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# 1 Harvesting microalgae using activated sludge can decrease polymer dosing and 2 enhance methane production via co-digestion in a bacterial-microalgal process

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## 7 **Abstract**

8 Third generation biofuels, e.g. biofuels production from algal biomass, have gained attention due to  
9 increased interest on global renewable energy. However, crop-based biofuels compete with food  
10 production and should be avoided. Microalgal cultivation for biofuel production offers an  
11 alternative to crops and can become economically viable when combined with the use of used water  
12 resources. Besides nutrients and water, harvesting microalgal biomass represents one of the major  
13 costs related to biofuel production and thus efficient and cheap solutions are needed. In bacterial-  
14 algal systems, there is the potential to produce energy by co-digesting the two biomasses. We  
15 present an innovative approach to recover microalgal biomass via a two-step flocculation using  
16 bacterial biomass after the destabilization of microalgae with conventional cationic polymer. A  
17 short solids retention time (SRT) enhanced biological phosphorus removal (EBPR) system was  
18 combined with microalgal cultivation. Two different bacterial biomass removal strategies were  
19 assessed whereby bacterial biomass was collected from the solid-liquid separation after the  
20 anaerobic phase and after the aerobic phase. Microalgal recovery was tested by jar tests where three  
21 different chemical coagulants in coagulation-flocculation tests ( $\text{AlCl}_3$ , PDADMAC and Greenfloc  
22 120) were assessed. Furthermore, jar tests were conducted to assess the microalgal biomass  
23 recovery by a two-step flocculation method, involving chemical coagulants in the first step and

bacterial biomass used in the second step to enhance the flocculation. Up to 97 % of the microalgal biomass was recovered using 16 mg polymer/g algae and 0.1 g algae/g bacterial biomass. Moreover, the energy recovery by the short-SRT EBPR system combined with microalgal cultivation was assessed via biomethane potential tests. Up to  $560 \pm 24$  ml CH<sub>4</sub>/gVS methane yield was obtained by co-digesting bacterial biomass collected after the anaerobic phase and microalgal biomass. The energy recovery obtained from the short-SRT EBPR system is about 40% of the influent chemical energy.

## **Keywords**

Green microalgae; enhanced biological phosphorus removal; bioflocculation; co-digestion; energy recovery

## **1. Introduction**

Due to the challenges related to greenhouse gas emissions, decreasing fossil fuel reserves and global and national pressure, new solutions are sought to produce renewable energy including the use of biomass for biofuel production. However, first generation biofuels (derived from agricultural crops) are of questionable sustainability as they compete for land with food crops, thereby affecting the global food security [1,2]. Similarly, second generation biofuels, e.g. non-food energy crops (e.g. vegetative grasses or short rotation forests), agricultural and forest residues, compete for land use in some cases and there are technological difficulties related to the conversion processes [1]. Third generation biofuels such as microalgae have the advantages that they can be produced all year round, do not compete food production as they can be grown on non-arable land, have rapid growth rates and the biochemical composition can be manipulated by varying cultivation conditions and strains [1,3]. The cultivation of microalgae for biofuel production can be economically viable when

46 coupled with wastewater treatment [3–6] which provides the water and nutrients (nitrogen and  
47 phosphorous) required for growth [7].

48 Conventional wastewater treatment has a high energy demand required mainly by the aeration  
49 process whereby organic carbon present in wastewater is oxidized to CO<sub>2</sub> and nitrification takes  
50 place under long sludge ages [8]. This leads to the loss of the energy potential of the activated  
51 sludge [9] together with the loss of nutrients (nitrogen and phosphorus) [8]. Short solids retention  
52 time (SRT) activated sludge systems propose a solution whereby rather than the oxidization of  
53 organic carbon, activated sludge preserves the organic carbon promoting higher potential for energy  
54 recovery [10].

55 Bacterial-algal systems can be coupled with wastewater treatment, whereby nutrients and energy  
56 can be recovered [3]. In a novel wastewater resource recovery approach, Valverde-Pérez et al. [11]  
57 proposed an enhanced biological phosphorus recovery and removal (EBP2R) process, able to  
58 provide optimal culture media for downstream microalgal cultivation. The system referred to as  
59 TRENS, consists of a modified short-SRT EBP2R process where an additional solid-liquid  
60 separation is included after the anaerobic phase (Fig. S1, Supporting Information, SI). Under  
61 anaerobic conditions, phosphorus accumulating organisms (PAO) take up the volatile fatty acids  
62 (VFA) from the wastewater and store them as polyhydroxyalkanoates (PHA) intracellularly while  
63 releasing intracellular phosphorus (poly-P) [12]. Under aerobic conditions the stored PHA are used  
64 to produce energy for biomass growth as well as phosphorus uptake and storage [12]. Thus, the  
65 effluent water after the solid-liquid separation after the anaerobic phase is rich in phosphorus, whilst  
66 the effluent after the solid-liquid separation after the aerobic phase is rich in nitrogen. The short-  
67 SRT EBP2R can provide optimal cultivation medium to a downstream photobioreactor (PBR) by  
68 mixing the phosphorus and nitrogen rich effluent streams in an optimal ratio.

69 When microalgal cultivation is coupled with wastewater treatment the lipid content of the  
70 microalgae is fairly low (4.9-11.3%) due to the relatively high nutrients supplied [3,13]. It is  
71 energetically favourable to apply anaerobic digestion when the lipid concentration is lower than  
72 40% [14]. In addition, anaerobic digestion is applicable for biomasses with high moisture content  
73 (80-90%), which makes it suitable for microalgal biomass conversion [1,15]. Thus, anaerobic  
74 digestion is the preferred route over biodiesel production when energy recovery is considered from  
75 microalgae cultivated on wastewater resources [13]. The nutrient rich effluents of the anaerobic  
76 digestion can be used for further cultivation of microalgae [1]. Anaerobic digesters are in many  
77 cases available in the existing wastewater treatment plants and biogas production can be increased  
78 by co-digestion of microalgae and activated sludge [16]. Nonetheless, not all microalgal species are  
79 suitable for biogas production, mainly due to their cell wall structure and their high nitrogen content  
80 [14,17].

81 A C/N ratio of 20 (g/g) is suitable for optimal digestion conditions [4,18]. While, in freshwater  
82 microalgae it is typically around 10 [14,19]. Many studies proposed co-digestion with other  
83 biomass sources, e.g. activated sludge, to improve digestibility by balancing the C/N ratio, thereby  
84 providing optimal nutrient balance for enhanced methane yield [3,15,16,18]. Additionally, the co-  
85 digestion of various waste lines reduce costs by using a single anaerobic digester unit for digestion  
86 of multiple substrates [3].

87 The major bottleneck of microalgal cultivation for biogas production is the cost related to biomass  
88 harvesting [15,20,21]. Energy-intensive and expensive methods, e.g. centrifugation or membrane  
89 technologies [20], are only applicable when the biomass is used to produce high value products  
90 [21]. Thus simple harvesting methods are required for reliable and safe downstream applications  
91 [3].

92 Flocculation is an alternative and cheap harvesting method [20,22]. During coagulation the negative  
93 surface charge of microalgae, caused largely by the presence of carboxyl groups, is destabilized.  
94 This is followed by a second flocculation step whereby aggregates are formed, thus promoting more  
95 effective gravity sedimentation [21,23]. Iron or aluminium salts, which form positively charged  
96 hydroxides when dissolved in water, are successfully used as coagulants that neutralize the negative  
97 algal cells promoting aggregate formation [24].  $\text{AlCl}_3$  addition is a common method in wastewater  
98 treatment to enhance the coagulation-flocculation process [25]. Nevertheless, aluminium salts  
99 require high dosage and the downstream usage is limited due to toxicity [21]. Cationic polymers can  
100 induce flocculation of algal biomass by surface charge neutralization or by inter-cellular bridging  
101 [24]. The effectiveness of the polymers depends on their size and charge density. Compared to  
102 metal salts, polymers usually operate at lower dosages [21]. Flocculation efficiency by polymers  
103 declines at high dosages due to restabilisation [20,21]. Bioflocculation has also been proposed: in  
104 this case a specific bacteria, fungi or algae are added to the microalgal culture promoting  
105 flocculation [20,26].

106 Bacterial-algal systems have the potential to recover energy through biomass production. Thus, a  
107 cost-effective harvesting method is needed whereby the algal and bacterial biomass can be  
108 recovered. The objectives of this study are (i) to test the effect of different chemical flocculants on  
109 microalgal recovery; (ii) to develop a cost-effective method of harvesting microalgae via a two-step  
110 flocculation using cationic polymer for destabilisation of microalgae and bacterial biomass from a  
111 short-SRT EBPR system to enhance the aggregation of the algae; (iii) to optimize the cationic  
112 polymer dosing; (iv) to assess the effect of different algae/bacterial biomass ratios and the effect of  
113 bacterial biomass settleability on algal biomass recovery; and (v) to assess the methane production  
114 potential by co-digestion of the harvested bacterial-algal biomass.

## 2. Materials and methods

### 2.1. Microalgal cultivation and EBPR operation

#### 2.1.1. Algal biomass used for pre-testing different coagulants

We cultivated a mixed green microalgal consortium consisting mainly of *Chlorella sorokiniana* and *Scenedesmus* sp. (see Wágner et al. [27]). The consortium was cultivated with effluent water from the Lundtofte WWTP (Denmark). Ammonium and phosphorus were spiked to reach 20 mg/L NH<sub>4</sub>-N and 2.75 mg/L PO<sub>4</sub>-P (16 N-to-P ratio) in the microalgal batch cultivation. 2 L glass reactors were used to cultivate the algae with constant stirring at 180 rpm using magnetic stirrers and with aeration with CO<sub>2</sub> enriched air (5 % CO<sub>2</sub>) at a flow rate of 10 L/h. Light was supplied from the two sides of the batches with fluorescent lamps (18 W, GroLux, Sylvania®, USA), providing 160 μmol photons m<sup>-2</sup> s<sup>-1</sup> continuously. The temperature in the room was regulated at 20 °C. 80% of the algal suspension was removed every 2-3 days from the batch reactor and the reactor was refilled with new effluent water. The pH of the algal culture varied between 6.84 and 7.95 during the experiments. The TSS of the algal suspension used for flocculation varied between 0.29 and 0.37 g/L. The algal TSS and OD values used for each flocculation experiment are reported in Table S1, SI.

#### 2.1.2. Algal and bacterial biomass used for the two-step flocculation

The same mixed green microalgal consortium was used in the two-step flocculation experiments. The microalgal culture was grown on effluent water from a laboratory scale EBPR system [28] operated at 3-3.5 days SRT as a sequencing batch reactor (SBR) (fed with pre-clarified wastewater from Lundtofte WWTP, Denmark). The ammonium and ortho-phosphate concentrations were adjusted to an N/P molar ratio of 17 in the beginning of each microalgal batch (adjusted to 23 mg/L

137 NH<sub>4</sub>-N and 3 mg/L PO<sub>4</sub>-P). 1.5 L glass reactors were used to cultivate the algae with constant  
138 aeration with CO<sub>2</sub> enriched air (5 % CO<sub>2</sub>) at a flow rate of 10 L/h. Light was supplied from the top  
139 of the batch reactor continuously with a custom-built lamp, providing 500  $\mu\text{mol photons m}^{-2} \text{ s}^{-1}$ ,  
140 with a metal-halide light bulb (OSRAM®, Germany). The reactors were kept at room temperature.  
141 The pH of the algal culture varied in the range of 7 – 8.5 during the experiments. 60% of the algal  
142 suspension was removed every 2-3 days and the batch reactor was refilled with new effluent water  
143 from the EBPR system (adjusted to N/P molar ratio of 17). The TSS of the algal suspension varied  
144 in the range of 0.27 - 0.52 g/L during the experiments. The algal TSS and OD values used for each  
145 flocculation experiment are reported in Table S1, SI. The bacterial biomass was taken from the  
146 short-SRT EBPR system using two biomass removal strategies: i) bacterial biomass removed at the  
147 end of the anaerobic phase; ii) bacterial biomass removed at the end of the aerobic phase. Samples  
148 for the biogas tests were taken during the course of 1 month, whilst the samples for the flocculation  
149 tests were taken throughout a 6 months period. Considering the use of real wastewater and the  
150 length of the experiments, results obtained can represent the effect of variability in used water  
151 resources, thereby allowing inferring experimental results more representative to real systems.

## 152 **2.2 Flocculation and bioflocculation tests**

### 153 *2.2.1. Pre-testing of flocculation with different coagulants*

154 The coagulation aids included AlCl<sub>3</sub> (Sigma Aldrich), the cationic biopolymer Greenfloc 120  
155 (Hydra 2002, Hungary) and the cationic polymer Poly(diallyldimethylammonium chloride)  
156 (PDADMAC) (Sigma Aldrich). Jar testing was done based on the standard practice for coagulation-  
157 flocculation jar test of water [29]. Each flocculation test included the parallel testing of six 1 L jars  
158 using a mixing device with a rotating impeller mixing each jar. In each jar, a chosen coagulation aid  
159 was spiked at varying concentrations, while mixing. During the first 2 min, a high mixing rate (150



rpm) was applied to evenly mix the added coagulants with the algae. This step was followed by a slow mixing at 25 rpm to let the particles flocculate for 10 min. After the flocculation, the mixing was stopped and the impellers were removed from the solution to initiate 30 min settling.

#### 2.2.2. Two-step flocculation

Two flocculation methods were tested in 1 L jars: i) the bacterial biomass was used to flocculate algae and ii) a two-step flocculation was tested where in the first step the algae were coagulated first with the cationic polymer PDADMAC and then bacterial biomass was added in the second step to enhance the flocculation. In the first case, high mixing was applied at 100 rpm for 2 min and different slow mixing times (i.e. 10 min, 1 h and 3 h) were tested at 20 rpm. For the two-step flocculation method the duration of flocculation is given in Table 1. Both methods were followed by 30 min settling period.

### <Table 1>

## 2.3 Biomethane potential

#### 2.3.1 Samples combinations for biomethane potential assays

The settled biomass samples were collected after the two-step flocculation tests and kept at -20 °C until further use. Additionally to the flocculated samples, microalgal and bacterial biomass were collected to assess the biomethane potential (BMP) of the single biomasses and their combination without polymer. All the samples were kept frozen until the BMP assays were set up. In total, eight different scenarios were assessed in BMP assays using triplicates: algae, algae + polymer (20 mg polymer/g algae), activated sludge (AS) alone (taken after the aerobic and anaerobic phase),  $AS_{AE}/AS_{AN}$  + algae (ratio 0.1 of g algae/g AS) and lastly  $AS_{AE}/AS_{AN}$  + algae + polymer (ratio 0.1 of g algae/g AS, 20 mg polymer/g algae). The SRT of the EBPR system was 3.5 days in all samples

182 used for BMP tests. The amount of substrate and inoculum as well as the total solids (TS) and  
183 volatile solids (VS) concentrations of each sample are reported in Table S2, SI. The composition of  
184 the substrates is reported in Table S3, SI.

### 185 2.3.2 Biomethane potential assays set up

186 The set up for the BMP assays was adapted from Angelidaki et al. [30]. Inoculum for the assay test  
187 was taken from the mesophilic anaerobic digester of Lundtofte WWTP. The defrosted biomass  
188 samples were added together with the inoculum to 1200 ml bottles, flushed with N<sub>2</sub> for 5 minutes,  
189 closed with air tight rubber stoppers, sealed with screw caps and stored at mesophilic conditions at  
190 37 °C. Avicel pH-101 was used as substrate for positive control and DI water as substrate for  
191 negative control. The methane concentration produced in the bottles was measured every 2-3 days  
192 using the GC-2010 (Shimadzu, Japan). On each measurement day a calibration curve was set up  
193 using 5, 10, 40 and 60 % methane content to be able to relate the methane content of the samples.  
194 Each time 50-100 µL sample was taken from the headspace using a pressure syringe and was  
195 injected into the GC.

## 196 2.4 Analytical methods and calculations

197 The optical density (OD) at 750 nm was measured in the initial algae suspension and in the bacterial  
198 biomass and was monitored during the 30 min settling by taking samples 5 cm below the liquid  
199 surface (approximately at 700 ml in the 1 L jar) to maintain uniform sampling in all experiments  
200 (adapted from [31]). In case the biomass blanket height was above 700 ml, due to poor settling of  
201 bacterial biomass, the final OD sample at 30 min was taken from the supernatant above the biomass  
202 height in order to calculate microalgal biomass recovery. OD samples were collected in 24 well  
203 microplates and OD measurements were conducted in the end of each jar test using Synergy Mx

204 Microplate Reader® (Biotek). The recovery was calculated based on the following expression  
205 (based on [32]):

$$206 \text{ Recovery (\%)} = \frac{OD_{750init} - OD_{750,30min}}{OD_{750init}} * 100 \quad \text{Eq. 1}$$

207 where  $OD_{750init}$  is the OD of the initial suspension,  $OD_{750,30min}$  is the OD measured at the end of the  
208 settling phase. Average recovery and the standard deviation were calculated based on the last three  
209 measurement points of the 30 min settling period.

210 The price of harvesting the microalgal biomass using different coagulants was calculated. The  
211 estimations were based on the price of  $AlCl_3$ , Greenfloc 120 and PDADMAC reported by the  
212 suppliers (see section 2.2.1) in 2014, when the experiments were conducted.

213 Total suspended solids (TSS), volatile suspended solids (VSS) and TS and VS of the algae and  
214 bacterial biomass were measured based on standard methods [33]. Sludge volume index (SVI) of  
215 the bacterial biomass was measured in 1L cylinder based on Ekama et al. [34]. Total nitrogen and  
216 phosphorus and COD measurements in the samples were done using commercial test kits (Hach-  
217 Lange®, USA) and measured with spectrophotometer DR2800 (Hach-Lange).

218 The average methane yield and the standard deviation were calculated based on triplicate batch tests  
219 conducted for each scenario. Each replicate was collected on a different day as the amount of  
220 bacterial and algal biomass was not enough for more than one flocculation test.

221 We calculated the methane yield produced during the co-digestion of algae and bacteria based on  
222 Wang et al. [35]:

$$223 \text{ Calculated methane yield} = Y_s * C_s + Y_a * C_a \quad \text{Eq. 2}$$

224 where  $Y_s$  and  $Y_a$  is the measured methane yield of bacterial biomass and algae produced  
225 individually and  $C_s$  and  $C_a$  is the mixing fraction of bacteria and algae in the co-digestion scenario.  
226 These numbers were confronted with the measured methane yields of the co-digestion scenarios,  
227 assessing the synergistic effect of co-digestion, and results are shown in section 3.3.

228 First-order kinetics is used to estimate the hydrolysis constant ( $k_h$ ) and the ultimate methane  
229 production ( $B_\infty$ ) based on Angelidaki et al. [30] and Ge et al. [36]:

230 
$$B = B_\infty(1 - e^{-k_h * t}) \quad \text{Eq. 3}$$

231 where B is the methane produced at a given time.

232 Student's t-tests were conducted, based on the triplicate samples, to compare the measured methane  
233 yields for the different digestion scenarios, using SigmaPlot (USA).

### 234 **3. Results and discussion**

#### 235 **3.1 Flocculation of microalgae using different coagulants**

236  $AlCl_3$  was effective for harvesting the microalgae, and dosing at 100 mg  $AlCl_3$ /g algae resulted in a  
237 recovery of 97% after 30 min settling time (Fig. 1a). A different trend in the recovery was obtained  
238 when using polymers. The optimum Greenfloc 120 dosing was 30 mg GF/g algae, yielding 84%  
239 recovery, based on visual observations (Fig. 1a). However, when a higher polymer concentration  
240 was added to the suspension, the recovery decreased. This is the likely consequence of the  
241 restabilisation process whereby increasing the amount of positive charges will result in repulsion  
242 between the aggregates [21]. Similarly, when coagulation was induced by the addition of  
243 PDADMAC an optimum recovery of 92% was found at the intermediary dose of ca. 27 mg  
244 PDADMAC/g algae (Fig. 1a).

**<Figure 1>**

The optimal dosage of  $\text{AlCl}_3$  is within the reported range for aluminium salts, 85 - 503 mg aluminium salt/g algae [37,38]. The optimal cationic polymer dosage reported in the literature, (e.g. Roselet et al. [21], 19.23 - 57.69 mg polymer /g algae) is in agreement with the range found in our study. Restabilisation, as we observed, is not always reported: whilst some observed restabilisation (e.g. [21,39]), others (e.g. [40]) found no restabilisation as the amount of polymer was increased. Gerde et al. [31] observed restabilisation at lower biomass concentrations (0.05-0.2 g/L), whilst at high biomass concentrations (1 g/L) this effect was not visible within the same dosing range. This may be important when considering cultivation conditions and reactor operation. Depending on the cultivation conditions, i.e. open ponds or closed photobioreactors, the biomass concentration during the cultivation can vary from 0.1 – 4 g/L [41]. The maximum biomass concentration that can be reached in open ponds and closed photobioreactors is 1 g/L and 4 g/L, respectively [42]. In this study, the system resembles an open pond reactor with comparably low biomass concentration, which may lead to algae restabilisation.

The optimum  $\text{AlCl}_3$  dosage would result in a cost of approximately 6000 EUR/ton algae harvested (Fig. 1b), whilst the use of Greenfloc 120 and PDADMAC at an optimal dose would be 30 – 60 times lower, about 100 and 900 EUR/ton algae, respectively (Fig. 1b). Moreover, the use of aluminium salts may pose negative effects in terms of downstream recycling of the effluent water [43] that can limit further usage of the biomass for land application or biogas production [44] due to their substantial toxic effects [45]. However, according to Udom et al. [46] polymers have substantial environmental and economic costs related to their production process. The greenhouse gas emission and the energy consumption costs related to the production of polymers are found to be nearly ten times higher than for ferric chloride [46]. Thus even though we save on the operational costs due to the lower dosage, there are additional energy-expensive costs related to the

269 use of polymers. Recovery rates obtained with PDADMAC and Greenfloc are not significantly  
270 different (based on t-test,  $P>0.05$ ). Due to the similar performance and the easier access on the  
271 market (Greenfloc had limited availability for research purposes) PDADMAC was chosen for  
272 further assessment.

### 273 **3.2 Bioflocculation of microalgae – an innovative approach**

274 Flocculation of microalgae with bacterial biomass by 10 min flocculation time resulted in 40%  
275 recovery (Fig. S2, SI). Furthermore, increasing the mixing time did not improve the microalgal  
276 recovery (Fig. S2, SI), in contrast to the observations by Manheim and Nelson [26]. Alternatively,  
277 we considered addition of cationic polymer as coagulation aid to destabilise the microalgae before  
278 the addition of bacterial biomass and to enhance the separation of microalgae.

279 Different concentrations of polymer addition were tested (Fig. 2a). With increasing polymer  
280 concentrations the microalgal recovery increased as well. This suggests that as the algal cells  
281 aggregate into larger particles the probability of collision with the bacterial biomass flocs can  
282 significantly increase, thereby increasing flocculation efficiency. No restabilisation effect was  
283 observed at the assessed dosing, likely due to the high concentrations of bacterial biomass addition,  
284 in accordance with the findings of Gerde et al. [31]. However, we *note* that there might be  
285 restabilisation at higher polymer dosages [31]. Recovery rate ca. 97% was obtained using a polymer  
286 dosage of 16 mg/g algae at a 0.1 g algae/g bacterial biomass ratio. Using bacterial biomass and  
287 polymer for the coagulation-flocculation can reduce the polymer dosing by 40% compared to the  
288 scenario when only algae was flocculated with the cationic polymer, PDADMAC (Fig. 2b).  
289 Consequently, harvesting costs are reduced.

290 <Figure 2>

291 The mixing ratio was fixed at 0.1 g algae/g bacterial biomass for most experiments. With increasing  
292 algae-to-bacterial biomass ratio, maintaining the same polymer dosage (16 mg polymer/g algae), the  
293 microalgal recovery decreased, on average, with more than 50% (Fig. 3). This shows the  
294 importance of assessing the optimum polymer dosing for the operational algal-to-bacterial biomass  
295 mixing ratio. However, some deviation from the optimum ratio will not compromise the recovery as  
296 we find similar recovery at 0.2 g algae/g bacterial biomass.

297 **<Figure 3>**

298 The flocculation efficiency of microalgae and the required dosing of coagulants and flocculants can  
299 be influenced by factors, such as mixing time [26], pH [40] or the growth stage and age of the  
300 microalgal culture [20]. Autoflocculation due to the increase of pH typically occurs above pH=10  
301 [32,47]. Therefore, the effect of pH should be negligible as it was kept below 8.5 during the  
302 experiments (section 2.1.2). We assume that the algae samples were in similar physiological state  
303 for all flocculation experiments as the algal biomass was harvested every 2-3 days. Moreover, it is  
304 reported in the literature that a certain concentration of inorganic coagulant can result in different  
305 recovery for different microalgal species [21,48]. Thus in a mixed microalgal culture if the  
306 dominance of the microalgal species changes the flocculation efficiency can potentially change.  
307 However, microbial community was not monitored in this study. In addition, this effect can be  
308 potentially compensated by the addition of the bacterial biomass as it can hinder the restabilisation  
309 effect in the tested dosing range.

310 The settleability of the bacterial biomass varied in the EBPR system due to filamentous bulking,  
311 which could have affected the observable flocculation efficiency. During the experiments, the SVI  
312 (an indicator of the settling characteristics of the bacterial biomass [34]) varied between 180 and  
313 760 ml/g, which allowed us to test the effect of bacterial biomass settling on the recovery of

microalgal biomass. The separation of the bacterial-microbial biomass after flocculation might be limited if bulking (high SVI) bacterial biomass is used (Fig. S3, SI). Even though the separation of the bacterial-algal biomass deteriorates, the recovery of microalgae is not affected by the increased SVI of bacterial biomass (Fig. 4). Thus the bacterial composition has no particular effect on the microalgal recovery. Additionally, the commonly believed particle screening effect of filamentous bacteria, whereby filaments are the backbone of flocs, responsible for incorporating colloidal particles into the floc structure [49] does not seem to play a significant role in the flocculation of algal biomass. Instead, the surface charge of the biomass may control the flocculation behaviour. The negative surface charge of the biomass comes in contact with the positive charges of the polymer that is attached to the algae, thereby promoting aggregate formation. Despite the low impact on microalgal recovery, from an operational perspective, the abundance of filamentous organisms in bacterial biomass is an important factor, responsible for causing foaming in anaerobic digesters, that could deteriorate digester performance [50].

**<Figure 4>**

### **3.3 Co-digestion of algal and bacterial biomass**

The biomethane potential (BMP) obtained after 27 days of digestion of the microalgal biomass is  $331 \pm 76$  ml CH<sub>4</sub>/gVS (Fig. 5). The methane yield obtained by digesting solely microalgal biomass is reported in a wide range in the literature (143-497 ml CH<sub>4</sub>/gVS) [2,18], which also corresponds to the results obtained in this study. Wang and Park [13] report slightly lower yields (230 ml CH<sub>4</sub>/gVS) when digesting *Chlorella sp.*, whereas Olsson et al. [51] report similar values when digesting algae (mixed green microalgal culture – approx. 370 ml CH<sub>4</sub> /gVS). In this study, we obtain similar methane yields without the pre-treatment of algae to those that are reported with different pre-treatment options in the literature (105-336 ml CH<sub>4</sub> /gVS ) [52]. Nevertheless,



337 Anbalagan et al. [53] showed that pre-treatment does not always result in higher BMP as, e.g., the  
338 nutrient balance and type of algae are also important factors affecting the methane yield. In  
339 addition, the variations reported through the literature might be due to the dominance of different  
340 species in a mixed culture, which can affect the biogas potential [51]. The addition of polymer does  
341 not significantly affect the biomethane potential of the microalgae.

342 **<Figure 5>**

343 The biomethane potential of the biomass removed after the aerobic phase is  $363 \pm 68$  ml  $\text{CH}_4/\text{gVS}$ ,  
344 whereas, for biomass removed after the anaerobic phase is  $449 \pm 17$  ml  $\text{CH}_4/\text{gVS}$  (Fig. 5, Table 2).  
345 The difference between these two digestion scenarios is not significant. Kuglarz et al. [54] reported  
346 generally lower methane yields compared to our measurements when digesting bacterial biomass,  
347 taken from a conventional wastewater treatment plant, even after pre-treatment (approx. 270 ml  
348  $\text{CH}_4/\text{gVS}$ ). It is reported by Bolzonella et al. [55] that higher biogas potential is reached when  
349 bacterial biomass is taken from shorter SRT (8 d in their study) wastewater treatment systems  
350 compared to systems with longer SRT (45 d in their study). Literature is relatively scarce in regard  
351 to assessing the biogas potential of short-SRT bacterial biomass. The study by Ge et al. [36] reports  
352 similar results to those obtained with the biomass removed after the aerobic phase in our study  
353 (BMP:  $306.4 \pm 12.6 - 332.4 \pm 19.7$  ml  $\text{CH}_4/\text{gVS}$ ). These BMP values are significantly lower than that  
354 obtained using bacterial biomass collected after the anaerobic phase in the short-SRT EBP2R  
355 process.

356 The hydrolysis rate and the ultimate biomethane potential were estimated by fitting Eq. 3 on the  
357 data obtained during the 27-day long digestion tests (Table 2). The  $k_h$  for the anaerobic digestion of  
358 microalgae found in this study is higher than those reported in the literature [56]. Only Ge et al. [36]

report  $k_h$  values ( $0.19 \pm 0.02 - 0.22 \pm 0.04 \text{ d}^{-1}$ ) that are comparable to those obtained in this study with aerobically and anaerobically harvested bacterial biomass.

**<Table 2>**

The co-digestion of the bacterial biomass removed after the aerobic phase and microalgal biomass resulted in higher amount of methane produced than by digesting them individually (not significantly different,  $P > 0.05$ ). Whereas, the co-digestion of algae with bacterial biomass collected after the anaerobic phase resulted in significantly higher methane yields compared to digesting the algal and bacterial biomass separately, based on the results of the t-test (Table S4, SI). Values of the measured yield obtained with and without polymer are  $424 \pm 14$  (Algae + AS<sub>AE</sub> + poly) and  $400 \pm 22$  (Algae + AS<sub>AE</sub>), respectively, expressed as ml CH<sub>4</sub>/gVS. This is approximately 10% higher than that reported in the literature [51]. The calculated methane yield (Eq. 2) for the co-digestion scenario with the bacterial biomass collected after the aerobic phase is  $360 \pm 62$  ml CH<sub>4</sub>/gVS (Table 2). We find no significant difference ( $P > 0.05$ ) between calculated (based on Eq. 2) and measured values. Thus, our results suggest no synergistic effect when co-digesting algae and bacterial biomass removed after the aerobic phase, in agreement with the literature [15]. The calculated methane yield (Eq. 2) for the co-digestion scenario with the bacterial biomass collected after the anaerobic phase is  $437 \pm 17$  ml CH<sub>4</sub>/gVS (Table 2). Values of the measured yield obtained with and without polymer are  $528 \pm 28$  (Algae + AS<sub>AN</sub> + poly) and  $560 \pm 24$  (Algae + AS<sub>AN</sub>), respectively, expressed as ml CH<sub>4</sub>/gVS. Thus, we find that the measured values are significantly higher ( $P < 0.05$ ) than the calculated values based on Eq. 2. These results suggest that – as opposed to using the bacterial biomass removed after the aerobic phase – there may be synergistic effects of co-digesting algae with biomass removed after the anaerobic phase, compared to digesting them individually. Furthermore, the biomethane potential of the co-digestion was significantly higher ( $P < 0.05$ ) with bacterial biomass taken after the anaerobic phase than with biomass taken from the aerobic phase

(Fig. 5, Table 2). The higher co-digestion potential with bacterial biomass removed after the anaerobic phase could be related to their content of PHA. It is well known that PAO store VFA in the form of PHA under anaerobic conditions [12] which is a more easily available substrate for the digestion than other organic materials, e.g. the cell wall. Interestingly, we do not find significant difference between the single digestion of biomass taken after the anaerobic and aerobic phase. This suggests that the single digestion of the bacterial biomass may be nutrient limited, thereby producing less methane. Whereas, co-digestion with a nutrient rich biomass, e.g. microalgae, could provide the additional nutrients needed to digest the increased organic carbon content, resulting in higher methane potential. Additionally, other studies suggest that the increased micronutrients content added with the microalgal biomass can improve the biogas potential when co-digesting algae with bacterial biomass [51]. Presence of the cationic polymer after the flocculation did not affect the co-digestion potential (no significant effect,  $P>0.05$ ), in agreement with the literature [23] (Table 2).

#### <Figure 6>

It was estimated that  $0.4\pm0.02$  g  $\text{COD}_{\text{CH}_4}/\text{g COD}_{\text{inf}}$  and  $0.36\pm0.07$  g  $\text{COD}_{\text{CH}_4}/\text{g COD}_{\text{inf}}$  energy can be recovered in the form of methane, for the anaerobic and aerobic bacterial biomass removal scenarios, respectively. These results are considerably higher than that obtained for conventional activated sludge systems ( $0.07\pm0.06$  g  $\text{COD}_{\text{CH}_4}/\text{g COD}_{\text{inf}}$ ) and are comparable to other short-SRT activated sludge systems ( $0.36\pm0.08$  g  $\text{COD}_{\text{CH}_4}/\text{g COD}_{\text{inf}}$ ) [57]. The assessment of the distribution of the influent COD (Fig. 7) shows that not only approximately 40% of the influent COD is recovered as methane, but the EBPR system effectively removes most of the influent COD leaving up to maximum 10% as inert material in the effluent wastewater. Compared to other short-SRT systems, these results show significantly lower loss of COD in the effluent while directing comparable amounts into the biomass [9,57,58]. This facilitates downstream unit process operation,

e.g. microalgal cultivation or autotrophic nitrogen removal based technologies. Taken together, our results suggest that there is an increased methane potential of the co-digestion of bacterial biomass generated through the short-SRT EBPR system and microalgal biomass. Furthermore, the associated environmental costs are lower as pre-treatment of the biomass is not necessary and less energy is invested for pollutant removal compared to systems with long solid retention times. Moreover, the COD recovered through the EBPR process is comparable to that found in the literature for other short-SRT systems, leaving up to maximum 10% as inert material in the effluent water.

<Figure 7>

#### 4. Conclusions

In this study we assessed an innovative bioflocculation method to harvest microalgal biomass and evaluated the potential to produce methane through digestion and co-digestion of the recovered microalgal biomass with bacterial biomass derived from an EBPR system. We found that:

- The cationic polymer (PDADMAC) proved to be a cost-efficient way to harvest microalgal biomass resulting in 92% recovery with 27 mg polyelectrolyte/g algae dosing.
- An innovative bioflocculation method was introduced to separate microalgal biomass. Bacterial biomass was used as a flocculant after the destabilization of microalgae with cationic polymer, whereby up to 97 % recovery was reached with 16 mg polymer /g algae and 0.1 g algae/g bacterial biomass ratio.
- The highest methane yield was found at  $560 \pm 24$  mlCH<sub>4</sub>/gVS when microalgae and anaerobically harvested bacterial biomass were co-digested.
- The short-SRT EBPR process combined with microalgal cultivation can serve as an energy recovery system whereby up to 40 % of the incoming COD is converted to methane through

anaerobic digestion. Moreover, the COD is successfully removed through the process, thereby leaving only up to 10% inert COD in the effluent wastewater. However, the optimization of the nutrient balance during the anaerobic digestion by co-digestion with nutrient rich biomass, e.g. microalgae, is important to potentially increase the COD recovery.

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