0.24 |
| 0.0911 | 0.252 |
| | EL. 52.52 |

Elevation C = 52.27 m

$$(15) H_d = \text{Elevation B} - \text{Elevation C} = 53.04 - 52.27 \\ = 0.77$$

The hydraulic profile for this analysis is illustrated in Figure 11.

Step 7: Confirm UFS Fits within Vertical Drop Available

The system profiles for the WQ flow path and bypass flow path are illustrated in Figures 12 and 11. The designer confirmed that there is adequate vertical drop to accommodate these profiles and that the UFS fits beneath the proposed finished grade at the site.

Step 8: Complete Hydraulic Analysis for Upstream Drainage System

A hydraulic analysis for the peak design flow rate (Q₅₂₅) was conducted for the upstream drainage system using a downstream water surface control equal to the maximum water surface elevation in the UFS, calculated in Step 3. This surface was set at 53.08 m (174.20 ft). The resulting hydraulic grade line (HGL) provided sufficient freeboard at all drainage inlets, confirming that the UFS design is acceptable.

Table 7: Design summary of the flow splitter.

| Design Summary | Metres |
|----------------------------|--------|
| WQ Diversion pipe diameter | 0.457 |
| UFS Type | Type 1 |
| Length Bypass Weri (Lw) | 3.34 |
| HcType1 | 0.45 |
| 3.34Hd | 0.762 |
| Elevation A | 52.57 |
| Elevation B (BCE) | 53.03 |
| Elevation C | 52.27 |

The UFS in this example effectively directs the required water quality volume (WQV) to the proposed BMP Infiltration Basin. The WQ diversion pipe meets the conveyance requirements for the peak design storm (Q_{25}). During bypass conditions, the maximum water surface elevation and flow rate in the BMP will not be exceeded. Additionally, the hydraulic evaluation confirmed that the UFS does not impair the hydraulic capacity of the upstream drainage system during the 25-year design storm.

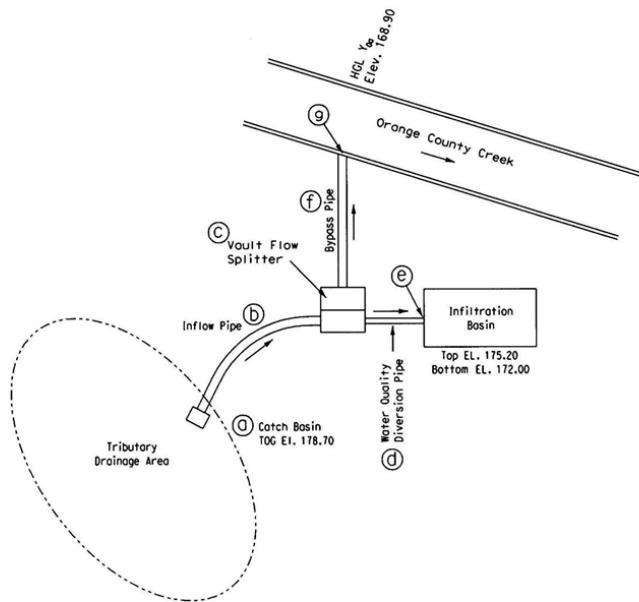


Figure 11: Plan of Project Site the water won't flow to country creek but it will flow to a different stage of treatment plant.

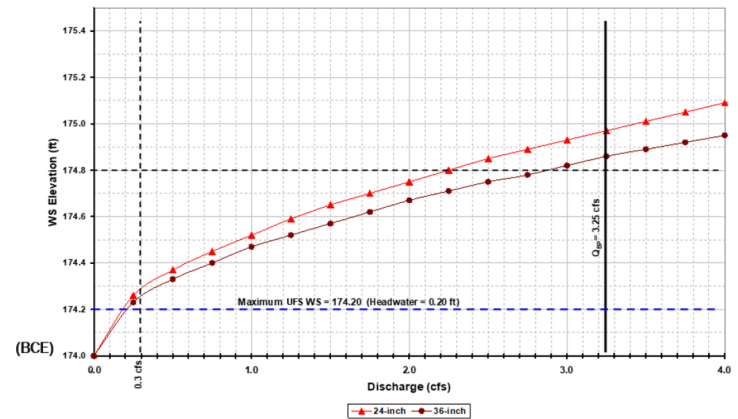


Figure 13: UFC Type 4- rating Curves

5. Sedimentation Tank

A research conducted by (Droste & Gehr 2018) to study the gravity-induced settling of particles with a density greater than the surrounding liquid is known as sedimentation. Basic settling tanks are commonly utilised for initial processing of water with high suspended solid content and for holding wastewater before treatment. These applications include:

- extracting grit from household sewage (often referred to as grit channels);
- holding raw sewage during stormy weather (typically called storm tanks, Figure 3.1) and
- equalising intermediate wastewater streams.

For lower concentrations of less dense solids, vertical and radial flow designs are employed. These are used for:

- eliminating large solids in primary wastewater treatment;
- removing flocs during water clarification;
- extracting lime softening precipitates;
- removing metal precipitates from industrial effluents and
- separating biomass in biological wastewater clarification.

Rectangular horizontal settling tanks are straightforward structures, typically measuring about 2m in depth with a length-to-width ratio ranging from 2 to 5 (Figure 14). These tanks feature an inlet at one end and an overflow weir at the opposite end for water discharge. As the water flows through, solid particles descend to the tank bottom. A mechanical system, usually comprising a chain and flight scraper, is employed to gather the accumulated sludge, moving it towards one end for extraction (Figure 15).

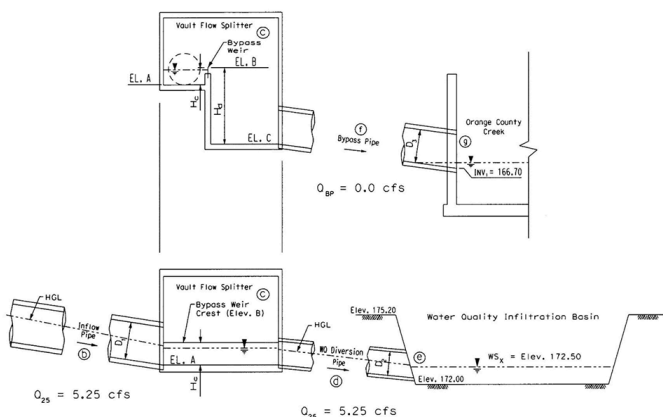


Figure 12: System Profile Showing Flow at Q_{25} Before Bypass

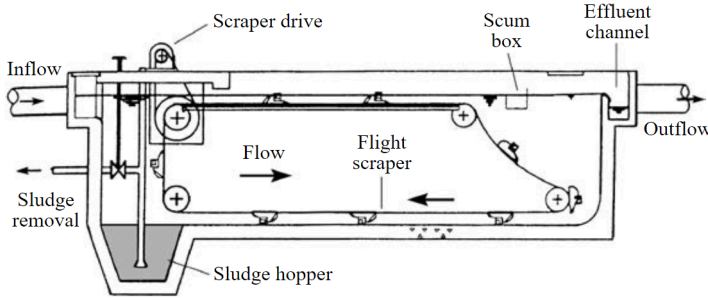


Figure 14: sedimentation tank (Droste & Gehr 2018)

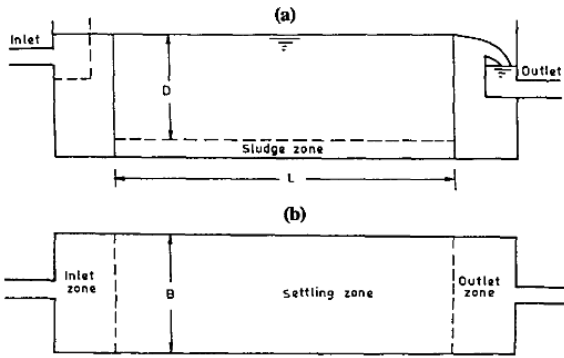


Figure 15: sedimentation tank model dimensions
(a) Longitudinal Section (b) Plan (Swamee & Tyagi 1996)

The design of primary settling tanks based on overflow rate and isoremoval plots, while widely used, has limitations in accurately predicting particle removal efficiencies across various operating conditions. A more rational approach would involve considering the specific removal efficiencies of different particle sizes and the scour criterion of deposited particles. This method would take into account the settling velocities of various particle fractions and the critical shear stress required to resuspend settled particles, providing a more comprehensive understanding of the tank's performance.

By incorporating the removal efficiencies of individual particle size ranges and the scour criterion, engineers like (Swamee & Tyagi 1996) can optimise the design of primary settling tanks for specific influent characteristics and treatment goals. This approach would allow for more precise control over the removal of suspended solids and associated contaminants, potentially improving the overall efficiency of the wastewater treatment process. Additionally, considering the scour criterion would help prevent the resuspension of settled particles during high-flow events, ensuring more consistent performance and reducing the risk of carry-over to subsequent treatment stages. (16) $V_0 = \frac{Q}{BL}$ by (Camp 1946) when

V_0 = Overflow Rate

$B \& L$ = Width and length. and (Rouse 1937) found

$$(17) \eta = 1 - P_0 + \int_0^{P_0} \frac{W}{V_0} dP \quad \text{when}$$

w = Fall Velocity of the particle for the proportion finer than P

P_0 = the fraction of the particles $W \leq V_0$. Stokes equation:

$$(18) w = \frac{(s-1)gd^2}{18\nu} \quad \text{when } s = \text{specific gravity of the particles}$$

g = gravitational acceleration d = particle size. (Rouse 1937) (19) $u_* / w \leq k$ since (20) u_* = shear velocity

$0.5 \leq k \leq 0.8$. (Cho & Sansalone 2013) found

$$(21) u_* = (\tau_0 / \rho)^{0.5} \quad \text{as } \tau_0 = \text{average bed shear stress}$$

ρ = mass density of water. And (PK Swamee 1991) found

$$(22a) P = \left[\left(\frac{d_*}{d} \right)^{m/m} + 1 \right]^{-n} \quad \text{for}$$

m = slope of the size distribution curve for smaller diameters (Figure 14), (22b) $n = -1.4427 \ln P_*$ for $P = P_*$.

$$(22c) P = (d/d_*)^m, \quad d = d_*. \quad (23) dP = m d_*^{-m} d^{m-1} dd,$$

$1 \leq m \leq \infty$ and $0 \leq n \leq 1$ with the increase in m and decrease in n the size of distribution tends to become more

uniform. $d \ll d_*$. And (24) $\eta = 1 - P_0 + \frac{m}{m+2} \frac{(s-1)gd_0^{m+2}}{18\nu V_0 d_*^m}$ From

(17) (18), and (22) the removal efficiency was found

$$(24a) d_0 = \left[\frac{18\nu Q}{(s-1)gBL} \right]^{0.5} \quad d_0 = \text{particle size corresponding to the overflow rate } w = V_0 \text{ and combining (16) and (18) find (22a).}$$

$$(24b) P_0 = \left[\frac{18\nu Q}{(s-1)gBLd_*^m} \right]^{0.5m} \quad \text{by combining (21c) and (23a).}$$

$$(25) \eta = 1 - \frac{2}{m+2} \left[\frac{18\nu Q}{(s-1)gBLd_*^m} \right]^{0.5m} \quad \text{when combining (21c) and (23A)}$$

$$(26) d_0 = d_* \left[(0.5m + 1)(1 - \eta_D) \right]^{1/m}$$

η_D = design efficiency when combining (9a) and (10).

(26) Vq/BDD = depth of the tank $V > V_{sc}$ sludge velocity it will scour the particles d_{sc} = particle size. (27) $\tau_0 = \rho g D S_f$

average tractive shear stress. (28) $S_f = \frac{fV^2}{8gD}$

f = Darcy friction factor Combining (20), (29), (30), and

settling $V = V_{sc}$ and $d = d_{sc}$ get : (31) $V_{sc} = \left(\frac{8}{f} \right)^{0.5} \frac{k(s-1)gd_{sc}^2}{18\nu}$

V_{sc} = sludge Viscosity. (32) $f = \frac{0.316}{R^{0.25}}$ friction for finished cement.

(33) $R = 4Q/\eta\nu BR$ = reynolds number,

(34) $f = 0.223(\nu B/Q)^{0.25}$ by eliminating R between (32) and

(35). (52) $V_{sc} = \frac{k(s-1)gd_{sc}^2}{3\nu} \left(\frac{Q}{\nu B} \right)^{0.125}$ after combining (31) and

(34). (36) $V_{sc} = \frac{6kQ}{BL} \left(\frac{Q}{\nu B} \right)^{0.125}$ If $V = V_{sc}$ and using (28) and (36)

one gets. (37) $L = 6kD(Q/\nu B)^{0.125}$ L = Length of settling tank

$$K = 0.8. \quad (38) B = \frac{3.51Q}{d_0 \sqrt{k(s-1)gD} \left[\frac{\nu}{d_0 \sqrt{k(s-1)gD}} \right]^{9/7}}$$

$$(39) L = 5.13Kd \left[\frac{d_0 \sqrt{k(s-1)gD}}{\nu} \right]^{2/7} \quad d_0 = \text{particle size}$$

- Minimum depth lying between 1.5m and 2.5m for mechanically cleaned settling tanks. (Swamee & Tyagi 1996)

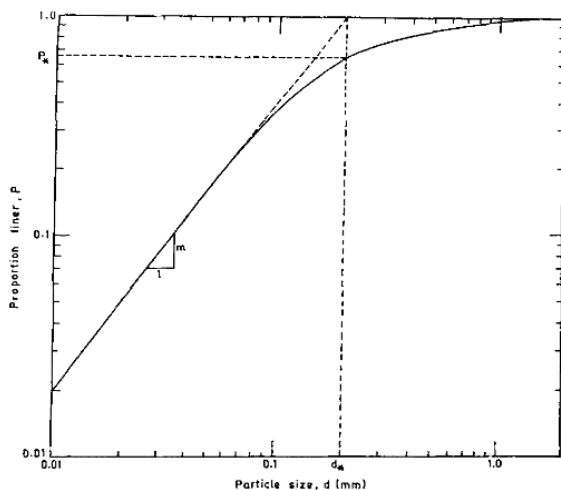


Figure 15: Particle size distribution curve (Swamee & Tyagi 1996)

The tank dimensions depend on the chosen dsc value. A small dsc leads to larger tank dimensions with only marginal increases in sediment yield. Conversely, a large dsc washes away most settled particles. As particles smaller than d_0 progressively decrease due to removal in the influent, dsc can be assumed equal to d_0 . Using equations (25a) and (35) with this assumption, we can determine the tank dimensions.

5.1 Chemical precipitation

This treatment involves the use of chemical coagulants and coagulant aids to remove and/or co-precipitate pollutants from landfill leachate and/or effluent. To get heavy metals like Cu, Ni, and Mn out of wastewater (Abdel-Shafy 2015), this study was done at different pH levels. The chemical and physical properties, as well as the concentrations of Ni, Cu, and Mn, were examined. The removal rate for Ni, Cu, and Mn was 96.0, 97.5, and 90.0%, respectively, when sodium hydroxide (NaOH) was used at a pH of 9.5. Additional research was conducted by combining sodium hydroxide with ferric chloride at a concentration of 50 mg/l and varying pH levels.

This investigation demonstrated that all metals in question were eliminated at a rate exceeding 98% at a pH of 12.0. The results obtained from the use of 70 mg/l of alum and NaOH at varying pH levels indicated that the optimal pH was 12.0, in which the removal rate for Ni and Cu was 100% and for Mn was 84%. The usage of lime (CaO) at different doses also demonstrated that the best pH level was 11.0, which removed more than 100% of Ni and Cu and 93% of Mn. The study also examined the use of limestone (CaCO_3) at varying concentrations. According to this study, adding 1.0

g/l limestone raised the pH from 2.0 to 5.85, at which point Ni, Cu, and Mn removals were 90.2, 100, and 75.1%, respectively. After increasing the CaCO_3 dose to 3.0 g/l, Ni and Cu were removed at a rate exceeding 100%, while Mn was removed at a rate of only 90.6% (Ghafari et al. 2009). It has been verified that the solubility product (SP) significantly influences the precipitation of contaminants, particularly metals. Given that the pH range for effluent guidelines should be between 6 and 9, the carbonate treatment is advised because of its pH 7 buffering capacity value. However, lime has a good commercial value and produces effective precipitation. Nevertheless, the disadvantage of utilising lime is the challenge of regulating the ultimate pH of the treated effluent. However, by applying acid as necessary to adjust the final pH, this issue can be resolved. In an effort to eradicate pollutants of considerable potency, the procedure is frequently implemented during the pretreatment phase (Abdel-Shafy 2015).

6. IDALs: Intermittently decanted aerated lagoons wastewater

An alternate secondary process is IDALs. The IDAL anaerobic zone receives settled wastewater via pumping from the major distribution structure. To aid in the removal of phosphorus, discarded pickle liquor that is high in iron is added. Wastewater in IDALs passes through three steps in a single tank: settling, decanting, and aeration.

- Aeration: Through diffusers, air is injected into the IDAL. It reduces biological oxygen demand (BOD) by breaking down organic matter and water (nitrification) with the help of microorganisms in the tank.
- Getting settled: The water is motionless and there is no longer any air pumped into the tank. In the absence of oxygen, bacteria turn nitrates into nitrogen gas by using the carbon in organic matter as food. The atmosphere is exposed to the gas. Particles that are solid sink to the bottom. Before being treated for the manufacture of biosolids, some pass through a thickening tank. In order to supply microorganisms for entering wastewater, the remaining solids are sent back to the IDAL.
- Decanting: The clear wastewater settles and then runs into an equalisation basin from the top of the lagoon over weirs. The flow to the tertiary treatment process is managed by this basin.

Using traditional systems like aerated stabilisation ponds, aerated and non-aerated lagoons, and manmade and natural wetland systems is the most straightforward method for the anaerobic-aerobic treatment. These systems undergo anaerobic therapy at the bottom end and aerobic treatment at the top. The retention period ranges

from a few days to 100 days, with an average organic loading of 0.01 kg BOD/m³ 3 days. (Wang et al. 2005)

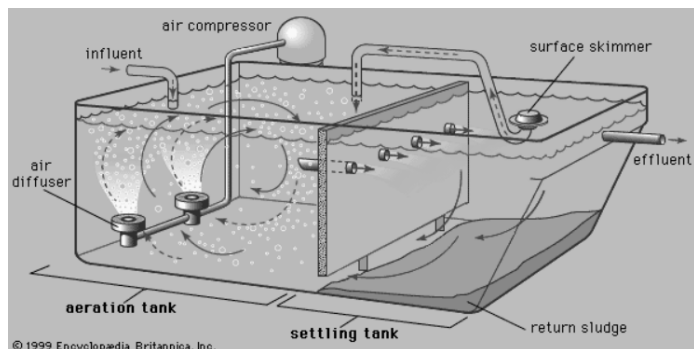


Figure 16: Diagram of IDAL's operation

For green olive debittering wastewater with COD ranging from 25,000 to 100,000 mg/L, (Aggelis et al. 2001) determined that neither anaerobic nor aerobic processes alone could effectively treat the waste. When dealing with such high-strength industrial wastewaters, singular anaerobic or aerobic treatment fails to produce effluents compliant with discharge limits. Utilising anaerobic-aerobic processes can reduce operating costs by a factor of eight compared to aerobic treatment alone, whilst achieving high organic matter removal efficiency, reduced aerobic sludge production, and eliminating the need for pH adjustment.

A study conducted by (Cakir & Stenstrom 2005) Cross-over points, which range from 300 to 700 mg/L influent wastewater ultimate BOD (BOD_u), are essential for aerobic treatment systems to operate efficiently. When treating influents at higher concentrations than the cross-over values, the benefits of anaerobic treatment exceed those of aerobic treatment, and anaerobic treatment often uses less energy with possible recovery of nutrients and bioenergy. However, aerobic systems remove more soluble biodegradable organic matter material than anaerobic systems do, and the biomass they produce is typically well flocculated, which lowers the concentration of suspended solids in the effluent. Consequently, an aerobic system's effluent quality is typically higher than an anaerobic system (Leslie Grady et al. 2009).

Anaerobic reactors are preferred for treating highly contaminated industrial wastewater due to their high COD levels, energy generation potential, and minimal excess sludge production.

However, practical applications face challenges such as slow microbial growth, poor settling rates, process instabilities, and the necessity for post-treatment of harmful anaerobic effluent containing NH₄⁺ and HS⁻ found by (Heijnen et al. 1991). Despite the high efficiency of anaerobic processes, complete organic matter stabilisation is often unattainable due to the wastewater's high organic content. The resulting anaerobic effluent contains solubilised organic matter

suitable for aerobic treatment, suggesting the viability of anaerobic-aerobic systems (Gray 2010) found the need for subsequent aerobic post-treatment to meet discharge standards.

(Vera et al. 1999) and (Cervantes et al. 2006) identified several benefits of the anaerobic-aerobic process:

- Significant resource recovery potential: Anaerobic pretreatment removes most organic pollutants, converting them into biogas, a useful fuel.
- High overall treatment efficiency: Aerobic post-treatment refines the anaerobic effluent, resulting in superior overall treatment efficiency and mitigating fluctuations in anaerobic effluent quality.
- Reduced sludge disposal: Digesting excess aerobic sludge in the anaerobic tank minimises total stabilised sludge production, lowering disposal costs and increasing gas yield.
- Low energy consumption: Anaerobic pretreatment acts as an influent equalisation tank, diminishing diurnal variations in oxygen demand and further reducing the required maximum aeration capacity.
- Effective volatile organic compound degradation: When present in the wastewater, volatile compounds are broken down during anaerobic treatment, preventing volatilisation in the aerobic stage.

(Leslie Grady et al. 2009) determined that the primary factor influencing microbial growth environments is the final recipient of electrons extracted during chemical oxidation for energy acquisition. These electron acceptors fall into three main categories: oxygen, inorganic substances, and organic compounds. An environment is deemed aerobic when dissolved oxygen is sufficiently available and not rate-limiting. This condition typically yields the most efficient growth and a high ratio of biomass production to waste decomposition. Technically, any non-aerobic environment is anaerobic. However, in wastewater treatment, the term 'anaerobic' typically refers to conditions where organic compounds, carbon dioxide, and sulphate serve as the principal terminal electron acceptors, resulting in a highly negative electrode potential and less efficient growth. When nitrate and/or nitrite act as the primary electron acceptors in oxygen's absence, the environment is termed anoxic. The presence of these compounds leads to a higher electrode potential and more efficient growth compared to anaerobic conditions, albeit not as high or efficient as in aerobic environments.

The biochemical environment significantly impacts the microbial community's ecology. Aerobic conditions generally support diverse food chains, ranging from bacteria to rotifers. Anoxic environments are more restricted, while anaerobic conditions are the most limited, predominantly supporting bacterial life. The biochemical environment also influences treatment outcomes due to the varying metabolic pathways of microorganisms in these

three environments. This distinction becomes crucial in industrial wastewater treatment, as certain transformations may occur aerobically but not anaerobically, and vice versa.

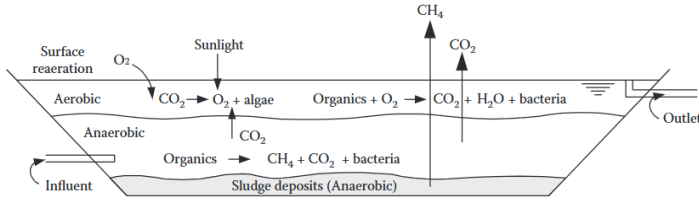


Figure 17: Chemical reaction of IDAL's(Leslie Grady et al. 2009)₂

In the past, lagoons have been constructed as expansive earthen reservoirs, reminiscent of typical "South Sea island lagoons" due to their size. Initially, these structures were unlined, but this approach proved problematic due to the risk of basin contents seeping into groundwater. As a result, current design standards mandate the use of an impermeable liner. The environmental conditions within lagoons can vary significantly, contingent upon the degree of mixing employed. Thoroughly mixed and aerated lagoons can maintain aerobic conditions throughout, whilst less mixing leads to solids settling, creating anoxic and anaerobic zones.

Completely mixed aerated lagoons (CMALs) are generally categorised as fully mixed reactors used primarily for soluble organic matter removal, although they can also facilitate the stabilisation of insoluble organic matter and nitrification. Facultative/aerated lagoons (F/ALs), as depicted in Figure 17, are mixed but not sufficiently to maintain all solids in suspension. Consequently, the upper regions tend to be aerobic, while the bottom contains anaerobic sediments. Anaerobic lagoons (ANLs) are not intentionally mixed; any mixing occurs solely due to gas evolution within them.

Lagoons represent one of the oldest biological wastewater treatment methods, with a history spanning over 3000 years(Rahman et al. 2019). They have been utilised as standalone treatment systems prior to surface water discharge, as well as for pretreatment and/or storage before conventional system or wetland treatment. A diverse range of industrial and municipal wastewaters has been processed using lagoon systems.

Aerobic biological processes that rely on the suspended growing biomass include aerated lagoons, sequencing batch reactors (SBR), and conventional activated sludge processes (Aziz, Aziz & Yusoff 2011; Hussein I. Abdel-Shafy et al. 2022). Attached-growth systems include the moving-bed biofilm reactor (MBBR) and bio-filters, which are seen in Figure bellow . Ammonium-nitrogen

nitrification and the reduction of biodegradable organic contaminants can both benefit from aerobic treatment. The membrane bioreactor, which combines aerobic bioreactors with the membrane separation process, garnered additional interest (Ahmed & Lan 2012).

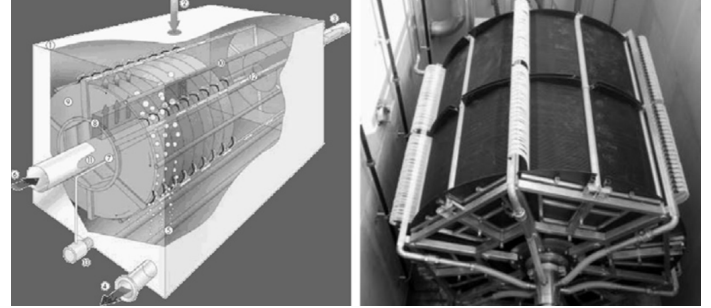


Figure 18: MBBR as a submerged, rotating, and aerated system (Abdel-Shafy & Abdel-Shafy 2017).

Theoretical consideration (Joe Middlebrooks et al. n.d.): Ammonia-N removal in facultative wastewater stabilisation lagoons can occur through the following three processes:

1. Gaseous ammonia stripping to the atmosphere
2. Ammonia assimilation in algal biomass, and
3. Biological nitrification

Nitrification often does not account for a major amount of ammonia-N removal, as seen by the low quantities of nitrates and nitrites in lagoon effluents. Temperature, organic load, detention period, and wastewater properties all have an impact on ammonia-N assimilation in algal biomass, which is contingent upon the biological activity in the system. Temperature, pH level, and lagoon mixing conditions are the primary determinants of the rate of gaseous ammonia losses to the atmosphere. The equilibrium equation $NH_3 + H_2O \leftrightarrow NH_4^+ + OH^-$ is shifted toward gaseous ammonia by alkaline pH, whereas the mass transfer coefficient's magnitude is influenced by the mixing conditions. Both the mass transfer coefficient and the equilibrium constant are impacted by temperature.

(40) $V \frac{dC}{dt} = Q(C_0 - C_e) - KA(NH_3)$ when $Q = \text{flow rate, } m^3/d$
 $C_0 = \text{influent concentration of } (NH_4^+ + NH_3), \text{ mg/L as N}$
 $C_e = \text{effluent concentration of } (NH_4^+ + NH_3), \text{ mg/L as N,}$
 $C = \text{average lagoon contents concentration of } (NH_4^+ + NH_3), \text{ mg/L as N,}$
 $V = \text{volume of the pond, } m^3,$
 $k = \text{mass transfer coefficient, } m/d,$
 $A = \text{surface area of the pond, } m^2 \text{ and } t = \text{time, days.}$

$$(41) K_b = \frac{[NH_4^+][OH^-]}{[NH_3]} \quad \text{when}$$

$$K_b = \text{ammonia dissociation constant.} \quad (42) [H^+] = \frac{K_w}{[OH^-]},$$

$$(43) C = NH_4^+ + NH_3, \quad (44) NH_3 = \frac{C}{1 + 10^{pK_w - pK_b - pH}}, \quad pK_w = -\log K_w,$$

$pK_b = -\log K_b$, Assuming steady state conditions where

$$C_e = C \quad \text{find equation} \quad (45) \quad \frac{C_e}{C_0} = \frac{1}{1 + \frac{A}{Q} k \left[\frac{1}{1 + 10^{\frac{K_N - pH}{10}}} \right]}$$

K = removal rate coefficient (1/t), and
 $f(pH)$ = function of pH. (46) $\propto e^{1.57(pH-8.5)}$ Ammonia loss rate constant (Stratton 1969). (47) $\propto e^{0.13(T-20)}$.

$$(48) \quad \frac{C_e}{C_0} = \frac{1}{1 + \frac{A}{Q} K \cdot f(pH)}, \quad K = \text{removal rate coefficient (1/t), and}$$

$$f(pH) = \text{function of pH}, \quad (49) \quad \propto_m = 10^{0.0413T - 0.944},$$

$$\propto_m = \text{maximum specific growth rate, } d^{-1}$$

$$K_N = \text{half - saturation constant for ammonium nitrogen, mg/L, } T = \text{water temperature } ^\circ\text{C.} \quad (50) \quad K_N = 10^{0.015T - 1.158},$$

$$(51) \quad \propto = \propto_m \frac{N}{K_N + Nk_{O_2} + O_2} [1 - 0.83(7.2 - pH)]$$

\propto = growth rate N = Concentration of dissolved oxygen, mg/L

O_2 = concentration of dissolved oxygen, mg/L,

$$K_{O_2} = \text{half saturation constant for DO, mg/L,} \quad (52) \quad \frac{X_{N1}}{X_{N2}} = \frac{Q_R}{Q + Q_R}$$

Estimated ratio of Nitrifier,

Q = average flow rate through the basin, m^3/d ,

Q_R = recycle flow rate, m^3/d

X_{N1}/X_{N2} = ratio of nitrifier biomass Calculating the proportion of aerobic solids retention time. All nitrification was thought to take place in Cell 2. It is necessary to create a suggested operating schedule that includes four one-hour settling and four one-hour discharging periods across the 24-hour cycle. For nitrifiers, the aerobic portion of the solids retention period will be 16/48, or 0.33.

$$(53) \quad \theta_s = \frac{F_s}{f_{O_2} \left[1 + \frac{X_{N1}}{X_{N2}} \right] \infty} \quad \theta_s = \text{solids retention time, } d,$$

F_s = safety factor, f_{O_2} = fraction of solids retention time that

, X_{N1}/X_{N2} = ratio of nitrifier biomass concentration in cell 1

$$\text{to that in cell 2,} \quad (54) \quad X_{H2} = \frac{Y_H(S_0 + X_{SO}) + F_1}{V_1/Q} \left[\frac{Q_s(Q + Q_R - V_1)}{Q_R + (V_2/V_1)(Q + Q_R)} \right],$$

X_{H2} = heterotrophic biomass concentration in cell 2, mg/L,

Y_H = heterotrophic growth yield, mg biomass/mgCBOD₅,

S_0 = soluble CBOD₅ of unwanted waste water, mg/L

X_{SO} = particulate CBOD₅ of untreated wastewater, mg/L,

F_1 = solids decay factor [Table Rich(1999)],

$$(55) \quad X_{H1} = \frac{Q_R X_{H2} + Q Y_H (S_0 + X_{SO}) F_1}{Q + Q_R} \quad \text{for}$$

X_{H1} = heterotrophic biomass in cell 1, mg/L and

$$(56) \quad X_{i1} = Q X_{io} \frac{V_2 + Q_R \theta_s}{Q_R V_1 + (Q + Q_R) V_2} \quad \text{when}$$

X_{io}, X_{i1}, X_{i2} = inert suspended solids concentrations (both organic and inorganic) in the influent wastewater, cell1 and

$$\text{cell2, respectively, mg/L.} \quad (57) \quad X_{i2} = \frac{\theta Q X_{io} - V_1 X_{i1}}{V_2} \quad \text{when}$$

$$(58) \quad X_{T1} = X_{H1} + X_{i1}$$

X_{T1}, X_{T2} = total MLS in Cells 1 and 2, respectively.

$$(59) \quad X_{T2} = X_{H2} + X_{i2}, \quad (60) \quad P = 0.004X + 5$$

(for $X \leq 2000$ mg/L), P = power level, W/m^3

$$(61) \quad P = 8.125 \ln X - 48.75, \quad (\text{for } X \leq 2000 \text{ mg/L}),$$

X = total suspended solids concentration, mg/L,

$$(62) \quad Q_a = 2.257 \times 10^{-3} + 0.244 \times 10^{-6} X - 8.482 \times 10^{-10} X^2$$

Q_a = air flow rate at standard conditions, $m^3 \text{ air}/m^3 \text{ min}$

X = total suspended solids concentration, mg/L.

$$(63) \quad k_{d20} = 0.48 \theta_s^{-0.415} k_{d20} = \text{specific decay rate at } 20^\circ\text{C, } d^{-1}$$

$$(64) \quad k_d = k_{d20} (1.05)^{T-20} K_d = \text{specific decay rate at } T^\circ\text{C, } d^{-1}$$

$$(65) \quad R_{O21} = 4.16 \times 10^{-5} Q [1.47(S_0 + X_{SO}) - 1.42Y_H(S_0 + X_{SO})F1]$$

$$(66) \quad R_{O22} = 4.16 \times 10^{-5} Q [4.57(N_0 + N_2) + 1.42V_2K_dX_{H2}]$$

$$(67) \quad P_{O2} = 10^3 \frac{R_{O2}}{NV} \quad P_{O2} = \text{power intensity, } W/m^3$$

$$(68) \quad N_e = N_0 e^{-K_T[t+60.6(pH-6.6)]} \quad pH = 7.3e^{0.0005ALK}$$

N_e = effluent total nitrogen, mg/L

, N_0 = influent total nitrogen, mg/L,

K_T = temperature depending on rate constant. based on (Reid

$$\& \text{Streibin 1979)} K_T = K_{20}(\theta)^{(T-20)}$$

$$K_{20} = \text{rate constant at } 20^\circ\text{C} = 0.0064, \quad \theta = 1.039,$$

t = detention time in system, d,

pH = pH of near surface bulk liquid.

$$(69) \quad N_e = \frac{N_0}{1 + t(0.000576T - 0.00028)e^{(1.080 - 0.042T)(pH - 6.6)}}, \quad \text{According to}$$

(Middlebrooks & Pano 1983) research

N_e = effluent total nitrogen, mg/L

N_0 = influent total nitrogen, mg/L,

T = temperature of pond water, degrees C

pH = pH of near surface bulk liquid

6.1. Chemical coagulation and flocculation

The procedure typically involves the use of chemical coagulation and flocculation in the treatment of old and stabilised landfill leachates, as well as in the treatment of wastewater (Abdel-Shafy & Mansour 2020; Assou et al. 2016). It is effectively used as a pre-treatment, either before the reverse osmosis phase or to eliminate organic contaminants that are not biodegradable. Ferrous sulphate, ferric chlorosulfate, ferric chloride, lime, and aluminium sulphate are the most often used coagulants (Abdel-Shafy 2015; Ghafari et al. 2009). According to the findings, bio-flocculants are a viable alternative to conventional inorganic coagulants. A dosage of 20 mg/L of bio-flocculant was sufficient to remove over 85% of humic acid and 90%

of heavy metals (Abdel-Shafy 2015). Process optimisation evaluated: pH effect evaluation, optimal experimental conditions, and selection of the most suitable coagulant (Ghafari et al. 2009). Iron salts were reported to yield COD reductions of up to 5%, while aluminium salts or lime produced moderate corresponding values of 10 to 40% (H. I. Abdel-Shafy et al. 2022). Enhancing the flocculation rate through the combination of coagulants or the coexistence of flocculants and coagulants could result in a 50% reduction in COD (Abdel-Shafy 2015). However, potential drawbacks include the consistent production of sediment volume and a concurrent rise in the concentration of aluminium or iron in the liquid phase.

7. Stirred tank Bioreactors aerobic treatment

Rather than utilising a single stirred tank bioreactor, it is recommended to employ a series of smaller stirred tanks, maintaining the same total volume as the single bioreactor. This configuration, which consistently enhances the overall bioreactor performance, is implemented in the activated sludge process through a technique known as step aeration (Rao & Subrahmanyam 2004), (Metcalf et al. 1991). The aerobic tank is segmented into multiple compartments, each receiving a separate burst of compressed air. The untreated wastewater enters the first compartment, with partially treated water flowing sequentially through subsequent compartments, and the final treated effluent being discharged from the last compartment. Whilst this approach yields improved BOD reduction, it also incurs higher operational costs. As each compartment is relatively small and independently aerated, its performance may approach ideal behaviour (100% back-mixing). In contrast, a single large aerobic tank may contain dead zones and bypass streams, which disrupt back-mixing and negatively impact bioreactor performance.

To further enhance the bioreactor (aerobic tank) performance, a series-parallel arrangement of stirred tanks can be employed after research conducted by (Narayanan 2011) as well as (Leslie Grady et al. 2009). In this configuration, the aerobic tank is again divided into multiple compartments, with each receiving a portion of the raw wastewater and separate aeration. This method incorporates both step feeding and step aeration. Each compartment, except the first, receives a fraction of the fresh feed alongside partially treated effluent from the preceding compartment. This arrangement is particularly suitable for large-capacity installations. Here too, each compartment can function equivalently to an ideal continuous stirred tank reactor (CSTR), facilitating thorough contact between the substrate and biocatalyst (microbial cells). The performance equation for each compartment then becomes:

$$(70) \tau = (V/Q_0) = (C_{S0} - C_{Se})/(-r_{se}) \quad \tau = \text{space time, s,}$$

$V = \text{reactor Volume, m}^3$ $Q_0 = \text{flow rate of substrate (feed), m}^3/\text{s}$

$C_{S0} = \text{substrate concentration in product solution, gL}^{-1} \text{ or } M$

$C_{Se} = \text{substrate concentration in feed solution, gL}^{-1} \text{ or } M$

$C_{Sp} = \text{substrate concentration at biofilm liquid interface}$

$, \text{gL}^{-1} \text{ or } M - r_{se} = \text{Monod type kinetics, there is internal decay,}$

There is internal decay (albeit slight) and the bioconversion (BOD destruction) proceeds according to monod-type kinetics.

$$(71) (-r_{se}) = [(\mu_m/Y_e)C_{Se}x_e/(K_s + C_{Se})] - (k_d x_e),$$

$\mu_m = \text{maximum specific growth rate, s}^{-1}$

$Y_e = \text{true yield coefficient, } x_e = \text{biomass in product solution,}$

$K_s = \text{kinetic constant, gL}^{-1} \text{ or } M$

$K_d = \text{endogenous decay coefficient, s}^{-1}$ Equation (70) becomes:

$$(72) \tau = (C_{S0} - C_{Se})(K_s + C_{Se})/F(C_s, x),$$

$$(73) F(C_s, x) = (\mu_m/Y_e)(C_{Se}x_e) - (K_s + C_{Se})(K_d x_e),$$

$$(74) (-r_s) = (\mu_m/Y)(C_s x)/(K_c x + C_s),$$

$K_s = \text{Contois kinetic constant}$

$Y = \text{overall yield coefficient mg mg}^{-1}$, (Zabot et al. 2011) for dairy wastes, (75) $\mu = [(\mu_m C_s)/(K_s + C_s)][K_N/(K_N + C_N)]$

Reserahced by (Narayan et al. 2005),

$K_N = \text{Liquid phase oxygen coefficient, g L}^{-1}$

The exceptionally high reactivity of nascent oxygen serves as the primary safeguard in LPO utilisation. Precise addition of hydrogen peroxide is crucial, as even a slight excess could destroy microbial cells, which explains the slow commercial adoption of LPO technology. Despite this, the H2O2 requirement remains low (5–7 M). It is feasible and often recommended to combine Membrane Based Technology with activated sludge processes. After preliminary treatments such as lime addition, coagulation, screen filtration and clarification, the wastewater can be fed into a reverse osmosis (RO) unit, producing reusable water as permeate. The RO concentrate then undergoes biological treatment in aerobic tanks and denitrification bioreactors. (Smith 1970) has documented a successful case study demonstrating enhanced BOD, phosphorus and nitrogen removal through the integration of RO with aerobic processes. (Narayanan 1993) has provided an economic analysis of this approach. The primary factors affecting the overall economy of RO systems are the operating pressure of the RO unit and the lifespan of the polymeric membrane, with membrane clogging and fouling presenting additional challenges. (Narayanan 1993) reported that by recovering two-thirds of the wastewater in the RO unit and subjecting the remaining third to biological treatment, the overall cost of treated water production could be reduced to three-quarters of conventional methods, including membrane replacement costs. Laboratory studies by (Thakura et al. 2015) have shown that employing a forward osmosis unit upstream and a nanofiltration unit downstream can achieve

high chemical oxygen demand (COD) removal (exceeding 97%) from pharmaceutical wastewaters. However, the overall economic viability of this proposal requires analysis, considering the high operating costs of nanofilters and the substantial volume of wastewater typically handled in industrial settings.

8. Stirred tank bioreactors for anaerobic waste treatment

Similar to aerobic waste treatment study conducted by (Narayanan 2012), stirred tank bioreactors remain among the most popular choices for anaerobic processing of industrial, domestic and municipal waste, primarily due to their large capacity and straightforward installation. Anaerobic biological waste treatment, particularly when utilising a diverse culture of acidogenic, acetogenic and methanogenic microorganisms, offers the added benefit of converting organic matter into valuable products like biogas, a mixture predominantly composed of methane and carbon dioxide. The resulting anaerobic digested sludge can be utilised directly as a low-grade nitrogenous biofertiliser or employed in the production of phosphatic biofertiliser (known as Phosphate Rich Organic Manure) through biochemical means through (Sekhar n.d.) research.

However, the anaerobic digestion process is comparatively slower. Furthermore, methanogenic microbes, being obligate in nature, are highly sensitive to the medium's operating temperature and pH, with optimal conditions being pH= 7.0 and T=330–35°C. Anaerobic digestion can also be conducted at higher temperatures (55–65°C) using thermophilic microbes, which accelerates pathogen destruction but incurs additional costs for heating pipe installation and external heat supply. The expense of extra energy input often negates the advantages of faster pathogen elimination and increased methane production. Moreover, thermophilic microbes generally grow more slowly than mesophilic ones. Unless waste heat is accessible, such as in Combined Heat and Power systems, thermophilic waste treatment is unlikely to be an attractive or beneficial option. Nevertheless, a thermophilic pretreatment may be applied to the feed slurry if pathogen destruction is a significant concern (Narayanan 2011).

(72) $(-r_s) = (\mu/Y)x$ (Graef & Andrews 1974) demonstrated for kinetics,

$$(73) \mu = (\mu_m C_s) / [K_s + C_s + C_s^2 / K_{Si}],$$

K_{Si} = substrate inhibition coefficient, $g L^{-1}$,

C_A = concentration of acetic acid in solution, (74) $\tau = (1/\mu)$,

$$(75) \mu = (\mu_m C_A) / [K_s + C_A + C_A^2 / K_a],$$

K_a = ionisation constant of acetic acid,

K_s = kinetic constant, $g L^{-1}$

, C_A = concentration of unionised acetic in solutions, g/L ,

$$(76) C_A = (K_1 C_{gl} C_{Se}) / [K_a (C_{N0} - C_{Se})], \quad \text{since}$$

K_1 = equilibrium constant of CO_2 dissolved,

C_{Se} = total concentration of acetic in solutions

C_{gl} = concentration of carbon dioxide in solutions,

C_{N0} = concentration of ammonia in solution.

And

$$(77) \tau = (C_{gl} / R_c),$$

Y_c = yield coefficient for the production of CO_2 , $mol mol^{-1}$,

$$(78) \tau = C_{gl} / [Y_c (\mu x_e) - (k_L a) (C_{gl} - C_{gl}^*)],$$

K_L = liquid phase mass transfer coefficient, ms^{-1} ,

a = specific interfacial area of mass transfer $m^2 m^{-3}$.

$$(79) \tau = C_{gl} / [Y_c (x_e / \tau) - (k_L a) (C_{gl} - He p_c)],$$

He = henry's law constant, $mol L^{-1} KPa^{-1}$

P_c = partial pressure of CO_2 in the gas space kPa

and

$$(80) Q(p_c / RT) = (k_1 a) (C_{gl} - He p_c) V,$$

Q = flow rate production of biogas m^3/s ,

C_g = molar concentration of CO_2 in the gas space ($C_g = p_c / RT$),

$$(81) Q p_g = Q (P / RT) = [(k_L a) (C_{gl} - He p_c) + Y_m (\mu x_e)] V \quad \text{since}$$

p_g = molar density of biogas, $mol m^{-3}$,

$$(82) (Y_m x_e / \tau) [p_c / (P - p_c)] = (k_L a) (C_{gl} - He p_c)$$

Y_m = yield coefficient for methane production, $mol mol^{-1}$

x_e = cell mass concentration in product solution $g L^{-1}$

8.1 Trickling filters leachate

The filters (refer to the figure below) have been evaluated for their efficacy in reducing biological nitrogen levels in municipal landfill leachates. As a cost-effective nitrification option, biofilters are highly desirable (Hussein & Mona 2019). Nitrification above 90% can be attained in laboratory or pilot aerobic on-site crushed brick filters at loading rates of 100 to 130 $mgNH_4^+ - NL^{-1} day^{-1}$ at 25 °C, and 50 $mg NH_4^+ - NL^{-1} day^{-1}$ at lower temperatures of 5–10 °C, respectively. Simultaneously, a removal efficiency of 97% for ammonia in a trickling filter was reported. In the aeration tank, the technique uses suspended carriers of porous polymers that move continuously while allowing a biofilm of the active biomass to form on the surfaces. The method's main benefits are increased biomass levels, reduced sensitivity to harmful chemicals, and sufficient sludge-settling times [41,52], in addition to significant ammonia and organic rejections in a single cycle. Furthermore, at high ammonia strength, no inhibition of nitrification was observed (Aziz, Aziz & Yusoff 2011; Schmidt et al. 2003). Granular activated carbon (GAC) can also be used as a porous surface to adsorb organic matter and provide favourable circumstances for enhanced biodegradation, according to reports. As a result, a stable equilibrium between the adsorption and biodegradation processes can be reached. By using an effective biological

AC fluidised bed, about 70% of the refractory organics could be eliminated. It was determined that it is feasible to decrease ammonia by 85–90% and COD by 60–81%.

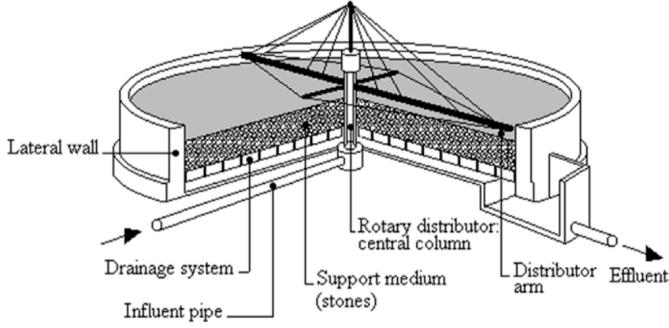


Figure 19: As a biological treatment method, trickling filters reduce nitrogenous chemicals in the leachate from municipal landfills.

Membrane packed bed biofilm technology (Bioreactor). A bioreactor in wastewater treatment is a chamber that provides a controlled environment for microorganisms to break down organic contaminants in wastewater with packed bed biofilm reactors for wastewater treatment:

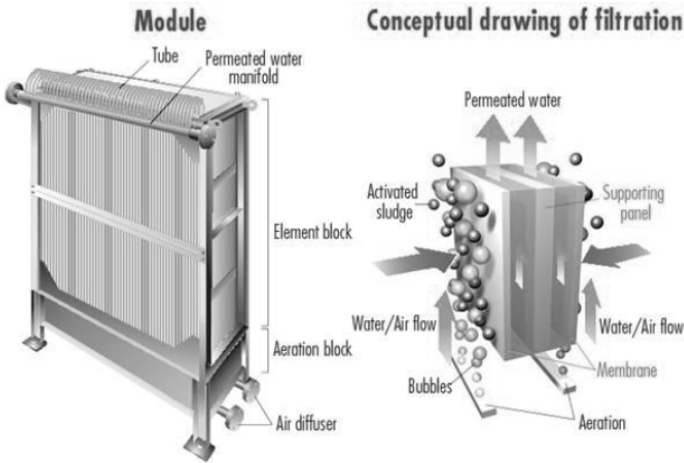


Figure 20 : Membrane bioreactor diagram.

A specially designed chamber that supports the growth of bacteria and algae, also known as biomass. The bioreactor regulates factors like temperature, pH, oxygen concentration, and nutrient supply to create a controlled environment. Microorganisms break down organic contaminants into less toxic compounds. A membrane module separates the treated wastewater from the microorganisms. The membranes are permeable to water molecules, but trap other pollutants like bacteria, viruses, and suspended particles. The microorganisms attached to the filter can be self cleaned through the injection of oxygen when needed.

The membrane separation process and the activated sludge process are combined in a membrane bioreactor (MBR).

Although subsequent clarification and tertiary processes like sand filtering are not required based on (Stephenson et al. 2000) research, the reactor functions similarly to a traditional activated sludge process.

Effluent and activated sludge are separated using low-pressure membrane filtration, either microfiltration (MF) or ultrafiltration (UF). In biofilm reactors, microorganisms develop in interconnected communities. These systems are characterised by multiple phases and handle diverse mixtures. With the exception of down-flow stationary fixed film (DSFF) bioreactors, which will be discussed subsequently, these reactors employ support materials such as silica granules, polymer beads, or activated carbon particles. Microbial cells form a biofilm that encases each of these particles. Within these bioreactors, these particle-biofilm complexes constitute the discrete phase. The diameter of each aggregate (d_{pm}):

$$(83) d_{pm} = (d_p + 2\delta) d_{pm} = \text{diameter of aggregate},$$

$$d_p = \text{diameter of support particle}$$

$$\delta = \text{biofilm thickness}, \quad (84) \rho_{sm} = f\rho_m + (1 - f)\rho_s$$

$$\rho_{sm} = \text{density of each aggregate},$$

$$f = \text{volume fraction of biofilm in aggregate},$$

$$\rho_s, \rho_m = \text{density of support particle and that of microbial cells respectively. Microbial cells multiply and divide within the biofilm; nevertheless, when the biofilm's thickness exceeds a certain threshold, } 0.3 < \delta < 0.5, (85) f = 1 - (d_p/d_{pm})^3,$$

$$(86) (-r_s) = \eta(-r_s)(int) 0.6 < \eta < 0.9,$$

$$(87) \eta = a - \tanh(\phi)[b \coth(n_d) - 1]/\phi \text{ according to (Leslie Grady et al. 2009).}$$

$$(88) \eta_d = (\sqrt{2}/\phi)((1 + \beta)/\beta\sqrt{\beta} - \ln(1 + \beta))\beta = (C_{sp}/K_s)$$

$$\phi = \text{Thiele - type modulus.} \quad (89) \phi = L^* \sqrt{\mu_m(app)/(D_e K_s)}$$

$$L^* = \text{characteristic dimension of particle biofilm aggregates.}$$

$$(90) L^* = (d_{pm}^3 - d_{pm}^2)/(6d_{pm}^2), \quad \mu_m(app) = (\mu_m/Y)x_f(1 - \epsilon_{pg}),$$

$$a = 1, b = \eta_d, \text{ if } \eta_d \leq 1, \quad a = \eta_d, b = 1, \text{ if } \eta_d \geq 1,$$

$$\epsilon_{pg} = \text{fractional gas holdup in fluidised bed,}$$

$$\epsilon_{pg} = \text{fractional liquid holdup in fluidised bed,}$$

$$(91) \epsilon_p = (\epsilon_{pL} + \epsilon_{pg}),$$

$$(92) (1/\eta)^2 = (1/\eta_d)^2 + \exp[\phi_b - (1/\eta_d)^2],$$

$$(93) \phi_b = 6\phi^2/\{5(1 + \beta)^2\}$$

$$(94) \phi_b = 6\phi^2/\{5(1 + \beta)^2\}$$

$$(95) \phi_b = 6\phi^2/\{5(1 + \beta)^2\}$$

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The membranes need to be cleaned periodically to maintain filtration performance. This is usually done weekly with chemical maintenance cleaning, and once or twice a year with recovery cleaning. The continuous filtration process eliminates a large portion of contaminants, ensuring that the treated water meets quality criteria.

Table 8: comparison of bioreactors.

| Bioreactors | Merits | Limitation |
|--|---|--|
| Stirred tank | Easy to build and use. makes use of suspended microbial growth. Both anaerobic and aerobic processes can use it. | Low capacity only |
| Trickle bed biofuel reactor | use of microorganisms' associated growth. The downflow mode of operation results in low operational costs. The rate of bioconversion is accelerated by a high concentration of cell mass in the biofilm. | mostly for the aerobic elimination of BOD. reduced capacity as a result of the constant low input flow rate. |
| Moving be biofilm reactor (slurry reactor) | a heterogeneous stirred tank variant. A high concentration of cells in the biofilm accelerates the pace of bioconversion. | less capable than column reactors in terms of capacity. The high rate of agitation could disturb the biofilm. |
| Fluidized bed biofilm reactor | Offers a high degree of bioconversion and operates at high volumes. The pressure drop over the bed doesn't rise when the feed flow rate increases once it has fully fluidized. Because of bed expansion, the degree of bioconversion rises as the feed flow rate increases. | Particle-biofilm aggregate entrainment loss may occur. Compared to trickling beds (packed beds), operating costs are higher. |
| Semi Fluidised bed biofilm reactor | Greater capacity and a lower reactor volume need for a higher degree of bioconversion (compared to fluidized beds). Even when the reactor volume remains constant, the degree of bioconversion rises as the feed flow rate increases. | more expensive to run than fluidized beds. It is impossible to operate in a continuous, circulating mode. |
| Inverse fluidised biofilm reactor | The downflow mode of operation results in low operational costs. Particles of a larger size could be employed. a respectably high level of bioconversion. | less capacity than a bed that is fluidized or semi-fluidized. greater need for reactor volume |
| DSFF bioreactor | Easy to build and use. No support particles are needed. The downflow mode of operation results in low operational costs. To improve capacity, more tubes or columns could be employed. | currently limited to anaerobic functioning. High capacity demands a large reactor volume. |
| UASB reactor | Easy to create. No particles of support were used. offers a significantly high degree of bioconversion at noticeably high capacities, even when the feedstock is strong. | limited to anaerobic procedures that use intricate microbial culture. excessively long starting time. |

8.2 Treatment membrane of lichate process

The ageing of landfill sites results in leachates that are more stabilised, which suggests the need for the implementation of more potent treatment alternatives. In order to achieve compliance with water quality regulations in numerous countries, membrane technology developed more influential treatments. Nanofiltration, microfiltration, ultrafiltration, reverse osmosis, and membrane bioreactor comprise the core membrane processes.

Membrane bioreactors (MBR): By integrating membrane separation technology with a bioreactor, these systems are extremely compact, resulting in high-quality effluent and minimal sediment production from high biomass concentrations (Chen et al. 2020; Abdel-Shafy & El-Khateeb

2011). The Ultrafiltration-biologically active carbon (UF-BAC) hybrid membrane bioreactor technology combines membrane filtration, adsorption, and biodegradation (Hussein & Mona 2019). The process's efficacy in terms of the total organic carbon reduction was within the range of 95–98%. Additionally, the organisms that degrade biodegradable materials at a sluggish pace, such as nitrifiers, are unlikely to be rinsed out of the process, in contrast to conventional methods (Abdel-Shafy & Abdel-Shafy 2017).

Microfiltration (MF): MF is effective for removing suspended particles, materials, and colloids. Therefore, it is regarded as advantageous when used in conjunction with chemical treatments or as a pre-treatment procedure for another membrane technique, such as RO, UF, or NF (Abdel-Shafy & Abdel-Shafy 2017) .

9. Mixing chamber flow analysis: Treatment methods and their impact on waste water (Anaerobic)

A study was conducted by (Abdelrahman et al. 2023) for wastewater treatment facilities to become more energy-efficient or even energy-neutral, biogas production from anaerobic sludge digestion is essential. A-stage treatment or chemically enhanced primary treatment (CEPT) in place of primary clarifiers are two examples of dedicated configurations that have been developed to maximise the diversion of soluble and suspended organic matter to sludge streams for energy production through anaerobic digestion. Because of the aeration energy demand, the A-stage design had the most energy consumption of the three, while CEPT had the highest operating expenses because of the use of chemicals. The utilisation of CEPT yielded the biggest energy surplus due to the highest percentage of recovered organic matter.

The preservation of ecosystems and public health depend on wastewater treatment. The primary purpose of wastewater treatment facilities (WWTPs) is to meet the necessary effluent criteria for nutrients and organics, which are expressed as chemical oxygen demand (COD) or biological oxygen demand (BOD). In addition, energy efficiency is becoming a bigger concern. In an effort to recover the energy and other resources contained in wastewater, WWTPs have even changed their name to water resource recovery facilities throughout the past ten years (Coats & Wilson 2017). Chemical energy (1.5 -- 1.9 kWh/m³ of wastewater) found in municipal wastewater is bound up in organic molecules' chemical bonds (Scherson & Criddle 2014). Moreover, wastewater itself (4.6–7.0 kWh/m³ of wastewater) might be regarded as a thermal energy source.

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(94) $B(t) = B_0 \cdot \exp\left\{-\exp\left[\frac{R_m \cdot \exp(1)}{B_0}(\lambda - t) + 1\right]\right\}$
 $B(t)$ = simulated cumulative methane yield (mL CH₄/gVS)
 $B(0)$ = simulated highest cumulative methane yield (mLCH₄/g, VS), R_m = maximum methane production rate (mL CH₄/g VS·d), λ = Lag phase, biogas production\ R_m = Maximum catabolic methane production.

(95) $COD_{min} = COD_{Inf} - COD_{eff} - COD_{sludge}$ when
 COD = (g/d).

(96) $COD_{methane} = \frac{Q_{methane}}{0.35} \cdot COD_{sludge} \cdot \left(\frac{VS}{COD}\right)_{sludge}$ as VS = influent sludge 0.35 theoretical methane production/g VS/COD = sludge ratio,

(97) $E_N = E_G - E_R - E_P - E_A - E_M - E_H$

E_N = Net energy recovery (Wh/m³)

E_G = recovery of methane in biogas,

E_R = energy consumption arm rotation in

clarifier, E_P = sludge pumping, E_A = aeration,

E_M = chemical tank mixing, E_H = Digester heating,

(98) $E_G = \frac{Q_m \cdot CV_m \cdot E \cdot 1000}{Q_{inf}}$

Q_m = m³/d Flow rate production, CV_m = (kWh/m³) Calorific energy, E = Heat and electric conversion efficiency

Q_{inf} = (m³/d) Influent wastewater flow theoretical (1000 from kWh to Wh). (99) $E_R = \frac{W \cdot r \cdot v \cdot 24}{e \cdot Q_{inf}}$ W = N/m Arm loading

factor r = m radius of tank (m), v = tip velocity (m/s), 24 for h/d, e = efficiency, (100) $E_P = \frac{Q_s \cdot H \cdot \rho \cdot g}{e \cdot Q_{inf} \cdot 3600}$ when Q_s = m³/d

Sludge rate production, H = pressure head and ρ = sludge density kg/m³, g = gravity m/s², 3600 = s/h.

(101) $E_A = \frac{COD_{Min} \cdot DO_{sat} \cdot 1000}{A \cdot E \cdot (DO_{Sat} - DO_{Dis}) \cdot Q_{inf}}$ E_A = (Wh/m³) aeration energy consumption, COD_{Min} = (KgCOD/d), DO_{Sat} = (kg/m³)

saturated concentration, DO_{Dis} = (kg/m³) dissolved oxygen,

$A \cdot E$ = (kgO₂/kWh). (102) $E_M = \frac{G^2 \cdot \mu \cdot V \cdot 24}{Q_{inf}}$ E_A = (Wh/m³)

mixing energy chemical G = (s⁻¹) gradient velocity

μ = (N·s/m²), V = m³ tank volume, 24 = h/d.

(103) $E_H = \left[\frac{Q_s \cdot (T_{AB} - T_{inf}) \cdot \rho \cdot C \cdot (1 - \phi)}{3600 \cdot Q_{inf}} \right] + \left[\frac{A \cdot (T_{AD} - T_{sur}) \cdot U \cdot 24}{Q_{inf}} \right]$ when

T_{AD} = temperature of anaerobic digest

T_{inf} = temperature of influent sludge, C = (J/kg·°C) specific heating capacity, ϕ = heat recovery efficiency

A = (m²) digester surface area, T_{sur} = (°C) surrounding temperature, U = (W/m²·°C) heat coefficient transfer

3600 = s/h, 24 = h/d,

(104) $C_{En} = C_{TSS} \cdot TSS + C_{COD} \cdot COD + C_N \cdot TN + C_P \cdot TP$

C = (kg/m³) concentration in the effluent primary unit TSS,

COD, TN and TP (\$/kg). (105) $C_F = C_p \cdot (1 + i)^n$

C_f = future value of (\$), C_p = present value (\$),

i = interest rate, n = number of years

To examine the destiny of influential COD, TN, and TP in each scenario, mass balances were set up (Figure 2). Of all the metrics, primary clarification had the lowest removal efficiency, whereas CEPT had the highest COD and TP removal efficiency. With a moderate TP removal rate of 32.2%, the A-stage's removal efficiency of COD (64.4%) and TN (22.8%) was comparable to CEPT. According to (Rahman et al. 2019), the influent entering sludge might contain 19–27% TN and 30–36% TP, which the A-stage could capture. Less COD was diverted to sludge for anaerobic digestion by the A-stage because 13% of the COD was lost through oxidation due to bacterial growth, which created CO₂. This value matched the oxidation values that (Ge et al. 2017) reported. wherein the COD loss from oxidation at various operational SRTs (0.5–3 days) was less than 25%. It is anticipated that the side stream will be impacted if an A-stage or CEPT is included in place of a primary clarifier. According to the COD mass balance, integrating A-stage and CEPT may recover more COD from the wastewater—37 and 67%, respectively—for later conversion into methane gas than primary clarity. Since the COD/TN ratio in the effluent was low (around 3), which is favourable for Anammox bacteria (usually 2–3), partial nitrification-Anammox technology with minimal aeration needs can be applied for the treatment of effluent of the A-stage and CEPT (Zhang et al., 2019).

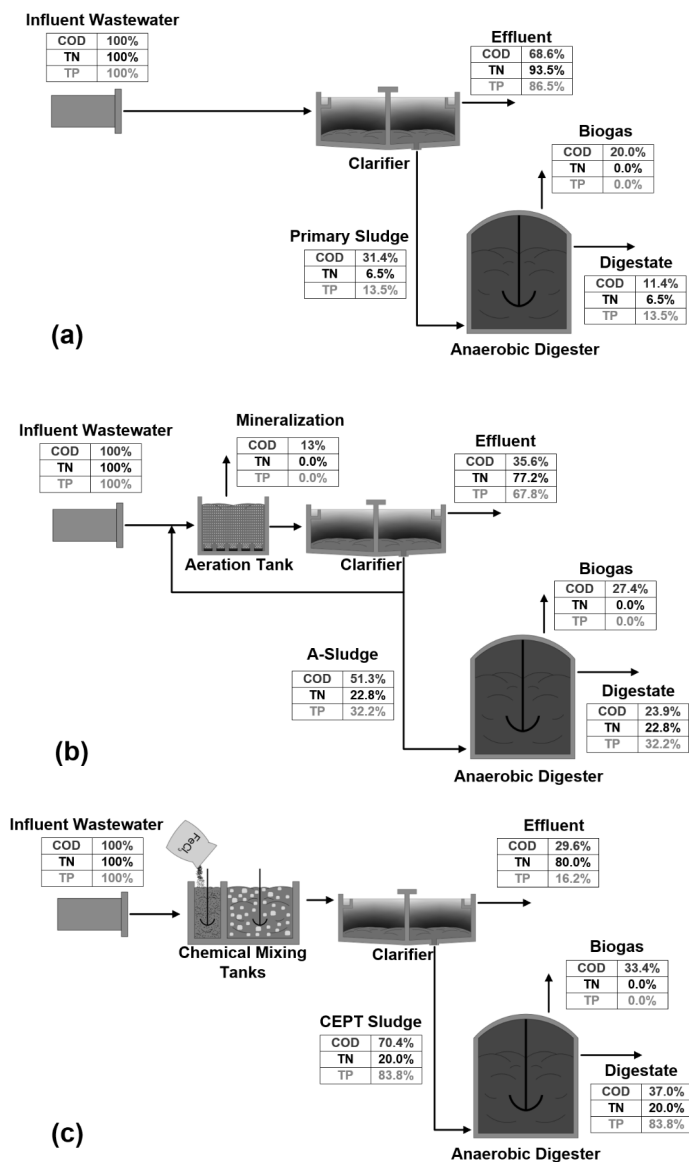


Figure 21 : COD, N and P mass balance: (a) primary clarification, (b) A-stage, (c) CEPT (Rahman et al. 2019).

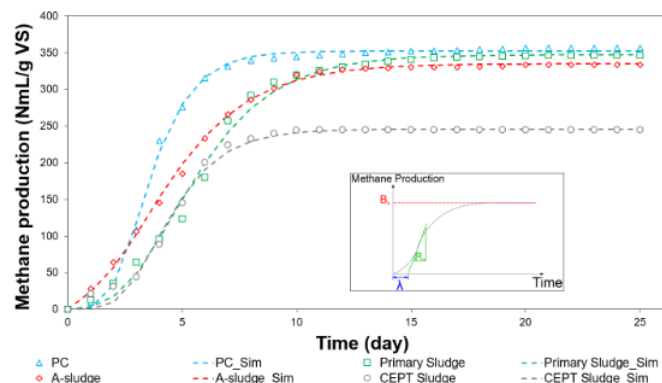


Figure 22: BMP results for each sludge after experimental stimulation (Rahman et al. 2019).

For every sludge, varying rates of methane generation and lag phases were noted (Figure 22). Consequently, rather than using a first-order rate model that was simplified, the modified Gompertz model was utilized to calculate the methane production (Kafle & Chen 2016). The updated Gompertz model ($R^2 > 0.95$ for all curves) provided a good fit for the methane production curves (Figure 3). For main, A-, and CEPT sludge, the average B_0 values were 347.3 ± 16.9 , 335.0 ± 5.2 , and 245.9 ± 5.5 mL CH₄/g VS, respectively. Out of all the sludges, the CEPT sludge had the greatest R_m (57.7 ± 0.6 mL CH₄/g VS/day) and λ (2.3 ± 0.1 day) (Figure 4). Primary sludge's kinetics were similar to those of CEPT sludge, with an average R_m and λ of 54.0 ± 2.0 mL CH₄/g VS/day and 2.2 ± 0.1 day, in that order. Table 1 shows that the digestion of A-sludge had the shortest lag phase (1.0 ± 0.0 day), which may be attributed to its comparatively high protein content. According to (Astals et al. 2014), proteins produced methane with a shorter lag time than fats and carbs. Compared to other sludges, A-sludge had a somewhat lower R_m (49.0 ± 0.3 mL CH₄/g VS/day).

| Characteristics | Primary Sludge | A-sludge | CEPT Sludge |
|-----------------------------------|------------------|------------------|------------------|
| <i>Physicochemical Parameters</i> | | | |
| TS (g/L) | 53.73 ± 0.55 | 11.14 ± 0.21 | 27.46 ± 0.16 |
| VS (g/L) | 23.35 ± 0.09 | 5.73 ± 0.06 | 14.61 ± 0.08 |
| VS/TS (%) | 43.5 ± 0.4 | 51.4 ± 0.4 | 53.2 ± 0.2 |
| COD (g/L) | 36.27 ± 0.07 | 10.20 ± 0.02 | 21.27 ± 0.41 |
| TN (%TS) | 2.06 ± 0.05 | 5.54 ± 0.21 | 3.20 ± 0.05 |
| NH ₄ -N (%TS) | 0.54 ± 0.02 | 2.04 ± 0.05 | 0.75 ± 0.02 |
| TP (%TS) | 0.48 ± 0.01 | 0.69 ± 0.01 | 1.00 ± 0.06 |
| pH | 5.82 ± 0.02 | 7.31 ± 0.01 | 7.25 ± 0.01 |
| <i>Organic fractions</i> | | | |
| VFA (mg/g VS) | 42.6 ± 1.4 | 17.3 ± 1.1 | 9.2 ± 0.1 |
| Proteins (mg/g VS) | 208.2 ± 7.8 | 403.4 ± 28.5 | 261.9 ± 15.9 |
| Lipids (mg/g VS) | 234.2 ± 10.0 | 158.6 ± 12.0 | 123.2 ± 11.5 |
| Soluble carbohydrates (mg/g VS) | 9.0 ± 0.3 | 8.4 ± 0.0 | 11.7 ± 0.8 |
| Cellulose (mg/g VS) | 195.5 ± 31.6 | 98.8 ± 17.0 | 41.7 ± 11.1 |
| Hemi-cellulose (mg/g VS) | 194.5 ± 14.9 | 121.6 ± 18.8 | 130.3 ± 25.5 |
| Lignin (mg/g VS) | 87.1 ± 6.5 | 50.0 ± 7.7 | 180.9 ± 19.1 |

Table 9: Sludge characteristics (Rahman et al. 2019).

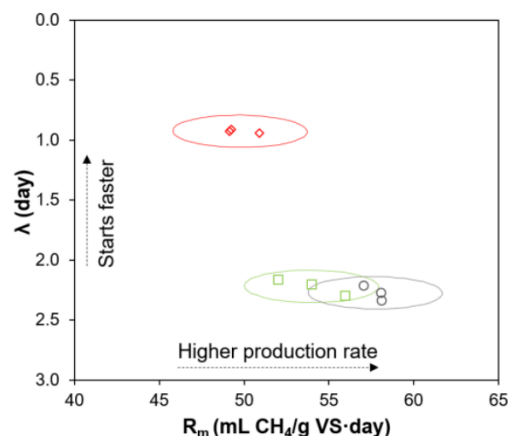


Figure 23: The relationship between the lag phase (λ) and the maximal methane production rate (R_m) for every sludge. Each sludge type's three identifiers denote triplicate samples. A 95% confidence level is represented by the circle.

9.1 Leachate treatment in mixing chamber

It has been demonstrated that leachate recirculation raises the moisture content of a reactor system, reducing methane output and COD while ensuring adequate distribution of nutrients and enzymes among methanogens and solid/liquids (Contrera et al. 2014; Deng et al. 2018). (Ghosh et al. 2017)c observed a 63 to 70% reduction in COD for the anaerobic pilot plant with recirculation. Recirculation has been demonstrated to reduce the stabilisation time to 2–3 years. However, elevated recirculation rates may adversely impact anaerobic decomposition. Leachate recirculation was also found to have the potential to hinder methanogenesis, which results in high organic acid levels (pH < 5), poisoning the methanogens (He et al. 2019; Ghosh et al. 2017). Additionally, issues including acidic conditions, saturation, and ponding may arise from extremely high volumes of leachate being recirculated (He et al. 2019; Ghosh et al. 2017).

10. Dual Media Filter

Access to clean water is crucial for human survival, ecosystem preservation, and societal well-being. The process of treating drinking water is intricate, involving various stages that are determined by regulations, contaminant elimination objectives, and associated expenses. In 1854, a breakthrough occurred when it was discovered that a cholera outbreak was transmitted through water. Areas with sand filters in place experienced less severe effects. British researcher John Snow (Belford 2013) identified that the primary cause of the epidemic was the contamination of a water pump by sewage. He utilised chlorine to sanitise the water, laying the groundwork for water disinfection techniques. This finding prompted governments to implement municipal water filtration systems, comprising sand filters and chlorination, marking the inception of public water regulation. Since that time, filtration has remained at the core of drinking water treatment, alongside disinfection, for more than a hundred years.

A study conducted by (Lund n.d.) in Scotland, water treatment primarily focuses on addressing pathogens and organic compounds, largely due to the region's landscape and prevalent livestock farming practices. This has led to a significant interest in enhancing treatment efficiency. Conventional filtration methods utilise granular media, such as sand, in either rapid or slow filters, depending on the applied flow rate. During the latter half of the 20th century, dual media configurations were introduced, incorporating an anthracite layer on top and occasionally a third thick gravel layer. Although these methods remain

widely employed globally and have demonstrated their reliability and effectiveness, recent legislative changes and a general drive towards greater efficiency have spurred research into alternative approaches. These new avenues of investigation encompass not only process modifications and the application of novel materials to improve treatment performance but also aim to reduce costs and, crucially, enhance energy efficiency.

10.1 Filtration Process

The efficacy of filters is determined by several physical properties, including grain size, shape, porosity, and the relationship between bed depth and media grain size. For filter bed design, a research conducted by (Kawamura & McGivney 2007) proposes utilising the L/de ratio, where L denotes the filter bed depth (mm) and de represents the effective size of the filter medium. This ratio fluctuates between 1000 and 2000 for various filter configurations, with specific ranges for different media types. The authors also recommend increasing L/de ratios by 15% to achieve filtered water turbidity under 0.1 NTU, and suggest conducting pilot studies when selecting filter depths for media exceeding 1.5 mm in size.

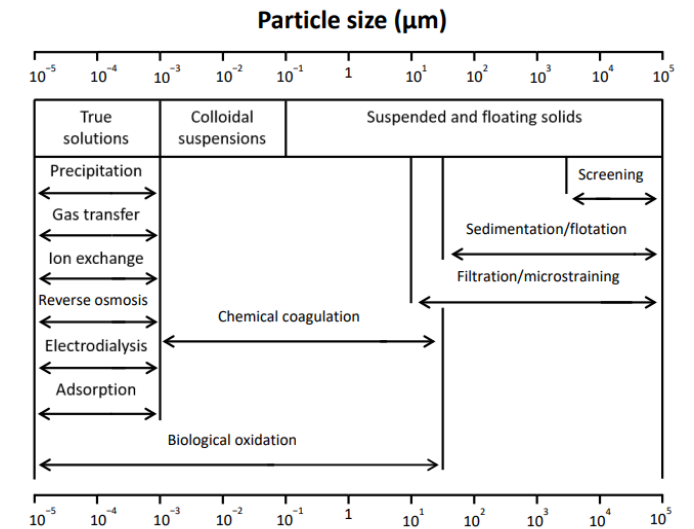


Figure 24: Treatment processes with respect to range of effectiveness (Tebbutt 1997).

Filtration processes can be categorised as either depth or cake filtration. Depth filtration involves particles being captured within the medium's pore network, while cake filtration results in a layer forming on the medium's surface through (Cheremisinoff 2019) research. Granular media filters predominantly function through depth filtration studied by (Gray 2010) . The filtration mechanism comprises two primary stages: transport, which moves particles towards the filter media, and attachment, which is contingent upon particle-surface interactions. These stages are not entirely discrete, as attachment mechanisms may

cause particles to deviate towards the grain surface through a study conducted by (Charles R. O'Melia n.d.). Some researchers argue that transport mechanisms exert a greater influence than surface forces (Ison & Ives 1969). Particle aggregation can also occur, forming clusters that are more readily transported and deposited. An additional stage, detachment, allows particles to re-enter the flow (Zamani & Maini 2009).

Research into deep bed filtration has conceptualised filter beds as assemblages of individual collectors, with efficiency calculated based on uniform spheres acting as collectors (Rajagopalan & Tien 1977). The removal at any given plane is a function of the number of collectors within that distance. This approach transforms the problem into one of particle transport and deposition onto individual grains, referred to as trajectory analysis or the microscopic model.

The analysis of particle trajectories is only applicable to a pristine filter; as particles accumulate, they alter the filter bed's characteristics and flow patterns. Deposited particles serve as additional collection points for subsequent particles (Amirtharajah 1988), necessitating their inclusion in efficiency calculations. Some researchers contend that these accumulated particles may be more effective collectors than the original filter grains (O'Melia & Ali 1979).

The filtration process comprises several phases (Figure 23). (Ison & Ives 1969) identify an initial clean filter stage, followed by a transitional phase. During this transition, filter performance initially improves (ripening), then stabilises during a working stage, before ultimately declining during breakthrough. The performance enhancement results from increased particle deposition, which eventually leads to higher velocities and reduced deposition. Breakthrough occurs when insufficient filter bed depth remains for particle removal, necessitating the termination of the filtration run. However, most researchers disregard the initial stage as atypical of average filtration runs, focusing instead on the three components of the transitional stage (Graef & Andrews 1974).

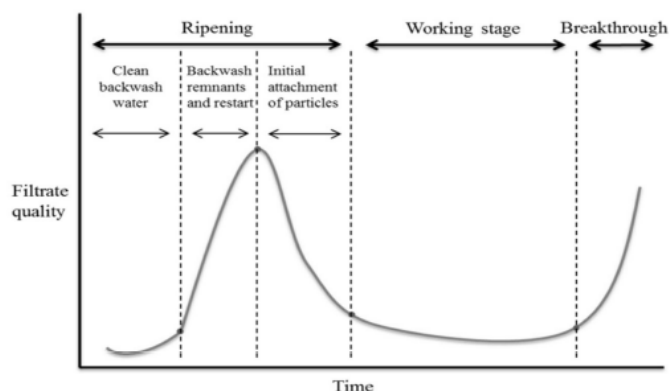


Figure 25: Cycle of filtration (American Water Works Association 2011).

11. Adsorption Sorption process

The primary parameter in adsorption operations is the empty bed contact time (EBCT) studied by (Stuetz & Stephenson 2009), which is calculated by dividing the volume of adsorbent by the flow rate. The EBCT for most water treatment applications is between 10 and 20 minutes. As long as the carbon at the bottom of the bed is not exhausted, the organic waste is normally removed at a rate greater than 99%.

Adsorption capabilities are high: typical commercial activated carbons may absorb up to 20% of their weight in organic compounds from water.



Figure 26: granular activated carbon tank.

Empty bed contact time: (Stuetz & Stephenson 2009) demonstrated $(106) EBCT = V/Qh$ when $V = \text{volume of GAC (m}^3\text{)} Q = \text{flow rate (m}^3\text{h}^{-1}\text{)}$

Adsorption is mostly used to remove organic materials from water and, occasionally, wastewater. Duties include removing taste, odour, colour, THM precursors, pesticides (before or after ozonation), natural organic matter, dechlorination, de-ozonation, solvents, and COD from industrial wastewater.

11.1 Chemical and physical treatment of leachate in adsorption

Activated carbon (AC) adsorption and biological treatment were the primary methods of effective landfill leachate treatment for a long time (Abdel-Shafy et al. 1998). This procedure has the potential to achieve a more substantial reduction in COD levels than chemical methods, regardless of the initial concentrations of the existing organic matter

(Da Silva et al. 2014; Aziz, Aziz, Yusoff, et al. 2011). Nevertheless, the primary drawback is the frequent regeneration of carbon columns. In the alternative, a high consumption of powdered AC may be implemented. The colour, inert COD, and non-biodegradable organics may also be reduced to acceptable levels in landfill leachate that has been biologically treated. A selective study demonstrated that AC possessed the highest adsorption capacities for the reduction of nonbiodegradable organic materials, with an 85% decrease in COD and a residual COD of 200 mg L⁻¹. The investigation involved the application of biological and adsorption treatments. The leachate was pre-treated through air stripping and coagulation–flocculation of ammonia, followed by a biological treatment in an aeration tank. The adsorbents used were 2 g L⁻¹ powdered AC and zeolite, and the conditions were under repeated fedbatch mode. The results were nearly 87% and 77% COD removals, respectively through the research of (Aziz, Aziz & Yusoff 2011; Da Silva et al. 2014). The pre-treatment procedure was also found to be effective in reducing metal concentrations [61] by filtration through granular carbon, prior to a conventional treatment. Additionally, it was demonstrated that limestone is capable of effectively removing metals from effluent by (Abdel-Shafy 2015; Abdel-Shafy & Mansour 2020).

11.2 Filtration operation setup

The configuration of filtration systems influences their effectiveness. Single sand medium filters often fall short in meeting treatment requirements, leading to the development of dual media filters. These employ a denser material at the bottom and a lighter one on top, with decreasing particle size. The most common arrangement features an anthracite layer above a sand layer, sometimes with an additional garnet layer (Ratnayaka 2009). While the filtrate quality is comparable to sand filters, dual media filters can operate 1.5–3 times longer at similar filtration rates. Efforts to increase filtration rates in dual configurations have been made, often involving coagulation aids. The size of coagulation-induced flocs is crucial; if too small, they may pass through the first layer and rapidly clog the sand layer, while overly large flocs could quickly obstruct the anthracite layer (Ratnayaka 2009). Anthracite's effective size is typically 1.5 mm, though this varies globally. The anthracite layer is usually 150–300 mm deep, with the sand layer at 450–600 mm (Twort et al. 2000), although (Belford 2013) suggests the reverse proportion.

(Zouboulis et al. 2007) conducted a comparative study of single medium sand and dual media sand/anthracite filters for conventional and direct filtration. In conventional filtration, the dual media setup operated for longer cycles, yielding 10% more water production. Dual media cycles lasted 2–3 times longer than single medium cycles, with final head loss values less than half (Figure 25a). Both

configurations achieved turbidity levels well below 0.2 NTU, with the dual media setup performing slightly worse (Figure 25b). Direct filtration proved more challenging to manage. With low coagulant doses, the single medium filter failed to meet the required turbidity level, showing values of 0.5–1 NTU. The dual media filter demonstrated greater efficacy, achieving turbidity values marginally higher than those in conventional operations (0.2–0.3 NTU) (Zouboulis et al. 2007).

An attempt was made to employ anthracite in coarser mono-medium filters with a deeper bed (1.8 m) in order to adopt greater filtration rates, but this approach was quickly given up by (Logsdon et al. 2006). The primary problems with anthracite are its high price and scarcity of global supplies; as a result, high-grade bituminous coal has occasionally taken its place (Ratnayaka 2009). In certain situations, other materials have been employed, such as GAC instead of sand or anthracite to eliminate smells. To boost the effectiveness of the GAC, it is more typical to add an adsorption step with a longer contact time after the filter (Logsdon et al. 2006).

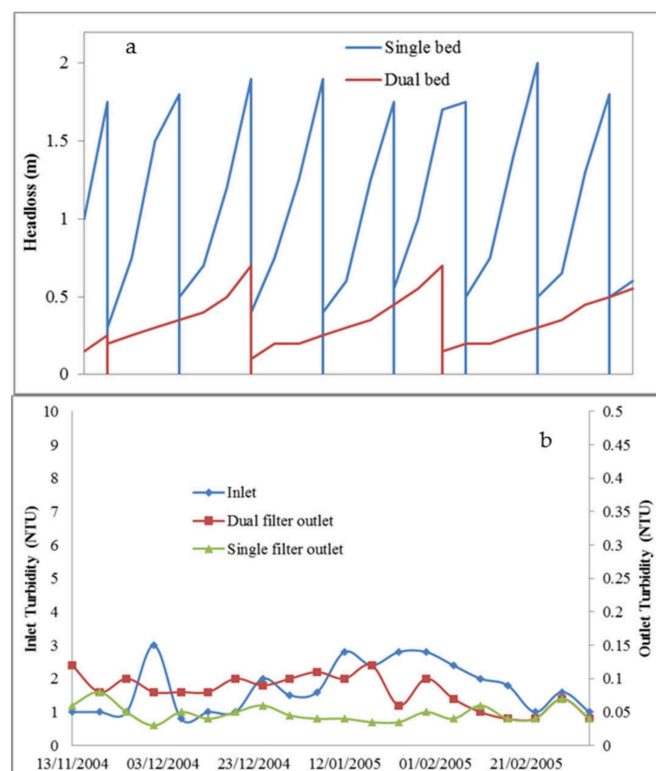


Figure 27: a- Head loss development, b- turbidity due to conventional treatment (Zouboulis et al. 2007).

A significant issue associated with backwashing is the elevated turbidity observed when the filter resumes operation. This period, known as ripening (Figure 23), occurs due to the loss of particles that aid filtration and the incomplete removal of flocs during the final stages of

backwashing study by (Slavik et al. 2013). Three distinct phases have been identified: initially, the filtrate is influenced by residual backwash water in the filter underdrains, followed by contaminants remaining above and within the filter bed.

Lastly, the filter's efficiency is compromised by the absence of additional particle retention capacity (Slavik et al. 2013). This problem has garnered increased attention as studies have shown that *Giardia* and *Cryptosporidium* cysts may be transmitted during this phase (Amirtharajah 1988). Various methods exist to mitigate this issue, including introducing coagulants to the backwash water or influent upon filter restart, implementing a terminal sub-fluidised rinse, allowing the filter to rest (delayed start), discarding the initial effluent (filter-to-waste), and adjusting filter rates (beginning with low rates and gradually increasing; termed slow start) (American Water Works Association 2011). It is considered typical for a filter to produce effluent with 0.5–1 NTU turbidity after restarting, which should then decrease to 0.2 or less within the first 30 minutes of operation and to 0.1 after an additional hour (Spellman 2008). (Logsdon et al. 2006) suggests a target of 0.3 NTU post-backwashing, dropping below 0.1 NTU within 15 minutes of resuming service.

11.3 Chemical and physical treatment of leachate in Nano-Filtration:

With a molecular cut-off of 200 to 2000 Da, nano-filtration (NF) polymeric membranes are utilised in NF technology. The rejection rate is high for both dissolved organic substances and sulphate ions. Conversely, the rejection rate for sodium and chloride is exceedingly low (Abdel-Fatah 2018; Amaral et al. 2016). The removal of 60–70% COD and 50% ammonia from landfill leachates can be achieved through NF treatment, regardless of the membrane material or design (i.e., tubular, spiral wound, or flat). This was achieved at a mean velocity of 3 m/sec and a trans-membrane pressure within the range of 6 to 30 bar (Amaral et al. 2016). The removal of 70–80% refractory COD was satisfactorily accomplished through the combination of physical methods and nanofiltration. However, membrane fouling necessitates highly effective control, typically caused by dissolved inorganic and organic matter, as well as colloidal and suspended particulates (Abdel-Shafy & Abdel-Shafy 2017).

12. Chlorination

The finest and least expensive disinfectant operator for deactivating germs and ensuring their persistence to assure their growth in the water supply network. Waterborne illnesses like cholera, typhoid fever, and dysentery were significantly reduced in assuring human health due to chlorine disinfection (Calderon 2000). Disinfection of municipal water with chlorine has provided significant community health benefits by controlling contagious

diseases; however, in raw water, the contact of natural organic matter (NOM) with chlorine produces chlorination disinfection by-products (DBPs), particularly trihalomethanes (THMs) and haloacetic acids (HAAs), which are of health concern (Bellar et al. 1974); (Rook 1972). DBPs' regular use in modest amounts may have a negative impact on human health, with a focus in the past on their cancer-causing nature. The generation of THMs in chlorinated water is determined by the raw water composition, operational characteristics, and residual chlorine in the water delivery network.

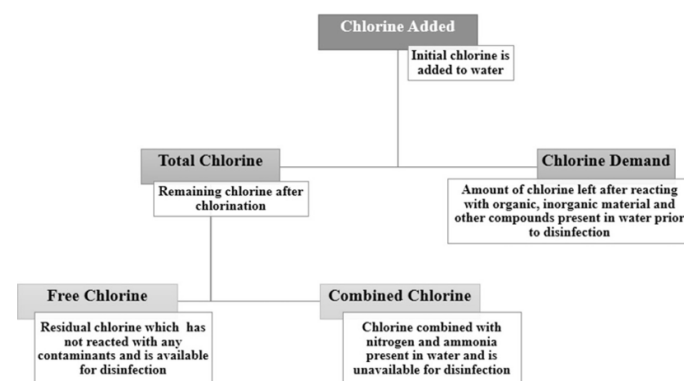


Figure 28: flowchart demonstrating chlorine addition (Ruyack 2019).

(Mazhar et al. 2020) demonstrated that when chlorine is added to water, a portion of the chlorine reacts with inorganic and natural elements and metals present and is not available for disinfection, which is known as the chlorine demand of water, and the remaining chlorine is known as total chlorine. Total chlorine is divided into mixed and free chlorine. When chlorine mixes with inorganic compounds such as nitrates and organic nitrogen-containing molecules such as urea, it functions as a weak disinfectant that is inaccessible for disinfection. The free chlorine is the residual chlorine that can be used to inactivate microorganisms; it is a measure of the water's potability. Thus, the sum of combined chlorine and free chlorine yields the total chlorine necessary. For example, when using completely clean water with no impurities, the chlorine demand is zero, and the combined chlorine demand is likewise zero because no inorganic or organic material is present in the water. In this method, the free chlorine concentration will be equal to the applied chlorine. Because of the existence of organic matter in surface water supplies, there will be a demand for chlorine, which will be met by inorganic chemicals such as nitrates. The free chlorine will be calculated as the sum of total and combined chlorine demand. (Figure 26) depicts a flowchart for the addition of chlorine.

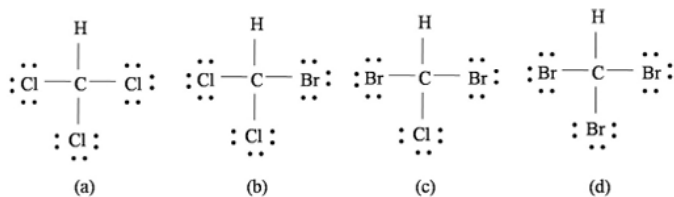
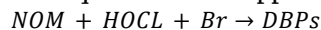


Figure 29: Chemical structures (a) chloroform (b) Bromodichloromethane (c) Dibromo-chloromethane (d) Bromoform.

The equation below applies to the production of DBPs:



(THMs and other halogenated products)

THMs, primarily comprising chloroform, dibromochloromethane (DBCM), bromodichloromethane (BDCM), and bromoform, are formed during chlorination when NOM precursors such as humic and fulvic acid react with chlorine (Thokchom et al. 2020). These DBPs are present in chlorinated water, with chloroform typically being the most prevalent THM. When bromide is oxidised in the presence of precursors, brominated THMs are produced. Some THMs are considered potentially carcinogenic to humans. Although chloramination generates lower THM concentrations compared to chlorination, it also produces cyanogen chloride as an additional DBP (Duong et al. 2003). Ozone can oxidise bromide to form hypobromous acid, a brominated THM precursor, whilst chlorine dioxide does not create THMs when reacting with organic precursors.

THMs are chemical compounds derived from methane CH_4 where three of the four hydrogen atoms have been substituted by halogens. The main THMs formed in drinking water due to chlorination are dibromochloromethane $CHClBr_2$, bromodichloromethane $CHCl_2Br$, and bromoform $CHBr_3$, with chloroform $CHCl_3$ being the most common. Their chemical structures are illustrated in the accompanying figure 27.

It is necessary to do extensive research to comprehend the chemistry of DBPs' occurrence in each situation in order to identify the best controlling approach. However (Mazhar et al. 2020) research have determined that, lowering the DBP level prior to, during, and following water treatment comes with a cost. Therefore, as a way to reduce the health risk associated with these DBPs, the recommendations values must be closely enforced in order to lower the permissible exposures or concentrations. There have been attempts to limit DBPs only by more stringent regulations.

12.1 Leachate in nano filtration

There is a molecular cut-off of 200 to 2000 Da for polymeric films to be used in NF technology. For both dissolved organic substances and sulphate ions, the rejection rate is considerable. Conversely, the rejection rate for sodium and chloride is exceedingly low (Abdel-Fatah

2018; Amaral et al. 2016). The removal of 60–70% COD and 50% ammonia from landfill leachates can be achieved through NF treatment, regardless of the membrane material or design (i.e., tubular, spiral wound, or flat). This was achieved at a mean velocity of 3 m/sec and a trans-membrane pressure within the range of 6 to 30 bar (Amaral et al. 2016). The removal of 70–80% refractory COD was satisfactorily accomplished through the combination of physical methods and nanofiltration (Amaral et al. 2016). However, membrane fouling necessitates highly effective control, typically caused by dissolved inorganic and organic matter, as well as colloidal and suspended particulates (Abdel-Shafy & Abdel-Shafy 2017).

13. Dissolved air flotation

Different flotation process types are available for a range of applications. Numerous industries have used the method, including potable water treatment (Childs et al. 1977; Nickols et al. 1995), wastewater clearing (Travers & Lovett 1985; Krofta & Wang 1982; Wang & Wang 2018), artificial recharge, and mineral processing (Rodrigues & Rubio 2007; Merrill & Pennington 1962). Flotation is essentially the act of employing air bubbles to separate solids from a body of liquid.

For more than a century, flotation has been employed in the mining and chemical processing sectors (Edzwald & Walsh 1992). But the origins of flotation are even more ancient. This method was employed more than 2,000 years ago by the ancient Greeks to separate minerals from gangue (Gregory & Zabel 1990; American Water Works Association (AWWA) 1990). It took several years for the process to evolve into its current, contemporary practices. In 1860, Haynes succeeded in separating minerals with oil, according to Kitchener (Kitchener 1984). He received a patent for his technique. Salman, Picard, and Ballot improved a method for separating sulphate granules from water in 1905 by incorporating air bubbles and a tiny bit of oil. It was referred to as "froth flotation." T. Hoover created the first flotation machine in 1910, and it wasn't all that different from what is used today. Caulk presented a novel technique known as "foam flotation" a few years later, in 1914 (Kitchener 1984). Air bubbles were introduced using submerged porous medium in this procedure. Actually, the mineral processing industry makes extensive use of froth and foam flotation methods, which are commonly referred to as dispersed air flotation systems. It was in 1904 when the electrolytic flotation procedure was developed. The Elmore brothers proposed the method after demonstrating that electrolysis could create floating bubbles. At the time, it wasn't used commercially.

In Scandinavia, Niels Peterson and Carl Sveen received a patent for dissolved air flotation in 1924 (Lundgren 1976). In the beginning, the paper industry employed it to recover fibres and white water. Dissolved air flotation (DAF) was

first used in the late 1960s to treat potable water and wastewater.

According to Edzwald and Walsh, DAF has been used for more than 20 years throughout Europe, particularly in Scandinavia, to clarify water (Edzwald & Walsh 1992). The first DAF plant for the clarification of potable water was built in 1965, and by 1988, there were 34 units operating, according to Heinanen's assessment on flotation's use in Finland (Heinanen 1988). Nonetheless, flotation was initially used for a water reclamation project in South Africa at the beginning of the 1960s (Longhurst 1987).

In 1976, the Glendye Treatment Works of the Grampian Regional Council in Scotland became the first full-scale water treatment plant in the United Kingdom to use this technology (Zabel & Melbourne 1978).

Flotation is a faster technique for solid-liquid separation than sedimentation, according to experiments conducted by specialists at the Water Research Centre (Packham & Richards 1975).

The decrease in pressure of an air-saturated water stream creates the bubbles in DAF. Pressure flotation is the most significant and frequently utilized of the three DAF types—vacuum flotation, microflotation, and pressure flotation—in the treatment of water and wastewater. Pressure flotation creates tiny air bubbles by dissolving air in water at high pressure and releasing it at atmospheric pressure via a needle valve or nozzle.

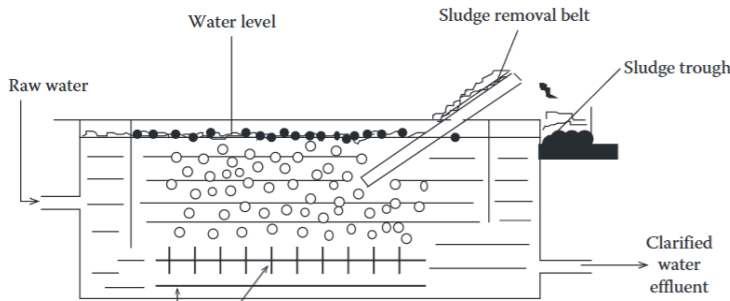


Figure 30: Kinetic foam flotation diagram through aid diffusion (Adlan 1998)

Bubbles that lower the density of the bubble-particle agglomerates cause the particles to float during the DAF process. Bubble-particle agglomerates will rise and float to the top as long as their density is lower than that of water (1.00 g/cm³). Compared to large particles, which need more bubbles to reduce density, small particles require fewer bubbles. The flotation tank's surface should be reached by the bubble-particle agglomerates (Adlan 1998). The cleared water is used to sweep out the agglomerates that do not rise to the surface. Stokes' law can be used to estimate the bubble-particle's increasing velocity.

The three primary DAF theories demonstrate that before using the DAF system for water or wastewater treatment, certain parameters impacting the system should be taken into account. Therefore, when constructing and using the

DAF system, all elements and system operation should be taken into consideration. Stokes law (Li & Lam 1964) determined the following model (107) $D = 6\pi a\mu U$, $D = \text{drag force (kN)}$, $a = \text{radius of bubble (m)}$, $\mu = \text{dynamic viscosity (kg/m/s)}$, $U = \text{terminal velocity (m/s)}$. (Packham & Richards 1972) model demonstrated the following:

$$(108) \frac{4}{3}\pi a^3 \sigma g = \frac{4}{3}\pi a^3 \rho g + 6\pi a\mu U,$$

$$U = \frac{2}{9}(\sigma - \rho)a^2 \frac{g}{\mu}, \quad a = \text{radius of sphere (m)},$$

$$\sigma = \text{density of sphere (kg/m}^3\text{)}$$

$$\rho = \text{density of liquid (kg/m}^3\text{)}. \text{ (Jameson 1984) demonstrated the following (109) } 6\pi\mu Ua = \frac{4}{3}\pi a^3 (\rho - \rho_g)g \text{ since}$$

$$\rho_g = \text{density of gas, negligible.} \quad (110) U = \frac{2\rho_g a^2}{9\mu}.$$

$$(111) C_D = \frac{\text{Force on bubble}}{1/2 U^2 \pi a^2} = \frac{4/3 \pi \rho g a^3}{1/2 \rho U^2 a^2} = \frac{4gd}{3U^2} \text{ by (Harper 1972)}$$

$$\text{when } C_D = \text{drag coefficient and (112) } C_D = \frac{24}{Re} \text{ for } Re = \text{Reynolds number} < 0.5. \text{ (Fukushi et al. 1995)}$$

$$(113) U = \frac{\rho g a^2}{3\mu} a < 2 \times 10^{-2} \text{ cm, (114) } \frac{ga^3}{3\nu^2} < 1 \quad \nu = \mu/\rho.$$

$$(115) R = 1 - e^{-\frac{V_r}{Q/A_h}}, \quad R = \text{ratio of total removal of solid concentration after flotation to the inflow} = 1 - C_0/C_i,$$

$$C_0 = \text{the effluent suspended solids}$$

$$C_i = \text{the influent suspended solids,}$$

$$V_r = \text{the rising velocity of a single particle/air Bubble (m/s)}$$

$$Q = \text{applied flow rate (m}^3\text{/s)},$$

$$A_h = \text{the horizontal area of the unit (m}^2\text{)}. (116) U = 25V^{1/6}$$

$$\text{since } V = \text{volume of the bubble. (117) } \frac{1}{2} C_D S \rho U^2 = \Delta \rho g V,$$

$$C_D = \text{the drag coefficient, } S = \pi d_e^3/6, \quad (118) C_D = \frac{4g\Delta\rho d_e^3}{3\rho d_h^2 U^2}$$

$$(119) U = \left(\frac{8gV}{\pi C_D d_h^3}\right)^{1/2} (120) U = \left(\frac{8g}{6^{2/3} \pi^{1/3} C_D}\right)^{1/2} V^{1/6} \frac{d_e}{d_h}$$

$$(121) C_D = \frac{24(1+0.173Re^{0.657})}{Re} + \frac{0.413}{1+16300Re^{-1.09}}$$

$$\text{for } Re < 1, C_D = 24/Re \text{ and } d_e/d_h = 1 = 1$$

Table 9: Bubble size relationship in between, rise velocity, temperature, and laminar flow(P. 1988)

| Bubble Size (μm) | Rise Velocity (m/h) Above Which Turbulent Flow Exists ^a | | Terminal Rise Velocity (m/h) Based on Stokes' Law | |
|----------------------------------|---|------|--|-------------------|
| | 4°C | 20°C | 4°C | 20°C |
| 10 | 565 | 360 | 0.125 | 0.196 |
| 20 | 283 | 180 | 0.499 | 0.783 |
| 30 | 188 | 120 | 1.12 | 1.76 |
| 40 | 141 | 90 | 2.00 | 3.13 |
| 50 | 113 | 72 | 3.12 | 4.89 |
| 80 | 70.7 | 45 | 7.99 | 12.5 |
| 110 | 51.4 | 32.7 | 15.1 | 23.7 |
| 120 | 47.1 | 30 | 18.0 | 28.2 |
| 130 | 43.5 | 27.7 | 21.1 | 33.1 ^b |
| 140 | 40.4 | 25.7 | 24.5 | 38.3 ^b |
| 160 | 35.3 | 22.5 | 31.9 | 50.1 ^b |
| 170 | 33.2 | 21.2 | 36.1 ^b | 56.5 ^b |

13.1 Application into landfill treatment of leachate

(Zouboulis & Avranas 2000) investigated the removal of humic acid from simulated leachate utilising the DAF technique. Although the DAF process has the potential to treat landfill leachate, no studies have actually used it in practice, according to the literature review. This signifies a deficiency in understanding DAF capabilities in leachate treatment. Considering the benefits provided by DAF, including increased hydraulic loading, reduced coagulant doses, and diminished sensitivity to flow fluctuations, among others, the DAF method represents a viable alternative for the treatment of landfill leachate. (Palaniandy et al. 2010; Adlan et al. 2011) recently conducted a study on the treatment of semi-aerobic landfill leachate utilising Dissolved Air Flotation (DAF).

Palaniandy et al.'s study proposes a method for large-scale application in landfill sites (Palaniandy et al. 2010). With and without alum coagulation, the use of DAF in leachate treatment is examined in this study. According to the study, a coagulation process needs to be implemented in order to help destabilise emulsions or colloidal particles. The DAF process in leachate treatment without coagulation demonstrates that the variations in the percentage removal of turbidity, colour, and COD were significantly minimal, indicating that the primary pollutants in the leachate comprised soluble organic and inorganic substances, including humic acid, fulvic acid, iron, sodium, potassium, sulphate, and chloride, (Adlan et al. 2011; Aziz et al. 2007). For colour, COD, and turbidity, the percentage elimination of the parameters under study rose to 70%, 79%, and 42%, respectively, with the addition of coagulant (alum). It is evident from this investigation that the DAF process is capable of treating landfill leachate. By employing response surface methodology (RSM) to optimise the coagulation and DAF process in semiaerobic landfill leachate, Adlan et al. have conducted additional study in this area. To cause coagulation, ferric chloride (FeCl_3) was selected for this study (Adlan et al. 2011). The findings indicate reductions of

50%, 75%, 93%, and 41% in turbidity, COD, colour, and $\text{NH}_3\text{-N}$, respectively.

13.2 Chemical and physical treatment of lechate in dissolved air flotation

Flotation has been widely used to reduce ions, colloids, humic acids, fibres, microbes, and macromolecules (Adlan et al. 2011; Dabaghian et al. 2018). To determine the best way to handle semi-aerobic liquid-landfill leachate, coagulation using FeCl_3 and dissolved air flotation (DAF) were coupled. Using central composite design (CCD) and response surface methodology (RSM), the investigated parameters—specifically, flow rate, pH, dosage of the FeCl_3 coagulant, and injection time—were optimised. Ammonia nitrogen ($\text{NH}_3\text{-N}$), colour, turbidity, and COD chemical oxygen demand were all removed to the greatest extent possible. For turbidity, COD, colour, and $\text{NH}_3\text{-N}$, the corresponding clearance rates were 50%, 75%, 93%, and 41%. Overall, the findings showed that pressure and flow rate had less of an impact on the removal of contaminants. However, in order to optimise the effectiveness and performance of the DAF system under study, these two elements are essential. These novel discoveries were immediately put into practice and applied to the commercial treatment of landfill leachate using such DAF [56]. On the other hand, flotation in columns was employed to eliminate non-biodegradable substances and residual humic acids from a landfill leachate that was replicated (Dabaghian et al. 2018). It is possible to remove about 60% of humic acids under ideal circumstances (Zouboulis et al. 2004; Lindamulla et al. 2022).

14. Aerobic digester

The idea behind all contemporary high rate biomethanation techniques is to immobilize bacterial sludge in some way in order to retain high viable biomass. One of the following techniques is used to accomplish these (Pol & Lettinga 1986): (Rajeshwari et al. 2000) found that anaerobic baffled reactors and upflow anaerobic sludge blanket reactors are examples of systems that combine the formation of highly settleable sludge aggregates with gas separation and sludge settling. adhesion of bacteria to high density particle carrier materials, such in anaerobic expanded bed reactors and fluidized bed reactors. Sludge aggregates become trapped between packing materials that are provided to the reactor, such as upflow and downflow anaerobic filters.

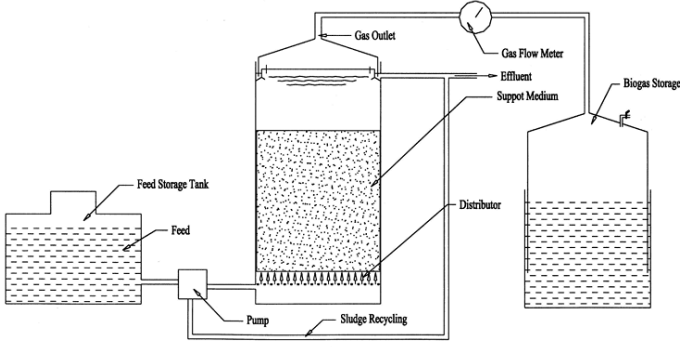


Figure 31: Fixed film digester.

The sludge can be evacuated through centrifugal techniques, a vacuum, a press, a horizontal band filter, a bore press, drying beds, and sludge lagoons. At the same time (Demirbas et al. 2017) mentioned a few systems like vacuum, press, and horizontal band filters are expensive to invest in and run, need specialized personnel, and require apparatus and equipment. After sludge dewatering, increasing the dry matter content from 20–25% to 90% results in a significant reduction in sludge mass. Energy is required for the drying process in order to dewater. 2,500 kJ/kg of water is the energy needed to evaporate it. This number is considered to be between 2,750 and 3,100 kJ/kg of water when heat losses are taken into account. The energy contents (MJ/kg) and organic dry matter contents (%) of dried sewage sludge and crude sludge are 8.4–11.5, 12.6–18.4, and 60–80, respectively (Fytli & Zabaniotou 2008).

14.1 Kinetic model development

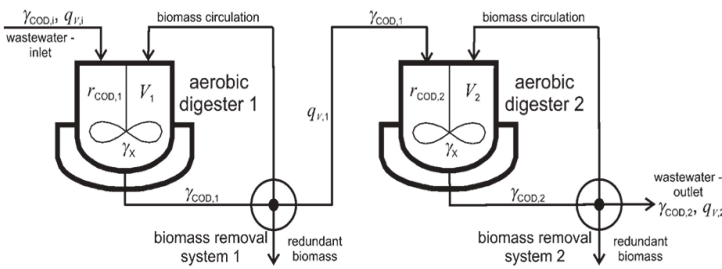


Figure 32: Two anaerobic digesters connected in series that are not equal (Goršek 2007)

Typically, the aeration basin (digester) is treated as a perfectly mixed vessel, and anaerobic sludge is viewed as a single pseudo-species with a growth rate that follows assumed dynamics. If we assume equal inlet and exit industrial wastewater volume flow rates ($q_{v,i} = q_{v,1} = q_{v,2} = q_v$), then the degradation rate of OM expressed as COD, using the CSTR model under steady-state

circumstances, may be given independently for anaerobic digester 1, by equation: for aerobic digester 1: (Bailey & Ollis 2018) demonstrated

$$(122) \quad -r_{COD,1} = \frac{1}{Y_{X/COD}} \mu Y_X = \frac{q_v}{V_1} (Y_{COD,i} - Y_{COD,1}) \quad \text{when}$$

$-r_{COD,1}$ = degradation rate of organic compounds (OM)

in aerobic digester 1, $Kg m^{-3} d^{-1}$,
 V_1 = volume of aerobic digester 1

μ = specific growth rate of biomass, d^{-1} ,

$Y_{X/COD}$ = Yield biomass degrading (OM), $Kg Kg^{-1}$,

$Y_{COD,i}$ = inlet mass concentration of OM, $kg m^{-3}$,

$Y_{COD,1}$ = outlet mass concentration of OM from digester 1, $kg m^{-3}$

Regarding aerobic digester 2 (Burhan et al. 2005) demonstrated the following model

$$(123) \quad -r_{COD,2} = \frac{1}{Y_{X/COD}} \mu Y_X = \frac{q_v}{V_2} (Y_{COD,1} - Y_{COD,2}) - r_{COD,2} \text{ is the degenerate rate of organic compounds (OM) in aerobic digester 2 in } Kg m^{-3} d^{-1},$$

V_2 = volume of aerobic digester 2,

Y_X = mass concentration biomass, $Kg m^{-3}$,

$Y_{COD,2}$ = outlet mass concentration of OM from digester 2, $kg m^{-3}$

. And (124) $\mu = \frac{\mu_{max} Y_{COD}}{K_s + Y_{COD}} \exp(-\frac{Y_{COD}}{K_{IA}})$ demonstrated by (Aiba et al. 1968) as

μ_{max} = maximum specific growth rate of biomass, d^{-1} ,

K_s = saturation constant of OM, $kg m^{-3}$

K_{IA} = Aiba's inhibition coefficient, $Kg m^{-3}$,

$Y_{X/COD}$ = Total OM concentration,

$$(125) \quad -r_{COD,1} = \frac{r_{max} Y_{COD,1}}{K_s + Y_{COD,1}} \exp(-\frac{Y_{COD,1}}{K_{IA}}) = \frac{q_v}{V_1} (Y_{COD,i} - Y_{COD,1})$$

Additionally, regardless of q_v , which is attained with biomass circulation, X in both digesters remains constant. Given equation (124) and the aforementioned presumptions, equations (122) and (123) can be defined as follows

$$(127) \quad -r_{COD,2} = \frac{r_{max} Y_{COD,2}}{K_s + Y_{COD,2}} \exp(-\frac{Y_{COD,2}}{K_{IA}}) = \frac{q_v}{V_2} (Y_{COD,1} - Y_{COD,2}),$$

$$(128) \quad V_1 = \left(\frac{r_{max} Y_{COD,1}}{K_s + Y_{COD,1}} \exp(-\frac{Y_{COD,1}}{K_{IA}}) \right)^{-1} \cdot q_v (Y_{COD,1} - Y_{COD,1})$$

r_{max} = maximum degradation of OM, $kg m^{-3} d^{-1}$,

$$V_2 = \left(\frac{r_{max} Y_{COD,2}}{K_s + Y_{COD,2}} \exp(-\frac{Y_{COD,2}}{K_{IA}}) \right)^{-1} \cdot q_v (Y_{COD,1} - Y_{COD,2}),$$

$$(129) \quad \frac{dr_{COD}}{dY_{COD}} = \left(\frac{r_{max}}{K_s + Y_{COD}} - \frac{r_{max} Y_{COD}}{(K_s + Y_{COD})^2} \right.$$

$$\left. - \frac{r_{max} Y_{COD}}{(K_s + Y_{COD}) K_{IA}} \exp(-\frac{Y_{COD}}{K_{IA}}) \right) = 0 \quad \text{determined by (Livenspiel 1972),}$$

Typically, the curve $-r_{COD} = f(COD)$ has a certain convex form when OM inhibition is present. Consequently, the curve's maximum follows the equation, adopting Aiba's

$$\text{formulation. (130) } \eta_{COD} = \frac{Y_{COD,i} - Y_{COD}}{Y_{COD,i}} \cdot 100$$

Table 10 lists the measured outlet mass concentrations of OM and COD at various industrial wastewater volume flow rates, q_v , under steady-state circumstances. Confidence intervals (CF), corresponding dilution rates (D), OM and COD degradation rates, and wastewater treatment efficiencies (COD) were computed. The difference between $COD_i = (5.10 \pm 0.10) \text{ kg m}^{-3}$ and COD is multiplied by D to get the $-r_{COD}$. Additionally, the following formula was used to determine the COD:

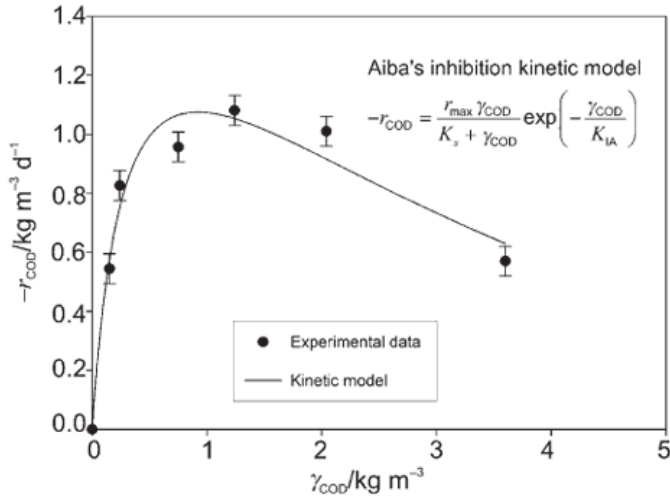


Figure 31: Model Base values of degradation of OM (Goršek 2007).

Over 97% of the organic matter (OM) found in the industrial wastewater at $D = 0.11 \text{ d}^{-1}$ was eliminated by the aerobic biomass sludge (see Table 10 6's sixth column). Additionally, the mass fraction of plant tannins with respect to overall outflow remained nearly the same as that with reference to inflow ($W_T = 30\%$), as demonstrated by the UV spectrophotometric method¹⁸ for chemical analysis of tannins. Thus, it can be said that the biomass sludge from the current wastewater treatment plant has a good capacity to break down plant tannins in industrial wastewater. However, as the dilution rate increases, the wastewater treatment efficiency drops off significantly. At $D = 0.28 \text{ d}^{-1}$, where the wastewater efficiency was only 76%, the maximum rate of OM degradation was attained.

Table 10: Measured outlet mass concentrations of OM and COD

| $q_v/\text{m}^3 \text{ d}^{-1}$ | D/d^{-1} | $\gamma_{COD}/\text{kg m}^{-3}$ |
|---------------------------------|-------------------|---------------------------------|
| 0.0010 | 0.11 ± 0.01 | 0.150 ± 0.02 |
| 0.0015 | 0.17 ± 0.01 | 0.240 ± 0.02 |
| 0.0020 | 0.22 ± 0.01 | 0.750 ± 0.02 |
| 0.0025 | 0.28 ± 0.01 | 1.240 ± 0.06 |
| 0.0030 | 0.33 ± 0.01 | 2.040 ± 0.07 |
| 0.0035 | 0.38 ± 0.01 | 3.600 ± 0.08 |

| $-r_{COD}/\text{kg m}^{-3} \text{ d}^{-1}$ | CF% | $\eta_{COD}/\%$ |
|--|-----|-----------------|
| 0.544 ± 0.051 | 9.3 | 97.1 ± 2.8 |
| 0.826 ± 0.052 | 6.2 | 95.3 ± 2.7 |
| 0.957 ± 0.050 | 5.3 | 85.3 ± 2.8 |
| 1.081 ± 0.051 | 4.7 | 76.7 ± 2.7 |
| 1.010 ± 0.051 | 5.0 | 60.0 ± 2.7 |
| 0.570 ± 0.051 | 8.9 | 29.4 ± 2.6 |

(Goršek 2007) demonstrated that the mass balance of OM data from studies in the laboratory bench-top aerobic digester was effectively used to calculate the kinetic parameters of the Aiba's inhibitory kinetic model. Then, using the criterion of a minimal total holding time, we calculated the ideal volumes of two aerobic digesters connected in series. Industrial wastewater volume flow rate ($q_v = 120 \text{ m}^3 \text{ d}^{-1}$) and wastewater treatment efficiency (COD = 98%) served as the basis for the evaluation. Two aerobic digesters with $V_1 = 467 \text{ m}^3$ and $V_2 = 228 \text{ m}^3$ were produced under these circumstances. A two-stage industrial wastewater treatment facility's technological and financial approval was primarily validated by the fact that its total digester volume was more than twice that of a one-stage plant. As a result, it was demonstrated how crucial it is to build aerobic digesters based on the minimal holding duration of two digesters connected in series.

15. Centrifuge and sludge thickening:

In municipal wastewater treatment plants it is best to use typical automated handling of biosolids through sedimentation type centrifuges, as listed in the following table 11:

Table 11: Different types of centrifuges employed in municipal waste water treatment plants (Wang et al. 2024).

| Type of centrifuge | Particle size | Maximum centrifugal force Level x G | Feed concentration % |
|--------------------|------------------------------|--|----------------------|
| Scroll-solid bowl | $6400\mu \text{ to } > 1\mu$ | 3,000 | 1 – 50 |
| Disc-Nozzle | $100\mu \text{ to } > 1\mu$ | 4,000 | 2 – 10 |
| Disc-valve opening | $100\mu \text{ to } > 1\mu$ | 4,000 | 2 – 10 |

| Type of centrifuge | Capacity range m^3/hr (GMP) | Solids Discharge | Manner of solids discharge |
|--------------------|---|------------------|----------------------------|
| Scroll-solid bowl | $\frac{1}{4} - 250$ (1 – 1000) | Solid | Continuous |
| Disc-Nozzle | $\frac{1}{4} - 250$ (1 – 1000) | Fluid | Continuous |
| Disc-valve opening | $\frac{1}{4} - 180$ (1 – 800) | Fluid | Controlled Cycle |
| Type of centrifuge | Capacity range m^3/hr (GMP) | Solids Discharge | Manner of solids discharge |
| Scroll-solid bowl | $\frac{1}{4} - 250$ (1 – 1000) | Solid | Continuous |

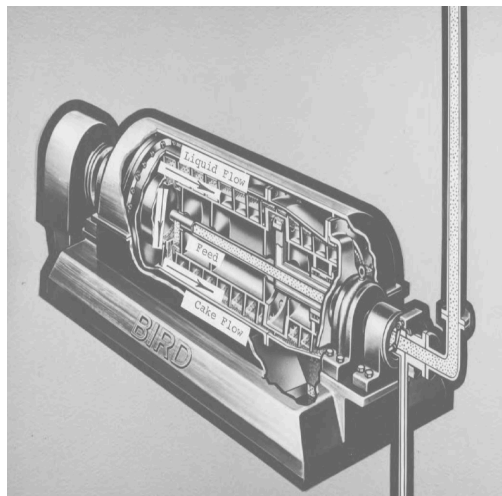


Figure 31: Centrifugal apparatus

The centrifugal apparatus is demonstrated in figure 31 by the following equipment, which pumps the solid waste intake into the centrifuge machine. The liquids collected at the outer chamber are fed back into the aerobic digester, leaving the dry solids suitable for transportation to a landfill. The centrifugal apparatus is made to rotate at a high speed in order to dewater the liquid from the particles.

Conclusion

This study examined the effects of primary treatment techniques (primary clarifier, A stage, or CEPT) on the digestibility and properties of sludge as well as the economics of the entire plant. The digestibility and properties of the sludge were significantly impacted by the treatment procedure and technology. Digestion of primary sludge produced the maximum quantity of methane, followed by A-sludge. Primary sludge included the highest amount of lipids, cellulose and hemicellulose, while A-sludge had the highest amounts of proteins and CEPT sludge had the highest amounts of lignin. According to plant-wide mass balances, the amount of organic matter in wastewater converted into methane gas was approximately 20, 27.4, and 33.4% with the implementation of primary clarifier, A-stage, and CEPT, respectively. CEPT sludge digestion yielded the lowest amount of methane, which was 30% lower than that of primary sludge.

Due to space, odour, and sludge limits, integrated anaerobic-aerobic bioreactors have gained popularity for combining aerobic and anaerobic processes in one unit.

Compact integrated bioreactors are expected to be able to treat a variety of high organic strength industrial and municipal wastewater due to its straightforward yet affordable technology, ability to generate renewable energy, and exceptional treatment efficiency. However, the majority

of the integrated bioreactors described in this paper have not been widely implemented in industry, and more research is necessary to assess these promising reactors' performance on a larger scale. Additionally, it is seen to be crucial to make other advancements like installing a biogas capture system and using packing medium or suspended carriers.

This research highlights the potential for significant cost reductions and improved efficiency in wastewater treatment through the integration of membrane technologies. The findings suggest that combining reverse osmosis, forward osmosis, and nanofiltration could lead to substantial improvements in both water recovery and contaminant removal from pharmaceutical wastewaters. However, further economic analysis is needed to determine the feasibility of implementing these technologies on an industrial scale.

Leachate levels vary based on landfill ages and site-specific factors. The ideal treatment should be simple and adaptive to the leachate parameters. However, the landfill leachate treatment technique relies on the following criteria:

- Initial leachate quality: treatment effectiveness based on metrics such as fill age, COD, BOD/COD, organic load, and ammonium compounds.
- Respect local discharge water limits and adapt to stricter pollution control criteria. Combining biological and physical-chemical treatment techniques can accomplish partial elimination of pollutants. This is due to "hard-COD," which makes it difficult to access the new regulations. In the past two decades, membrane filtration methods have become a proven solution for meeting water quality criteria. Increased landfill restrictions and management hindered traditional treatment efficiency. Alternatively, biological pre-treatment techniques, like as reverse osmosis [R.O.], are often ineffectual. With rejection rates of 98 to 99%, pressure-driven membrane RO outperforms biological treatment in efficiency and versatility. Lime dose of 5 g L⁻¹ can reduce salinity by 15-40% by de-carbonating leachates and removing hardness-related magnesium and calcium, as well as significant CaCO₃ precipitation. The pre-treatment additionally eliminates 20-30% COD and refractory organic macromolecules such humic acids through co-precipitation.

Lime precipitation is an effective pre-treatment method for removing colloidal particles and organic macromolecules from landfill leachate. Microfiltration and ultrafiltration have proven effective techniques. The leachate that has been pre-treated with lime has a number of advantages over decantation. These include the elimination of small, non-settleable particles, a reduction in the quantity of generated sludge, and a smaller facility size in comparison to the decanter. A study found that biological approaches may effectively remove heavy metals, COD, and NH₃-N in young leachate compared to conventional treatments. Physicochemical treatment can effectively remove organic refractory elements from less bio-degradable biologically treated or stabilised leachate during the refining process. An integrated biological-chemical-physical process can improve the efficacy of the entire treatment process, regardless of order. The review of leachate collection, control, and treatment highlights the relevance and complexity of this environmental issue. Recently, landfill leachate wastewater treatment issues may occur in many treatment plants globally.

Examination of landfill wastewater treatment systems is crucial to determine the most successful and cost-effective options based on the final effluent characteristics. The final decision and the selected treatment procedure should also be contingent upon the energy consumption, the required chemicals, and the maintenance. Operate with trained staff and enhanced technology to improve wastewater. In rare circumstances, landfill leachate wastewater treatment may encounter issues.

To pick the most successful and cost-effective landfill wastewater treatment system, it is crucial to analyse the properties of the final treated effluent.

The final decision and the selected treatment procedure should also be contingent upon the energy consumption, the required chemicals, and the maintenance.

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