OPTIMISATION OF SLUDGE LINE MANAGEMENT TO ENHANCE PHOSPHORUS RECOVERY IN WWTP

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Abstract

The management of the sludge treatment line can be optimized to reduce uncontrolled phosphorus precipitation in the anaerobic digester and to enhance phosphorus recovery in WWTP. In this paper, four operational strategies, which are based on the handling of the prefermented primary sludge and the secondary sludge from an EBPR process, have been tested in a pilot plant. The separated or mixed sludge thickening, the use of a stirred contact tank and the elutriation of the thickened sludge are the main strategies studied. Both the reduction of phosphorus precipitation in the digester as well as the supernatant suitability for a struvite crystallization process were assessed in each configuration. The mixed sludge thickening combined with a high flowrate elutriation stream reduced the phosphorus precipitation in the digester by 46%, with respect to the separate sludge thickening configuration (common practice in WWTP). Moreover, in this configuration, 68% of the soluble phosphorus in the system is available for a possible phosphorus recovery process by crystallization (not studied in this work). However, a high Ca/P molar ratio was detected in the resultant supernatant which is pointed out as a problem for the efficiency of struvite crystallization.

Keywords

Anaerobic digestion, phosphorus, precipitation, sludge treatment, struvite, wastewater

Nomenclature

5 ACP Amorphous calcium phosphate

ALK_P Bicarbonate alkalinity

ALK_T Total alkalinity

HAP Hydroxyapatite

m Sludge mass flowrate

10 MAP Struvite

NH₄-N Ammonia nitrogen

OLR Organic Loading Rate

PAO Polyphosphate Accumulating Organisms

MP Mixing power

15 PO₄-P Orthophosphate

PP Potential Phosphorus

PPS Prefermented Primary Sludge

SGP Specific gas production

SRT Solid retention time

20 TCOD Total chemical oxygen demand

TP Total phosphorus

TS Total solids

TVS Total volatile solids

VFA Volatile fatty acids

25 w_P Mass of phosphorus per mass of treated sludge

Q Volumetric flowrate

%P-MAP Percentage of phosphorus fixed as MAP

%P-HAP Percentage of phosphorus fixed as HAP

%P-ads Percentage of phosphorus adsorbed

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Subscripts

av available

elut elutriation

10 fix fixed

lost lost

ORGrel release from organic matter degradation

PAOrel release from Poly-P hydrolysis

prec precipitated

15 rem removal

TOTrel total release

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1. Introduction

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Eutrophication of water bodies caused by an increased input of inorganic phosphorus is a major worldwide problem (de-Bashan and Bashan, 2004). Among the different methods developed, the enhanced biological phosphorus removal process (EBPR) appears as a feasible technology to remove phosphorus in wastewater treatment plants (WWTP). In contrast to conventional plants, the phosphorus content of activated sludge in EBPR processes reaches values of up to 7% (Jardin and Pöpel, 1994) due to polyphosphate (Poly-P) storage by polyphosphate accumulating organisms (PAO).

- On the other hand, phosphorus recovery and reuse has become important in recent years since phosphorus is a limited resource. After human consumption, most of the phosphorus remains in wastewaters. For this reason, phosphate recovery from wastewaters is considered to be a suitable method to promote phosphorus recycling.
- In the sludge treatment line, especially in the anaerobic digestion process, phosphorus from Poly-P hydrolysis and from organic matter degradation is released, increasing the orthophosphate content in the system and, therefore, the potential for phosphorus precipitation. The concentration of other ions such as ammonium, potassium and magnesium also increases in the digester. Furthermore, the hardness of influent wastewater also controls precipitation processes of phosphorus inside the digester since it determines Ca²⁺ and Mg²⁺ concentrations in the wastewater.

According to the literature (Wild et al., 1997), different phosphate compounds are likely to precipitate in the sludge treatment system causing significant operational problems that lead to an increase in maintenance costs. Accumulation of struvite

(MgNH₄PO₄.6H₂O) on pipe walls and equipment surfaces of anaerobic digestion and post-digestion processes has been reported as a frequent problem in the wastewater treatment industry (Ohlinger et. al., 1998; Parsons and Doyle, 2004). Struvite precipitation occurs when the combined concentrations of Mg²⁺, NH₄-N and PO₄-P exceed the struvite solubility product. Since specification of these components is a pH function, struvite solubility also varies with pH. As pH increases, struvite solubility decreases. Therefore, anaerobic digestion and post-digestion processes, which show a higher pH value than the previous stages of the wastewater treatment, are more susceptible to struvite formation.

Phosphorus precipitation in the sludge line also determines the dissolved phosphorus concentration in the return liquor stream, which causes variations that might affect the efficiency of the EBPR process. Therefore, in order to assure a successful nutrient removal process, recirculated liquors to the wastewater treatment should be treated.

Phosphorus recovery by struvite crystallization is one of the technologies developed for treating sludge digester liquors. The product obtained, struvite, is a valuable slow-release fertiliser for agriculture and, hence, an economical benefit can be obtained. In order to guarantee P rich-streams and thereby to obtain a high struvite production in the crystallization process, the uncontrolled phosphorus precipitation in the digester should be reduced. Other factors, such as the calcium concentration in the crystallizer influent stream, can significantly affect the production of struvite (Pastor et al., 2007). Therefore, suitable characteristics of the supernatants treated by crystallization should be guaranteed.

A simple methodology to assess phosphorus precipitation in anaerobic digesters based on routine experimental analysis and mass balances was developed in a previous work (Martí et al., 2007). This methodology was successfully applied in an anaerobic digestion pilot plant treating prefermented primary sludge and secondary sludge from an EBPR process and in a full-scale WWTP in Murcia, Spain (Barat et al., 2008).

In this work, the above methodology has been applied to optimise sludge treatment line management in order to minimise uncontrolled phosphorus precipitation in the digester. Hence, operational costs associated to the formation of deposits can be reduced and the phosphorus recovery can be enhanced. Four operational strategies, which are based on the management of the prefermented primary sludge and the secondary sludge from an EBPR process, were tested in a pilot plant.

2. Materials and methods

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15 **2.1.** Anaerobic digestion pilot plant

The anaerobic digestion pilot plant treats the prefermented primary sludge (PPS) generated in a fermentation/elutriation pilot plant and the waste activated sludge from an EBPR pilot-scale process (EBPR sludge). The three plants are located in the Carraixet WWTP, which is near the city of Valencia (Spain). A detailed description of the anaerobic digestion pilot plant can be found in Martí et al., (2007).

2.2. Process schemes

The four operational configurations studied are based on the separated or mixed sludge thickening, the use of a contact tank, the elutriation of the thickened sludge and the flow rate of the elutriation stream. Table 1 summarizes the main characteristics of each

configuration. Figure 1 outlines the four operational configurations. The supernatants and centrates obtained in the different configurations described below were used in a phosphorus recovery process by struvite crystallization, which is not the aim of this work. The study performed in each configuration only considers the sludge treatment management and no crystallization results are presented. For that reason, this latter process is drawn with grey lines in Figure 1.

2.2.1. Configuration C1: Separated thickening of sludges

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In this configuration (Figure 1a) the PPS (1) is directly pumped from the primary settler/fermenter to the digester. No thickening of this prefermented sludge is carried out due to the high solid concentrations achieved in the primary settling tank. EBPR sludge (2) is wasted from the aerobic reactor of the EBPR system and pumped to the gravity thickener, where is settled and thickened. Once both sludges are mixed in the influent pipe (4), they are fed to the anaerobic digester. The digested sludge (5) is dewatered by centrifugation and the centrate obtained is sent to the phosphorus recovery process by struvite crystallization.

2.2.2. Configuration C2: Mixed thickening and sludge elutriation

In this configuration (Figure 1b) PPS (1) and EBPR sludge (2) are pumped from the primary settler/fermenter and the secondary settler respectively to the gravity thickener. PPS is thickened to provide a suitable contact and mixing with EBPR sludge. Elutriation of the soluble phosphorus is carried out by recirculation of the thickened sludge (4) from the thickener bottom to the inlet sludge zone. The thickener supernatant (3) is mixed with the centrate obtained in the digested sludge dewatering and directed to the phosphorus recovery process by struvite crystallization.

2.2.3. Configuration C3: Contact tank previous the mixed thickening of sludges

In order to increase the mix and contact time between the two sludges a stirred contact tank previous to the mixed thickening step is placed in the pilot plant (Figure 1c). PPS (1) and EBPR sludge (2) are pumped to the contact tank. The retention time in the tank was set at 3.6 hours. Previous experiments showed that this retention time (3.6 h) was enough to enhance phosphorus released from Poly-P in the tank. The contact tank outlet stream is sent to the gravity thickener, where no elutriation of the thickened sludge takes place, and is afterwards digested as in the previous configurations. Both the thickener supernatant (3) and the centrate of the digested sludge dewatering are sent to the phosphorus recovery process.

2.2.4. Configuration C4: Mixed thickening with sludge high elutriation flowrate

In this configuration (Figure 1d), PPS from the primary settler/fermenter (1) and EBPR sludge from the aerobic reactor of the BNR system (2) are pumped to the thickener. Elutriation of the thickened sludge is also carried out, as previously described. Moreover, the EBPR sludge from the aerobic reactor is used as a "washing-stream" that enhances the thickened sludge elutriation. The thickened sludge (4) is digested and dewatered by centrifugation. The centrate obtained in the dewatering system and the thickener supernatant (3) are mixed and directed to the crystallization process.

2.3. Experimental procedure

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The anaerobic digestion process was operated at mesophilic conditions (T=35±1°C) and at a solids retention time (SRT) of 20 days in all experiments. Table 2 provides a summary of the anaerobic digestion process for the four experiments carried out,

including the operational parameters in the digester, the influent and effluent sludge characterization, and the anaerobic digestion results. The organic loading rate (OLR), the mass flowrate of sludge fed to the digester (m), and the percentage of primary and secondary sludge in the total sludge fed to the digester (%PPS and %EBPR, respectively) depend on the operation of the fermentation/elutriation and the EBPR processes. The elutriation flowrate (Q_{elut}) was maintained at 12 1 d⁻¹ and 15 1 d⁻¹ for configurations C2 and C4, respectively.

2.4. Analytical methods

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Total solids (TS), total volatile solids (TVS), total COD (TCOD), total phosphorus (TP), and total alkalinity (ALK_T) were performed in accordance with the Standard Methods (APHA, 2005). Volatile fatty acids concentration (VFA) and bicarbonate alkalinity (ALK_P) were measured using the 5-point titration method (Moosbrugger et al., 1992). PO₄-P, NH₄-N, and soluble calcium, magnesium and potassium were analysed by ion chromatography (Metrohm IC, Switzerland).

Biogas composition (CO₂, N₂, and CH₄) was analyzed by gas chromatography separation (6890N Agilent Technologies, USA) with 2 capillary columns (HP- PLOT Molesieve and HP- PLOT Q) and thermal conductivity detector (TCD). Injector and detector temperatures were fixed at 110°C and 25 °C, respectively. Oven temperature was maintained at 40°C for 12.5 minutes. The carrier gas was He and the flow rate used was constant at 4 ml min⁻¹.

2.5. Calculation of phosphorus precipitation and potential phosphorus for crystallization

PO₄-P, Ca²⁺, Mg²⁺, and K⁺ mass balances were carried out to assess phosphorus precipitation in the digester using the methodology developed in Martí et al. (2007). As explained in that work, PO₄-P, K⁺ and Mg²⁺ are released during sludge anaerobic digestion due to polyphosphate hydrolysis and organic matter degradation, whereas Ca²⁺ is considered to be released only due to organic matter degradation. The organic content of phosphorus, potassium, magnesium and calcium in the mixed influent sludge was calculated using the proportion of PPS (%PPS) and EBPR sludge (%EBPR) in each experiment (Table 2) and the content of each element in the organic matter (Martí et al., 2007).

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In order to confirm the precipitates formed in the digester conditions, the Saturation Index values (SI) for the precipitates considered were calculated. The SI is used to describe the saturation state of the aqueous phase composition versus different solids (Eq. 1). When SI = 0, the solution is in equilibrium; when SI < 0, the solution is undersaturated and no precipitation occurs; when SI > 0, the solution is supersaturated and precipitation occurs spontaneously. Therefore, the SI values can be used to evaluate the effect of the solution composition on the tendency and extent of the precipitation, according to the following equation:

$$SI = log \frac{IAP}{K_{SP}}$$
 (Eq. 1)

where IAP represents the ion activity product, and K_{SP} represents the thermodynamic solubility product.

The potential phosphorus (PP), which represents the phosphorus that can be recovered in the crystallization process with regard to the maximum phosphorus available in the sludge treatment system, was also calculated. The PP value in each configuration was evaluated as follows:

$$PP(\%) = \frac{w_{PAV}}{w_{PAV} + w_{PLOST}} \cdot 100 = \frac{w_{PAV}}{w_{PTOT}} \cdot 100$$
 (Eq. 2)

where:

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w_{P AV}: g of phosphorus available per kg of treated sludge

WPLOST: g of phosphorus lost per kg of treated sludge

w_{P TOT}: g of phosphorus available in the system per kg of treated sludge

10 **3. Results**

3.1. Anaerobic digestion

The digester influent and effluent sludge characteristics are shown in Table 2. The data included in these tables represent the average value of each parameter once the stationary conditions (a minimum of two times the SRT) were reached. The standard deviation (s.d.) is also included. As Table 2 shows, differences in the influent sludge characteristics can be observed among the four configurations tested. As stated above, the anaerobic digestion pilot plant treats the PPS and the EBPR sludge from the two other pilot plants located in the Carraixet WWTP. Therefore, the digester influent sludge is widely influenced by the operation of the previous wastewater treatment processes. Moreover, the variability of the influent wastewater also affects operation of the pilot plants.

The results obtained in the anaerobic digestion process confirm that the four configurations tested show a TVS removal (TVS $_{\text{rem}}$) between 40% and 60%. Process

stability in the digester was assessed by means of pH, VFA and alkalinity measurements in the reactor. The pH evolution was continuously monitored showing values around 7.0±0.2 in the four experiments. With regard to the VFA concentration in the digester, a wide variability for the different configurations was obtained, reaching high values in configurations C3 and C4. The ratio between the alkalinity associated with the VFA (calculated as the difference between total alkalinity and bicarbonate alkalinity) and the total alkalinity, ALK_{VFA}/ALK_T, was lower than 0.25 in all the experiments.

3.2. Precipitation assessment in the anaerobic digester

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The results obtained from the mass balances application in each configuration studied are shown in Table 3. The highest concentration of fixed phosphorus ($P_{\rm fix}$) was obtained in configuration C1 due to the great concentration of available phosphorus ($P_{\rm av}$) that was achieved in this configuration. As Table 3 shows, in this configuration, 403.4 mg l⁻¹ of phosphorus precipitated in the anaerobic digester. The mixed thickening of both sludges and the elutriation of the resulting thickened sludge (configuration C2) made it possible to reduce the phosphorus release in the digester, which led to a decrease in the fixed phosphorus. The introduction of a stirred contact tank (configuration C3) slightly reduced the phosphorus fixation. Finally, the configuration C4 made it possible to reduce the phosphorus fixation in the digester by 43% compared with the separated thickening configuration.

The magnesium release in the digester is due to both the Poly-P hydrolysis and the organic matter degradation. As observed with phosphorus, the magnesium released associated to Poly-P hydrolysis (Mg_{PAOrel}) was significantly reduced in configuration

C4 giving rise to the lowest precipitation of magnesium in the digester. With regard to calcium precipitation, the highest value was achieved in configuration C1.

3.3. Precipitates formed in the anaerobic digester

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The most common salts found in anaerobic digesters are phosphates and carbonates of calcium and/or magnesium. In this work, magnesium has been considered to precipitate in the digester mainly as struvite, making it possible to estimate the amount of phosphate precipitated as struvite (%P-MAP) from its stoichiometry. The calcium that precipitated in the digester was considered to be in the form of hydroxyapatite (HAP).

Thus, the amount of phosphate precipitated with calcium (%P-HAP) was calculated using a molar ratio Ca/P of 1.67. The precipitation of aluminium and iron phosphates was not considered in this work due to the negligible soluble iron and aluminium concentrations in the digester influent and effluent streams. Therefore, the rest of the phosphorus fixed in the digester was considered to be adsorbed on the surface of solids (%P-ads).

According to these considerations, the phosphorus fixed in the digester as struvite (%P-MAP), as hydroxyapatite (%P-HAP) or adsorbed on solid surfaces (%P-ads) has been estimated for the four configurations. The results obtained and the molar ratios Mg/P, Mg/Ca and Ca/P in the digester are shown in Table 4. As this table shows, the highest percentage of phosphorus fixed as MAP is obtained in configuration C2, in agreement with the highest Mg/Ca molar ratio and the lowest Ca/P molar ratio. The highest percentage of phosphorus fixed as HAP is achieved in C4 in agreement with the highest Ca/P molar ratio. In all the experiments, the main phosphorus precipitate that formed is struvite.

The Saturation Index values (SI) have been calculated for the precipitates considered and they are shown in Table 5. For the four configurations, the digester sludge composition is supersaturated with regard to MAP, HAP and its precursor ACP, which confirms the possible formation of these precipitates in the digester. With regard to potassium struvite formation, the results obtained show that the solution is undersaturated at the conditions achieved in the digester, which proves that no potassium struvite precipitates. Moreover, the presence of MAP in solid deposits found in the outgoing digester pipe was confirmed by means of XRD analysis (Figure 2).

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3.4. Potential phosphorus for the struvite crystallization process.

In order to quantify the phosphorus that could be recovered in a crystallization process, the loss of PO₄-P in any stage of the sludge treatment system and the available PO₄-P in the supernatant and centrate produced were both calculated for each configuration. In order to compare the results among the different configurations, the results were calculated considering the mass flowrate of sludge fed to the digester.

Two P-rich streams can be treated by struvite crystallization: the centrate from the digested sludge dewatering and the thickener supernatant. Table 6 shows the average values of the volumetric flowrate (Q) and the PO₄-P concentration for both streams. The volumetric flowrate in the digester effluent is reduced by a factor of 4/6 in the dewatering system, whereas the PO₄-P concentration does not undergo significant variations. With regard to the thickener supernatant, important differences in both Q and PO₄-P values can be observed. These differences can be attributed to the operational characteristics of each configuration.

In the sludge treatment line, phosphorus losses may occur in the anaerobic digester due to precipitation as different phosphate salts, and always occurs in the dewatering process where part of the soluble phosphorus remains in the dewatered sludge.

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The maximum phosphorus that could be recovered in a crystallization process with regard to the maximum phosphorus available in the sludge treatment system, (i.e. potential phosphorus, PP) is shown in Table 7. Configurations C1 and C3 show low PP values of around 20%. Configuration C2 increases the PP value up to 37%, and configuration C4 shows the best results, reaching a PP value of 68%.

4. Discussion

4.1. Operation of the anaerobic digester

As Table 2 shows, the VFA concentration in the digester shows a wide variability for the different configurations tested, which can be related to the differences in the VFA concentration in the influent sludge. VFA in the digester inlet stream comes mainly from the PPS. Therefore, the operation of the primary sludge fermentation process determines the VFA concentration in the sludge that is fed to the digester. In this work, the ratio ALK_{VFA}/ALK_T was below the maximum recommended values (i.e., 0.3-0.4) to ensure operational stability of the process (Lema et al., 1993).

With regard to the performance of the anaerobic digestion process, the four configurations tested show TVS_{rem} values between 40% and 60%, in agreement with the typical ones observed in anaerobic mesophilic digesters that treat mixed sludges (Grady et al., 1999).

The specific gas production (SGP) measured in all the studied configurations is in agreement with the literature data observed when digesting primary and secondary sludges (Zhao and Viraraghavan, 2004). Biogas composition shows values between 63%-65% of methane content in the four configurations, which is a typical value found for proper anaerobic digestion processes.

4.2. Precipitation in the anaerobic digester

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The precipitation study in the anaerobic digester (Table 3) points out that the highest P_{fix} was achieved for configuration C1, which can be explained by the highest P_{av} in this configuration. The conventional configuration of a sludge treatment line (i.e., separate thickening of primary and EBPR sludge) does not enhance Poly-P hydrolysis in the thickener due to the lack of high VFA concentration, which is provided principally by primary sludge. Therefore, Poly-P hydrolysis takes place mainly in the digester, which increases phosphate concentration and as a consequence, increases the potential for phosphorus precipitation inside the digester. The reduction of Pfix in configuration C2, with respect to configuration C1, has been attributed to a decrease in the phosphorus released in the digester. Both the presence of significant VFA concentrations from primary sludge in the thickener and the anaerobic conditions achieved in it enhanced Poly-P hydrolysis. Thickened sludge recirculation favoured the elutriation of the released phosphorus and thus the phosphorus withdrawal from the digestion system. In configuration C3, the use of a stirred contact tank did not significantly reduce the Poly-P hydrolysis in the digester compared with configuration C2. Configuration C4 shows the lowest phosphorus precipitation in the digester. The mixed thickening of PPS and EBPR sludge combined with an elutriation system using

an external stream of high flowrate reduced the available phosphorus in the digester. As Table 3 shows, only 11.1 mg l⁻¹ of phosphorus was released in the digester from Poly-P hydrolysis, with the phosphorus release in the digester being due mainly to organic matter degradation.

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Unlike magnesium ions, calcium ions do not play any role in the stabilization of the charge of Poly-P. Therefore, no release of calcium associated to Poly-P hydrolysis during anaerobic digestion is expected since organic matter degradation is the only mechanism for calcium release. The calcium available to precipitate in the digester is mainly the soluble calcium concentration in the digester influent. Hence, both the hardness of the influent wastewater and the sludge handling prior to digestion determine the precipitation of calcium in the digester. Table 3 also shows that the highest calcium precipitation was achieved in configuration C1, which can be attributed to the highest soluble calcium concentration in the digester (influent + release from organic matter degradation).

4.3. Evaluation of the precipitates formed

According to the literature, four possible phosphate magnesium salts can precipitate from a solution containing Mg²⁺, NH₄-N, and PO₄-P (i.e., struvite (MgNH₄PO₄.6H₂O-MAP); magnesium hydrogen phosphate trihydrate or newberyite (MgHPO₄·3H₂O); and trimagnesium phosphate in two states of hydration (Mg₃(PO₄)₂·22H₂O and Mg₃(PO₄)₂·8H₂O)). According to Martí et al. (2007), of these, struvite was considered the main precipitate formed in the digester at the experimental conditions existing in the four experiments. The XRD analysis (Figure 2) in the solids deposits found in the effluent digester pipe confirmed this point. As other studies show (Heinzmann and

Engel, 2005), struvite formation is favoured in the anaerobic digestion and post-digestion processes, especially in high turbulence zones such as pipes, valves, etc., where the spontaneous release of carbon dioxide leads to a pH increase. Moreover, precipitation of magnesite (MgCO₃) has not been considered due to its high solubility when compared to struvite. Different authors (Schuiling and Andrade, 1999; Wilsenach et al., 2006) have observed that potassium struvite (KMgPO₄·6H₂O) could precipitate instead of MAP only in the case of low ammonium concentrations, which it is not the case in any of the experiments carried out. Therefore, magnesium can be considered to precipitate in the digester mainly as struvite, making it possible to estimate the amount of phosphate precipitated as struvite (%P-MAP) from its stoichiometry.

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With regard to calcium precipitates, according to Martí et al. (2007), hydroxyapatite (HAP) was considered to be the calcium phosphate formed due to the high process retention time, which allows precursors (i.e., mainly amorphous calcium phosphate - ACP) to transform into HAP. Calcite formation was not considered due to the high phosphate concentrations in the digester (Plant and House, 2002; Lin and Singer, 2005). The rest of the phosphorus fixed neither as MAP nor as HAP was considered to be adsorbed on the surface of solids (%P-ads). Although phosphate adsorption during anaerobic digester is still not a well-understood process, it is considered to be a possible mechanism for phosphorus fixation (Jardin and Pöpel, 1994; Wild et al., 1997).

4.4. Evaluation of potential phosphorus for struvite crystallization

The potential phosphorus (PP) defined in this work, indicates the efficiency of each configuration to guarantee available PO₄-P in a struvite crystallization process. The low PP value in configuration C1 (16%), is a result of the high phosphorus precipitation in

the digester and the low PO₄-P mass flowrate value in the thickener effluent (Table 7), which made this stream not suitable for a crystallization process. Configuration C3 also shows a low PP value (18%). Although phosphorus precipitation in the digester is slightly reduced in C3, the available phosphorus is very low principally in the thickener supernatant. This result indicates that the use of a contact tank instead of the thickened sludge recirculation does not enhance the phosphorus extraction from the system, due to a possible phosphorus precipitation in the contact tank. Configuration C2 shows a high PP value with respect to the previous configurations due to a higher availability of phosphorus (w_{P AV}) in the thickener supernatant as well as in the dewatering system centrate. Nevertheless, the best results were obtained in configuration C4, with a PP value of 68%, which is a consequence of the low phosphorus precipitation in the digester and the high phosphorus content in the thickener supernatant. Therefore, this latter configuration obtains the highest phosphorus availability for the crystallization system.

However, as Table 8 shows, the crystallizer inlet stream in configuration C4 has the highest Ca/P molar ratio. This value is due to the high thickener supernatant flowrate produced as a consequence of using the EBPR sludge (i.e., not settled) as a "washing-stream". In fact, if the dewatering centrate and the thickener supernatant volumetric flowrates are compared (i.e., 2.8 1 d⁻¹ vs 43.8 1 d⁻¹, respectively (Table 6)), it can be observed that the stream that would be feed to the crystallizer consists mainly of the thickener supernatant. Therefore, most of the calcium content in the wastewater is transferred to the crystallizer inlet stream, which, in this case, leads to a high molar ratio Ca/P.

High calcium content in the crystallizer influent stream has pointed out as a detrimental influence on struvite formation during crystallization processes. Pastor et al. (2007) studied this effect in real supernatants showing a precipitation of calcium phosphates rather than struvite in the crystallizer. Thus, although configuration C4 decreases the precipitation in the digester due to the high elutriation provided with the "washing stream", the characteristics of the crystallizer inlet stream in this configuration could affect the struvite precipitation efficiency. For this reason, if the objective is to obtain struvite in the crystallization process, this strategy would not be the best one, especially if the water hardness is high, due to the high calcium content in crystallizer influent stream.

Future investigations will focus in the phosphorus adsorption during anaerobic digestion and in the fate of phosphorus in the global wastewater treatment system.

15 Conclusions

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In this paper, different operational strategies, which are based on the handling of the sludges produced in wastewater treatment, have been tested. The main conclusions are shown below:

- In the four configurations, the anaerobic digestion of PPS and EBPR sludge showed TVS_{rem} values between 40% and 60%, ALK_{VFA}/ALK_T ratios below 0.25 and typical values for biogas production and composition.
- MAP was found as the main precipitate that formed in the digester. The Saturation Index (SI) values confirm the possible formation of MAP, HAP and ACP and prove that potassium struvite does not precipitate at the studied conditions. Phosphorus adsorption on solid surfaces was also observed.

Phosphorus precipitation in the digester can be reduced using mixed thickening
of PPS and EBPR sludge, and a high flowrate "washing-stream" in the
thickener. With this operational strategy 68% of the soluble phosphorus in the
system would be available for a crystallization process.

The characteristics of the "washing stream" (flowrate and composition) will determine the thickener supernantant suitability for phosphorus recovery as struvite. For this reason, special attention must be paid to the Ca/P molar ratio in waters with high levels of hardness.

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 ${\bf Table~1~-Operational~characteristics~of~the~configurations~tested}$

Configuration	Sludge thickening	Elutriation	Secondary sludge settling	Contact tank
C1	separated	no	no	no
C2	mixed	yes	yes	no
C3	mixed	no	yes	yes
C4	mixed	yes	no	no

Table 2-Operational parameters and digestion process results

Operational pa	arameters			C 1				C 2				C 3		C4			
Q	(1 d ⁻¹)		6.	.75			6.	.75	· 		8				4.	.25	
MP	$(W l^{-1})$		0.1	.11	J	1	0.11				0.	.09	l	0.36			
OLR	(kg COD 1 ⁻¹ d ⁻¹)			.03	J	1		.44	,			.4	l	1.8			
m	(kg SST d ⁻¹)		0.2	251	J	1		293	,		0.7	283	I	0.166			
%PPS	(%)			51	J	1		33	,			55	I	45			
%EBPR	(%)		4	49			6	7	!		3	35		55			
		Influ	uent	Efflu	aent	Infl	luent	Efflu	aent	Infl	uent	Effl	luent	Infl	luent	Effl	uent
Sludge charact	terization	average	s.d.	average	s.d.	average	s.d.	average	s.d.	average	s.d.	average	s.d.	average	s.d.	average	s.d.
TS	(mg l ⁻¹)	37257	± 8444	18554	± 3224	26671	± 3516	17619	± 2616	26085	± 5887	16736	± 659	29583	± 3138	16850	± 3905
TVS	$(mg l^{-1})$	23194	± 4403	10484	± 1683	18734	± 2517	10848	± 1632	18206	± 4164	9764	± 352	18637	± 1859	7545	± 1835
TVS	(%)	64	± 7	56	± 3	69	± 1	62	± 1	70	± 6	58	± 1	63	± 4	45	± 1
TCOD	$(mgO_2 l^{-1})$	40650	± 8129	16660	± 2623	28875	± 4221	15975	± 1995	28514	± 4372	13482	± 722	32650	± 1314	11076	± 1166
TP	(mg l^{-1})	1020	± 102	985	± 22	981	± 126	902	± 145	730	± 147.1	687	± 100.4	577	± 20.4	443	± 46.9
PO ₄ -P	(mg l ⁻¹)	70.7	± 19.8	152.2	± 25.8	253.7	± 67.4	201.4	± 28.1	109.1	±25.1	69.1	± 11.05	98.6	± 24.1	67.6	± 14.3
NH ₄ -N	$(mg l^{-1})$	104.1	± 51.8	617.4	± 60.6	69.0	± 35.0	580.0	± 81.1	126.3	± 46.9	651.4	± 26.2	110.6	± 18.9	796.0	± 68.2
VFA	(mgHAc l ⁻¹)	762	± 458	120	± 56	301	± 155	22	± 10	520	± 304	1013	± 547	592	± 220	808	± 341
pН		6.4	± 0.1	7.0	± 0.1	6.1	± 0.1	7.0	± 0.2	6.2	± 0.1	6.9	± 0.1	6.2	± 0.1	7.2	± 0.1
ALK_P	$(mgCaCO_{3.}l^{-1})$	407	± 106	2381	± 371	414	± 105	2308	± 307	523	± 127	2069	± 482	397	± 105	2245	± 330
ALK _{VFA} /ALK	_T			0.10	± 0.02			0.11	± 0.02			0.24	± 0.03			0.20	± 0.02
\mathbf{K}^{+}	(mg l ⁻¹)	71.8	± 23.6	237.4	± 13.2	160.2	± 39.9	258.6	± 30.3	100.4	± 19.1	179.8	± 59.0	65.9	± 20.3	132.6	± 6.2
Mg^{2+}	(mg l^{-1})	60.2	± 14.8	11.9	± 5.8	113.5	± 29.9	26.1	± 9.2	72.8	± 6.1	62.8	± 5.2	72.3	± 9.1	40.4	± 9.5
Ca ²⁺	(mg l ⁻¹)	151.3	± 20.9	49.8	± 18.6	104.9	±19.9	40.8	± 7.6	123.1	± 12.3	70.5	± 10.8	124.7	± 23.6	57.0	± 10.7
Anaerobic dige	estion results					-1								T			!
$TCOD_{rem}$	(%)		59	9.0	J	1	44	4.7	,		52	2.2	ļ		66	6.1	
TVS_{rem}	(%)		52	4.8		1	42	2.1		1	46	6.4		<u></u>	59	9.5	

(%) s.d.: standard deviation

Table $\bf 3$ - Precipitation results in the anaerobic digester

Parameter (mg l ⁻¹)	Configuration						
_	C1	C2	C3	C4			
K_{TOTrel}	165.6	98.5	79.4	66.7			
K_{ORGrel}	67.8	49.7	39.0	62.8			
K _{PAOrel}	97.8	48.7	40.4	3.9			
P _{PAOrel}	276.3	137.7	114.1	11.1			
PorGrel	208.6	139.7	130.5	186.9			
P_{TOTrel}	484.9	277.4	244.5	198.0			
P_{av}	555.6	531.1	353.7	296.6			
P_{fix}	403.4	329.7	284.6	229.0			
Mg_{PAOrel}	78.2	39.0	32.3	3.1			
Mg_{ORGrel}	57.4	34.1	39.3	49.4			
Mg_{TOTrel}	135.5	73.1	71.5	52.5			
Mg_{prec}	183.8	160.5	81.5	68.4			
Caliborg	24.8	21.3	11.8	24.4			
Caprec	126.3	85.4	64.4	92.1			

 $Table\ 4-Fixed\ phosphorus\ distribution\ and\ ion\ molar\ ratios\ in\ the\ digester$

	C1	C2	C3	C4
%P-MAP	58.1	62.0	36.5	38.1
%P-HAP	14.5	12.0	10.5	18.6
%P-ads	27.4	25.9	53.0	43.3
Mg/P	0.45	0.45	0.52	0.54
Mg/Ca	1.83	2.44	1.76	1.38
Ca/P	0.24	0.18	0.29	0.39

Table 5 - Saturation Index (SI) values at the digester conditions ${\bf r}$

Mineral	pK _{ps}	Reference	SI			
			C1	C2	C3	C4
Struvite (MAP)	13.3	Ohlinger et al. (1998)	1.462	1.288	1.142	0.968
Hydroxyapatite (HAP)	48.6	Murray and May (1996)	12.497	12.589	11.199	11.253
Amorphous calcium phosphate (ACP)	24.0	Mamais et al. (1994)	4.024	4.011	3.123	3.331
Potassium struvite	10.6	Taylor et al. (1963)	-2.389	-2.746	-2.474	-2.851

Table 6 - Characteristics of the P-rich streams produced

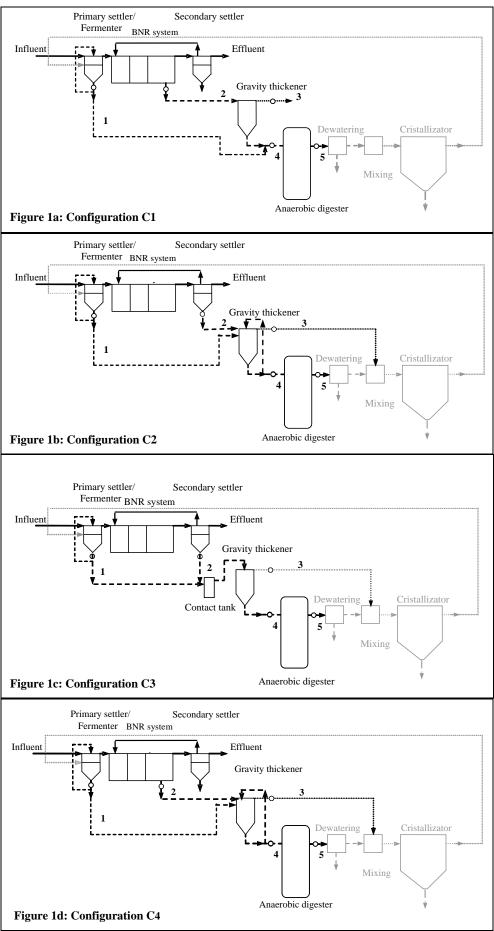
		Thickener su	pernatant		Dewatering centrate			
_	C1	C2	C3	C4	C1	C2	C3	C4
$Q (1 d^{-1})$	41.3 ± 3	5.3 ± 0.4	2.0 ± 0.2	43.8 ± 4	4.5 ± 0.2	4.5 ± 0.2	5.3 ± 0.3	2.8 ± 0.1
PO ₄ -P (mg l ⁻¹)	11.3 ± 2.3	127.0 ± 23.3	79.8 ± 22.3	46.5 ± 11.6	152.2 ± 18.2	201.4 ± 23.4	69.1 ± 7.6	67.6 ± 8.3

 Table 7 - Phosphorus recovery assessment

	C1	C2	C3	C4
WP LOST (gP kg sludge ⁻¹)				
Thickener supernatant	1.9	0	0	0
Precipitation in digester	10.8	7.6	8	5.8
Dewatered sludge	1.4	1.5	0.7	0.6
_Total	14.1	9.1	8.7	6.4
WP AV (gP kg sludge ⁻¹)				
Thickener supernatant	0	2.3	0.6	12.3
Dewatering centrate	2.7	3.1	1.3	1.2
Total	2.7	5.4	1.9	13.5
WP TOT (gP kg sludge ⁻¹)	16.8	14.5	10.6	19.9
Potential Phosphorus (%)	16	37	18	68

 ${\bf Table~8~-~Crystallizer~inlet~stream~molar~ratios}$

	C1	C2	C3	C4
Mg/Ca	0.39	1.67	1.35	0.61
Mg/P	0.10	0.43	1.12	1.50
Ca/P	0.25	0.25	0.83	2.45



Grey lines: processes not studied in this work

• Sampling points

Figure 1 - Layout of the four pilot plant configurations

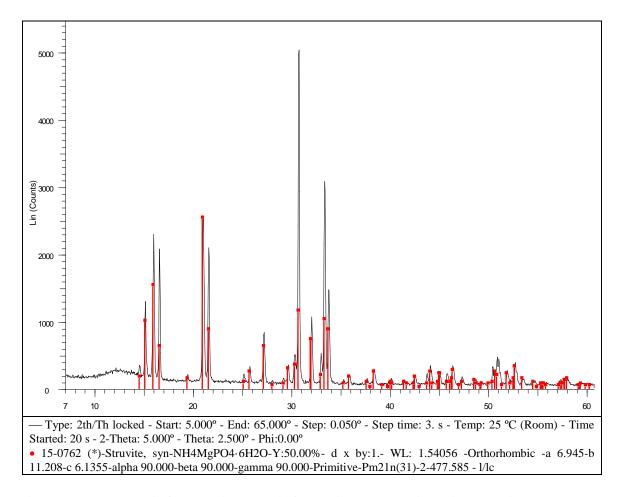


Fig. 2 - XRD analysis for a solid deposit found in the outgoing digester pipe.