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Title: Exploiting the unwanted: sulphate reduction enables phosphate recovery from energy-rich sludge during anaerobic digestion

Running title: Phosphate recovery through sulphate reduction

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22 **Abstract**

23 Anaerobic digestion is shifting from a single-purpose technology for renewable energy
24 recovery from organic waste streams to a process for integrated resource recovery. The
25 valorisation of high-rate energy- and phosphorus-rich sludge creates the opportunity for their
26 combined recovery. This phosphate is present in a precipitated form in the sludge, and its
27 release into the liquid phase is an important issue before recovery can be achieved. The
28 objective of this research was to exploit the “unwanted” sulphate reduction process for the
29 release of phosphate into the liquid phase during anaerobic digestion, thus, making it available
30 for recovery. Two different treatments were considered, *i.e.*, a control digester and a digester to
31 which sulphate was added, each operated in triplicate for a period of 119 days. The control
32 digester showed stable methane production at $628 \pm 103 \text{ mL CH}_4 \text{ L}^{-1} \text{ d}^{-1}$, with a feedstock COD
33 (chemical oxygen demand) conversion efficiency of $89.5 \pm 14.6 \%$. In contrast, the digester
34 with sulphate addition showed a $29.9 \pm 15.3 \%$ decrease in methane production, reaching an
35 “inhibited steady state”, but phosphate release into the liquid phase increased with a factor 4.5,
36 compared to the control digester. This inhibited steady state coincided with a clear shift from a
37 Methanosaetaceae to a Methanosarcinaceae dominated methanogenic community. Overall, the
38 sulphate reduction process allows phosphate release during the anaerobic digestion process,
39 yet, at the cost of a reduced methane production rate.

40

41 **Keywords:** Biogas, methanogenesis, resource recovery, sulphate reducing bacteria

42

43 **1. Introduction**

44 Anaerobic digestion (AD) has been a key technology for the recovery of renewable energy from
45 organic waste streams for decades. Initially, however, the main purpose of AD was the
46 stabilisation of organic waste streams to avoid environmental pollution (Acosta and De Vrieze
47 2018). The ability to use the energy-rich methane in a combined heat and power (CHP) unit for
48 electricity and heat production quickly allowed the transition of AD from a waste treatment
49 technology to an integrated system for renewable energy recovery. Different organic waste
50 streams, such as animal manure (Holm-Nielsen et al. 2009), waste activated sludge (Appels et
51 al. 2008), the organic fraction of municipal solid waste (Hartmann and Ahring 2006) have been
52 valorised through AD, either as such or through co-digestion with other waste streams (Björn
53 et al. 2017, Mata-Alvarez et al. 2011). In the framework of the current transition from “waste-
54 to-energy” to “waste-to-resource”, the recovery of nutrients, in addition to energy, has become
55 more and more pressing to (1) safeguard natural resources and (2) ensure long-term economic
56 viability of the AD process.

57 The recovery of nutrients in refined products through AD has been demonstrated through
58 numerous technologies in multiple configurations. Stripping/absorption is a well-established
59 method for ammonia recovery in AD, either as pre-treatment (Bonmatí and Flotats 2003, Zhang
60 et al. 2012), in side stream (Pedizzi et al. 2017, Serna-Maza et al. 2014), or post-treatment
61 technology (Bonmatí and Flotats 2003, Gustin and Marinsek-Logar 2011). An alternative
62 approach involves the electrochemical recovery of ammonium and other cations *via* a cation
63 exchange membrane (Desloover et al. 2012), thus, allowing their recovery in a “clean” stream
64 (De Vrieze et al. 2018). Both technologies are often combined for efficient ammonia recovery,
65 as electrochemical extraction, followed by stripping/absorption enables the creation of a
66 continuous concentration gradient (Desloover et al. 2015, Zhang and Angelidaki 2015).

67 The recovery of phosphorus in combination with AD is often problematic, because phosphates

68 precipitate with multivalent cations, such as Ca^{2+} , Mg^{2+} and Fe^{2+} , say in the case of waste
69 activated sludge (De Vrieze et al. 2016). Hence, the release and recovery of phosphorus from
70 waste activated sludge requires alternative approaches, such as a microwave treatment (Liao et
71 al. 2005), a free ammonia-based pre-treatment (Xu et al. 2018), or pressurized AD (Latif et al.
72 2018). This allows the release of phosphate into the liquid phase, and the potential for
73 subsequent recovery either through (1) struvite precipitation, to be used as slow-release
74 fertilizer (Li et al. 2019, Vaneeckhaute et al. 2018), (2) electrodialysis, using an anion exchange
75 (Ebbers et al. 2015) or bipolar (Shi et al. 2018) membrane system, or (3) a combination thereof
76 (Zhang et al. 2013). These different technologies for the release of phosphate from the solid
77 phase, however, require the input of chemicals and/or a coincide with an additional energy cost
78 per unit of phosphorus released. Given the low and variable global market value of phosphorus,
79 *i.e.*, € 350-1200 tonne⁻¹ P for phosphate rock with a P_2O_5 content of 30% since 2010, alternative
80 low-cost strategies for phosphate release should be targeted (Mayer et al. 2016).

81 Combining AD of waste activated sludge with *in situ* sulphate reduction by sulphate reducing
82 bacteria could be an alternative approach for phosphate release into the liquid phase with no
83 additional requirements in terms of energy or chemicals by using sulphate-rich waste streams
84 as co-feedstock, such as vinasse or paper mill wastewater (Pokhrel and Viraraghavan 2004,
85 Rodrigues Reis and Hu 2017). As the solubility product of multivalent cations with sulphides
86 is conventionally lower than with phosphates, the *in situ* formation of sulphides could release
87 phosphate into the liquid phase. The reduction of sulphate to sulphide during AD could,
88 however, negatively impact methane production due to (1) competition between sulphate
89 reducing bacteria and methanogens for “reducing power”, (2) direct inhibition of methanogens,
90 due to H_2S toxicity, and (3) reduced trace metal bioavailability, due to precipitation with
91 sulphides (Karhadkar et al. 1987, Paulo et al. 2015). Hence, accurate control of this “unwanted”
92 sulphate reduction process, by monitoring the ingoing sulphate concentration and H_2S content

93 in the biogas, is essential to achieve a long-term stable integrated process of methane production
94 and phosphorus release.

95 The key objective of this study was to obtain integrated energy recovery, through the production
96 of biogas, and phosphate release from high-rate P-rich activated sludge during AD without the
97 need for additional chemicals. Sulphate reduction by sulphate reducing bacteria, which is
98 commonly considered an “unwanted process”, was carefully steered during AD operation to
99 maximise the release of phosphate into the liquid phase, whilst limiting the impact of the
100 sulphate reduction process on methane production.

101

102 2. Material and methods

103 2.1. Inoculum and feedstock

104 The high-rate activated sludge (A-sludge) that was used as feedstock during operation of the
105 digesters was obtained as a single batch from the A-stage of the wastewater treatment plant
106 Nieuwveer, Breda, the Netherlands (Table 1). The A-sludge was stored at 4°C until use. This
107 wastewater treatment plant was operated at a short sludge retention time (< 2 days) to maximise
108 the recovery of organics, according to the Adsorptions-Belebungsverfahren or AB-system
109 principles (Boehnke et al. 1997, Meerburg et al. 2016). The inoculum for the anaerobic digesters
110 was obtained from the sludge digesters at the full-scale wastewater treatment plant the
111 Ossemeersen, Ghent, Belgium (Table 2).

112

113 2.2. Experimental design and operation

114 Six Schott bottles with a total volume of 1 L and a working volume of 800 mL were operated
115 as lab-scale anaerobic digesters. The bottles were sealed with air-tight rubber stoppers and
116 connected to a water displacement system *via* gas-tight PVC tubing to monitor biogas
117 production (Figure S1). The liquid in this system was kept at a pH < 4.3 to avoid CO₂ in the
118 biogas from dissolving. A Laboport[®] vacuum pump (KNF Group International, Aartselaar,
119 Belgium) and glass sampling tube of 250 mL (Glasgerätebau Ochs, Lenglern, Germany) were
120 used to collect samples for biogas composition analysis. The initial inoculum biomass
121 concentration in each digester was fixed at 10 g VSS L⁻¹ (volatile suspended solids) by diluting
122 the inoculum with tap water. The digesters were operated in a continuous stirred tank reactor
123 mode with manual mixing, thus, the solids and hydraulic retention times were identical.
124 Mesophilic conditions were maintained by operating the digesters in a temperature-controlled
125 room at 34 ± 1°C. Digestate removal and feeding was carried out manually in fed-batch mode
126 three times per week.

127 A start-up period of 14 days was implemented during which a sludge retention time of 40 days
128 and an average organic loading rate of $1 \text{ g COD L}^{-1} \text{ d}^{-1}$ (chemical oxygen demand) were applied
129 to allow adaptation of the microbial community in the inoculum to the new feedstock. From
130 day 15 till day 119 (end of the experiment), a hydraulic retention time of 20 days and an average
131 organic loading rate of $2 \text{ g COD L}^{-1} \text{ d}^{-1}$ were used. Until day 41, all six reactors were operated
132 under identical conditions. From day 42 on, three biological replicates were subjected to
133 sulphate addition at a predefined fixed sulphur to phosphorus molar ratio S:P of 2 by adding
134 Na_2SO_4 to the feed (Sulphate digester). The other three biological reactors served as control
135 digester with no sulphate addition (Control digester).
136 Biogas production and composition were monitored three times per week, together with the
137 digester pH. Biogas production values were reported at standard temperature (273 K) and
138 pressure (101325 Pa) conditions (STP). The sulphate, phosphate, sodium, total ammonium and
139 volatile fatty acids (VFA) concentrations were measured on a weekly basis. The free ammonia
140 (NH_3) concentration was calculated based on the pH and total ammonia concentration
141 (Anthonisen et al. 1976). The overall salinity in the digesters was estimated through a weekly
142 measurement of the conductivity. Samples for microbial community analysis were taken on day
143 0 (inoculum and A-sludge), and day 42, 82 and 119 from each digester, and stored at -20°C
144 until DNA extraction was performed.

145

146 2.3. Microbial community analysis

147 The frozen samples were subjected directly to DNA extraction with the ZymoBIOMICS™
148 DNA Miniprep Kit (Zymo Research, Irvine, CA, USA), using a PowerLyzer® 24 Bench Top
149 Bead-Based Homogenizer (MO BIO Laboratories, Inc, Carlsbad, CA, USA), and following the
150 manufacturer's instructions. The quality of the DNA extracts was validated with agarose gel
151 electrophoresis and through PCR analysis using the universal bacterial primers 341F (5'-

152 CCTACGGGNGGCWGCAG) and 785Rmod (5'- GACTACHVGGGTATCTAAKCC) that
153 target the V3-V4 region of the 16S rRNA gene (Klindworth et al. 2013), following the protocol
154 of Boon et al. (2002). The samples were sent to BaseClear B.V., Leiden, The Netherlands, for
155 Illumina amplicon sequencing of the V3-V4 region of the 16S rRNA gene of the bacterial
156 community on the MiSeq platform with V3 chemistry. The amplicon sequencing and data
157 processing are described in detail in SI (S2). Real-time PCR analysis was carried out to quantify
158 total bacteria, the methanogenic orders Methanobacteriales and Methanomicrobiales, and the
159 methanogenic families Methanosaetaceae and Methanosarcinaceae, as described in SI (S3).

160

161 2.4. Statistical analysis

162 A table with the relative abundances of the different bacterial OTUs (operational taxonomic
163 units), together with their taxonomic assignment (Supplementary file 2) was generated
164 following the amplicon data processing. All statistical analysis were carried out in R Studio
165 version 3.3.1 (<http://www.r-project.org>) (R Development Core Team 2013). First, a repeated
166 measures analysis of variance (ANOVA, *aov* function) was used to validate that the biological
167 replicates showed no significant ($P < 0.05$) differences in bacterial community composition.
168 Next, the different samples were rescaled *via* to the “common-scale” approach (McMurdie and
169 Holmes 2014) by means of which the proportions of all OTUs were taken, multiplied with the
170 minimum sample size, and rounded to the nearest integer. Sampling depth of the different
171 samples was evaluated through rarefaction curves (Figure S2) (Hurlbert 1971, Sanders 1968).
172 The packages *vegan* (Oksanen et al. 2016) and *phyloseq* (McMurdie and Holmes 2013) were
173 used for in-depth microbial community analysis.

174 A heatmap was created on the Phylum level (1% cut-off) with the *pheatmap* function (*pheatmap*
175 package), and biological replicates were collated according to the method described by
176 Connelly et al. (2017). The order-based Hill's numbers (Hill 1973) were used to estimate

177 differences in α -diversity between the different digesters. These Hill's numbers represent
178 richness (number of OTUs, H_0), the exponential of the Shannon diversity index (H_1) and the
179 Inverse Simpson index (H_2). The non-metric multidimensional scaling (NMDS) plots, based on
180 the bacterial amplicon or methanogenic real-time PCR data, were constructed using the Bray-
181 Curtis (Bray and Curtis 1957), Chao (Chao 1984), Jaccard, Kulczynski (Faith et al. 1987), and
182 Mountford (Wolda 1981) distance measures. The OTUs with a significant difference ($P < 0.05$)
183 in relative abundance between the Sulphate and Control digester were determined with the
184 *DESeqDataSetFromMatrix* function from the DESeq2 package (Love et al. 2014)

185

186 2.5. Analytical techniques

187 Total solids (TS), total suspended solids (TSS), volatile suspended solids (VSS), volatile solids
188 (VS), Kjeldahl nitrogen (TKN) and COD were measured according to Standard Methods
189 (Greenberg et al. 1992). The total ammonium and sodium concentrations were measured on a
190 761 Compact Ion Chromatograph (Metrohm, Herisau, Switzerland), which was equipped with
191 a Metrosep C6-250/4.0 main column, a Metrosep C4 Guard/4.0 guard column and a
192 conductivity detector. The eluent consisted of 1.7 mM HNO_3 and 1.7 mM dipicolinic acid.
193 Samples were centrifuged at 3000g for 3 min with a Labofuge 400 Heraeus centrifuge (Thermo
194 Fisher Scientific Inc, Merelbeke, Belgium), filtered over a 0.45 μm filter (type PA-45/25,
195 Macherey-Nagel, Germany), and diluted with Milli-Q water to reach the desired concentration
196 range for quantification between 1 and 100 mg L^{-1} . The phosphate and sulphate concentrations
197 were measured on a 930 Compact Ion Chromatograph Flex Deg (Metrohm, Herisau,
198 Switzerland), with a Metrosep A supp 5 guard, A supp 5 150/4.0 main column and a
199 conductivity detector. Sample preparation was identical to the sodium and ammonium
200 concentrations, as well as the concentration range for quantification. The pH and conductivity
201 were measured with a C532 pH and C833 conductivity meter (Consort, Turnhout, Belgium),

202 respectively. The biogas composition was measured with a Compact Gas Chromatograph
203 (Global Analyser Solutions, Breda, The Netherlands), while the different VFA (C2-C8) were
204 measured with a GC-2014 Gas Chromatograph (Shimadzu®, The Netherlands), as described in
205 SI (S5). The total phosphorus and iron were analysed *via* Inductive Coupled Plasma Optical
206 Emission Spectrometry – VARIAN Vista MPX, following destruction in a CEM Mars 6
207 Microwave Digestion System (CEM Corporation, Matthews, NC, USA).

208

209 2.6. Data submission

210 The raw fastq files that served as a basis for the bacterial community analysis were deposited
211 in the National Center for Biotechnology Information (NCBI) database (Accession number
212 SRP185611).

213

214 3. Results

215 3.1. Digester performance

216 An efficient start-up was obtained for all six digesters, with a steady increase in biogas
217 production, which reached a plateau around day 25 (Figure 1), and no residual VFA (Figure
218 2a). From day 26 until day 42 (before sulphate addition was initiated in the Sulphate digester),
219 a stable average methane production rate of $610 \pm 60 \text{ mL CH}_4 \text{ L}^{-1} \text{ d}^{-1}$ was obtained over the six
220 digesters. This corresponded with a COD conversion efficiency of the A-sludge to CH_4 of 86.9
221 $\pm 8.6 \%$, indicating an efficient AD process.

222 The Control digester continued to show stable biogas production from day 43 till day 119 (end
223 of the experiment), with an average methane production rate of $628 \pm 103 \text{ mL CH}_4 \text{ L}^{-1} \text{ d}^{-1}$,
224 which corresponded to a COD conversion efficiency of $89.5 \pm 14.6 \%$ (Figure 1). Residual VFA
225 concentrations did not exceed 1.5 g COD L^{-1} (Figure 2a), which corresponded to a maximum
226 loss in COD via the effluent of $3.8 \pm 1.3 \%$ on day 91. The main VFA fractions in the Control
227 digester from day 43 on were acetate ($70.7 \pm 9.5 \%$) and propionate ($14.8 \pm 11.8 \%$). The pH
228 remained stable throughout the entire process, with an average value of 7.29 ± 0.10 from day
229 43 till day 119 (Figure 2b). Total salinity, as measured *via* conductivity, did not exceed $11.9 \pm$
230 0.1 mS cm^{-1} (Figure S3a), and the sodium concentration remained below $0.3 \text{ g Na}^+ \text{ L}^{-1}$ (Figure
231 S3b). The total ammonium concentration, with a maximum value of $1.34 \pm 0.02 \text{ g N L}^{-1}$ on day
232 77 (Figure S3c), and free ammonia concentration, with a maximum value of $32 \pm 2 \text{ mg N L}^{-1}$
233 on day 77 (Figure S3d), did not reach potentially inhibitory values.

234 The addition of sulphate to the feed initially did not negatively impact methane production, as
235 an average methane production rate of $613 \pm 114 \text{ mL CH}_4 \text{ L}^{-1} \text{ d}^{-1}$ and a corresponding COD
236 conversion efficiency of $87.4 \pm 16.3 \%$ were obtained between day 43 and 61 (Figure 1). After
237 day 61, methane production rate slowly, but steadily decreased to reach a minimum value of
238 $210 \text{ mL CH}_4 \text{ L}^{-1} \text{ d}^{-1}$ on day 77. From day 77 on, methane production rate again slowly increased

239 to reach a new steady state from day 98 on, with an average methane production rate of $445 \pm$
240 $53 \text{ mL CH}_4 \text{ L}^{-1} \text{ d}^{-1}$ and a corresponding COD conversion efficiency of $63.5 \pm 7.6 \%$ (Figure 1).
241 This decreasing trend in methane production and subsequent increase towards a new steady
242 state, though at lower methane production rates, is reflected in the VFA concentration profile.
243 The total VFA concentration increased from $0.8 \pm 0.9 \text{ g COD L}^{-1}$ on day 62 to $7.5 \pm 1.4 \text{ g COD}$
244 L^{-1} on day 84, but decreased again to $3.6 \pm 2.1 \text{ g COD L}^{-1}$ on day 119 (Figure 2a). This increase
245 in VFA concentration was reflected in a decrease of the pH, though a minimum value of only
246 6.95 ± 0.13 was reached on day 84 (Figure 2b), which is still within the optimal range for stable
247 AD. Total salinity was slightly higher, related to the addition of Na_2SO_4 , with a maximum
248 conductivity value of $15.1 \pm 0.2 \text{ mS cm}^{-1}$ on day 70 and maximum sodium concentration of 2.1
249 $\text{g Na}^+ \text{ L}^{-1}$ on day 112 (Figure S3), but this was insufficient to cause direct AD process failure.
250 In addition, neither total ammonium concentration, with a maximum value of $1.27 \pm 0.01 \text{ g N}$
251 L^{-1} on day 62, nor free ammonia concentration, with a maximum value of $40 \pm 3 \text{ g N L}^{-1}$ on day
252 62, reached potentially inhibitory concentrations.

253

254 3.2. Phosphate release

255 The key objective of this research was to evaluate to which extent sulphate reduction by
256 sulphate reducing bacteria could assist the release of phosphate from A-sludge, related to the
257 lower solubility product of multivalent cations with sulphides than with phosphates. In the
258 Control digester, the phosphate concentration in the liquid phase slowly increased during the
259 start-up from $34 \pm 8 \text{ mg PO}_4^{3-} \text{ L}^{-1}$ on day 7 to an average value of $256 \pm 38 \text{ mg PO}_4^{3-} \text{ L}^{-1}$ between
260 day 56 and 119 (end of the experiment) (Figure 3a). Based on the total P-content of the A-
261 sludge feedstock (Table 1), this corresponded with an average release of $12.9 \pm 2.0 \%$ of total
262 P in the liquid phase as phosphate. As no sulphate was added to the Control digester, the residual
263 sulphate concentration in the liquid phase remained below $25 \text{ mg SO}_4^{2-} \text{ L}^{-1}$, except for day 0

264 (Figure 3b). No H₂S could be detected in the biogas of the Control digesters, except for one
265 replicate on day 98 (0.14% H₂S) and day 117 (0.31 % H₂S).
266 Following the addition of Na₂SO₄ in the Sulphate digester on day 42, the phosphate
267 concentration in the liquid phase increased rapidly from only 106 ± 29 mg PO₄³⁻ L⁻¹ on day 42
268 to 1120 ± 280 mg PO₄³⁻ L⁻¹ on day 56 (Figure 3a). An average phosphate concentration of 1160
269 ± 130 mg PO₄³⁻ L⁻¹ was maintained between day 56 and day 119. This corresponded with an
270 average release of 58.7 ± 12.9 % of total P into the liquid phase as phosphate, which is a factor
271 4.5 higher than the Control digester. The addition of sulphate, however, also resulted in an
272 increased residual sulphate concentration in the liquid phase. A first initial peak of 298 ± 43 mg
273 SO₄²⁻ L⁻¹ could be observed immediately following the first sulphate addition on day 49 (Figure
274 3b). A second lower peak of 150 ± 97 mg SO₄²⁻ L⁻¹ was detected on day 84 after which the
275 sulphate concentration in the liquid phase decreased to similar values as the Control digester.
276 In contrast to the Control digester, H₂S was detected in the biogas at multiple time points
277 between day 89 and 119, with values up to 0.50 % H₂S in the biogas, which corresponded with
278 maximum 8.2 % of the sulphur added to the ingoing feedstock.

279

280 3.3. Microbial community analysis

281 3.3.1. Bacterial community

282 Amplicon sequencing of the bacterial community yielded an average of 17,570 ± 6,505 reads
283 and 1,431 ± 400 OTUs per sample (including singletons). Following removal of singletons and
284 normalisation according to the common-scale approach, this was reduced to an average of 9,631
285 ± 183 reads and 557 ± 115 OTUs per sample. No significant differences (repeated measures
286 ANOVA, *P* < 0.0001) in bacterial community composition were detected between the
287 biological replicates.

288 The bacterial community consisted mainly of the Bacteroidetes (24.4 ± 5.6 %), Firmicutes (26.4

289 ± 6.5 %), Proteobacteria (13.8 ± 6.7 %) and Chloroflexi (9.1 ± 6.5 %) phyla, averaged over all
290 samples, excluding the feedstock A-sludge (Figure 4). The A-sludge was mainly comprised of
291 Proteobacteria (62.7 %) and Firmicutes (26.6 %) phyla. A clear increase in the Proteobacteria
292 phylum could be observed in the Sulphate digester, reaching 17.6 ± 1.6 % on day 82 and 19.5
293 ± 3.1 % on day 119 relative abundance, in contrast to 7.5 ± 0.7 % on day 82 and 8.6 ± 0.9 %
294 on day 119 relative abundance in the Control digester (Figure 4). In total 222 OTUs (0.9 % of
295 all OTUs), considering all time points, showed a significant difference
296 (DESeqDataSetFromMatrix, $P < 0.05$) in relative abundance between the Sulphate and Control
297 digester. The difference in the Proteobacteria phylum between the Sulphate and Control
298 digester mainly related to OTU00025 (*Pseudomonas*, $P < 0.0001$) and OTU00057 (*Rhodoferrax*,
299 $P < 0.0001$) (Table 3). The sulphate reducers OTU00198 (*Desulfovibrio*, $P < 0.0001$),
300 OTU00219 (*Desulfobulbus*, $P < 0.0001$) and OTU00393 (*Desulfomicrobium*, $P < 0.0001$) also
301 showed a significantly higher relative abundance in the Sulphate than in the Control digester
302 (Table 3).

303 The α -diversity analysis on the different levels of diversity (H_0 , H_1 and H_2) did not reveal clear
304 differences between the digesters (Figure S4). In contrast, β -diversity analysis, based on the
305 Bray-Curtis distance measure, revealed clear divergence in the bacterial community
306 composition in the Sulphate digester on day 82 and 119, compared to the Control digester
307 (Figure 5a). As sulphate addition only started on day 42, no differences in community
308 composition were observed yet between the Control and Sulphate digester. This result was
309 confirmed for the Jaccard, Chao, Kulczynski, and Mountford distance measures (Figure S5).

310

311 3.3.2. Methanogenic community

312 Real-time PCR analysis of the total bacteria and the different methanogenic groups revealed an
313 overall dominance of the bacteria in absolute abundance, as the methanogens comprised only

314 0.26 ± 0.12 % of the microbial community, averaged over all samples, excluding the feedstock
315 A-sludge. Both the A-sludge and Inoculum were dominated by the Methanosaetaceae (Figure
316 6). This was also reflected in the Control digester during the entire experiment, though an
317 increase in relative abundance of the Methanomicrobiales could be observed at the cost of the
318 Methanosaetaceae on day 119. The Sulphate digester showed a similar pattern as the Control
319 digester, though on day 119, a strong increase in the relative abundance of the
320 Methanosarcinaceae could be observed, which coincided with a reduced relative abundance of
321 the Methanosaetaceae and Methanomicrobiales. The β -diversity analysis of the methanogenic
322 community, based on the Bray-Curtis distance measure, confirmed this shift in the
323 methanogenic community (Figure 5b). This shift was confirmed for the Jaccard and Kulczynski
324 distance measures (Figure S5).
325

326 4. Discussion

327 The stimulation of sulphate reducing bacteria by supplementing the A-sludge feedstock with
328 sulphate enabled the release of phosphate into the liquid phase. The methanogenesis process
329 was not strongly affected, yet, lower methane production rates and increased concentrations of
330 residual VFA were observed. The bacterial and archaeal community demonstrated a clear shift
331 in composition in response to the sulphate addition, with a clear increase in sulphate reducing
332 genera.

333

334 4.1. Sulphate reduction enables phosphate release from high-rate activated sludge

335 The addition of sulphate during the AD process resulted in an increase in the phosphate
336 concentration in the liquid phase up to a factor 4.5, compared to the Control digester. Even
337 though, on average only 58.7 ± 12.9 % of total P in the A-sludge could be released into the
338 liquid phase. As residual sulphate remained behind in the liquid phase and H₂S was measured
339 in the biogas in the Sulphate digester, this indicates that the sulphate reduction potential was
340 not fully used, especially since an S:P molar ratio of 2 was applied. This S:P ratio of 2 was
341 chosen to provide enough sulphides for phosphate release, yet, at the same time avoid too severe
342 negative effects on methanogenesis, related to sulphide formation. A similar observation was
343 made when subjecting manure to acidification either prior to or after AD, *i.e.*, only about 60 %
344 of total P could be released into the liquid phase (De Vrieze et al. 2019). Organic phosphorus,
345 such as DNA components, cannot be released into the liquid phase by the direct effects of
346 sulphate reduction, and requires pre-treatment methods, such as a free ammonia pre-treatment
347 (Xu et al. 2018) or other methods that improve sludge biodegradability (Carrère et al. 2010).
348 The effect of such a pre-treatment, however, will be limited. This is because of the already high
349 conversion efficiency of COD, and, thus, also the organic phosphorus components, to methane
350 in this study (86.9 ± 8.6 %), as also reported earlier (De Vrieze et al. 2013, De Vrieze et al.

351 2015, Ge et al. 2013, Meerburg et al. 2015). In addition, the economic feasibility of such a pre-
352 treatment strategy depends on the increased COD conversion efficiency (Ma et al. 2011), which
353 is anticipated to be limited, and phosphorus release. A strategy that could actively and
354 simultaneously benefit both the COD conversion efficiency and phosphate release, such as a
355 nitrous acid pre-treatment (Pijuan et al. 2012, Wei et al. 2018), potentially could be
356 economically feasible.

357 The addition of sulphate resulted in a so-called “inhibited steady-state”, as described previously
358 for ammonia toxicity (Nielsen and Angelidaki 2008), and salt toxicity (De Vrieze et al. 2014).
359 Such an inhibited steady-state is characterized by elevated concentrations of residual VFA, and
360 a lower, yet, steady methane production. This allows two different potential approaches for
361 further process optimisation. A first option involves accurate control of sulphate dosing, based
362 on online monitoring of residual VFA concentrations, residual sulphate concentrations, H₂S in
363 the gas phase, and/or methane production rates to sustain an optimal combined methanogenesis
364 and phosphate release process. Alternatively, sulphate addition can be increased to selectively,
365 but completely inhibit the sensitive methanogens (Karhadkar et al. 1987), thus, evolving
366 towards fermentation with the objective to directly produce VFA instead of methane, combined
367 with phosphate release. Both approaches require an alternative way of process engineering for
368 targeted resource recovery of which technical and economic aspects will determine the case-
369 specific application potential.

370

371 4.2. The microbial community response reflects a shift in response to sulphate addition

372 The accumulation of VFA and decrease in methane production in the Sulphate digester
373 coincided with a clear shift in the microbial community. The increase in relative abundance of
374 confirmed sulphate reducing bacteria in the Sulphate digester, relative to the Control digester
375 is to be expected, yet, their relative abundance, except for OTU00198 in one of the biological

376 replicates on day 119, remained below 1 % of the bacterial community. One could question the
377 involvement of these sulphate reducing bacteria in the overall process, yet, given their ability
378 to reduce sulphate and complete absence in the different biological replicates of the Control
379 digester on day 82 and 119, their involvement in the sulphate reduction process is apparent. The
380 potentially important role of low-abundant OTUs in AD has been indicated frequently (Guo et
381 al. 2015, Theuerl et al. 2018, Vanwonterghem et al. 2016), and is also reflected in the present
382 study, with an overall low relative abundance of the methanogens, which are nonetheless
383 essential in the AD process.

384 The methanogenic community in the Sulphate digester showed a clear shift towards an
385 increased relative and absolute abundance of the Methanosarcinaceae at the expense of the
386 Methanosaetaceae, in contrast to the Control digester. The overall higher tolerance of
387 *Methanosarcina* sp., compared to *Methanosaeta* sp., to different stressors in AD (Conklin et al.
388 2006, De Vrieze et al. 2012), explains this shift, in response to the formation of sulphides. A
389 similar shift has been observed in other studies, in response to multiple stressors, although the
390 methane production pathway by the *Methanosarcina* sp., *i.e.*, either acetoclastic or
391 hydrogenotrophic methanogenesis, may vary (De Vrieze et al. 2012, Lins et al. 2014, Lu et al.
392 2013, McMahon et al. 2001, Poirier et al. 2016, Venkiteshwaran et al. 2016). The apparent
393 “inhibited steady state” in this study, as also observed previously, thus, seems to be a
394 consequence of the shift from a *Methanosaeta* sp. to *Methanosarcina* sp. domination, and
395 reflects their different metabolic features.

396

397 4.3. The cost of phosphate release through sulphate reduction

398 The release of phosphate into the liquid phase due to sulphide formation was up to a factor 4.5
399 higher than when no sulphate was added to the feedstock. This enables an integrated
400 valorisation of the A-sludge, with combined energy (through biogas) and nutrient (through

401 phosphate) recovery. For this purpose, the “inhibited steady-state” phase (day 89-119) of the
402 Sulphate digester ($428 \pm 53 \text{ mL CH}_4 \text{ L}^{-1} \text{ d}^{-1}$) was compared to the Control digester (611 ± 110
403 $\text{mL CH}_4 \text{ L}^{-1} \text{ d}^{-1}$) during that same period, which corresponded with a $29.9 \pm 15.3 \%$ lower
404 methane production in the Sulphate compared to the Control digester. An electrical efficiency
405 of 40% for the CHP unit (Deublein and Steinhauser 2008, Szarka et al. 2013), a higher heating
406 value of $9.95 \text{ kWh m}^{-3} \text{ CH}_4$, a 95% recovery of methane from the digester, and a hydraulic
407 retention time of 20 days were assumed. At a current electricity market price of $\text{€ } 0.10 \text{ kWh}^{-1}$
408 (De Vrieze et al. 2016), this amounts $\text{€ } 126 \text{ m}^{-3} \text{ year}^{-1}$ for the Control and $\text{€ } 89 \text{ m}^{-3} \text{ year}^{-1}$ for the
409 Sulphate digester per unit digester volume for electricity from biogas, thus, a deficit of $\text{€ } 37 \text{ m}^{-3}$
410 year^{-1} related to sulphate reduction. Projected electricity market prices of $\text{€ } 0.03 \text{ kWh}^{-1}$ by
411 2020-2025, and even down to $\text{€ } 0.01 \text{ kWh}^{-1}$ by 2030-2040 (Fraunhofer 2015, van Wijk et al.
412 2017) reduce this deficit to $\text{€ } 8 \text{ m}^{-3} \text{ year}^{-1}$ & $\text{€ } 3 \text{ m}^{-3} \text{ year}^{-1}$, respectively. Based on the market price of
413 phosphate rock of $\text{€ } 350\text{-}1200 \text{ tonne}^{-1} \text{ P}$ since 2010 (Mayer et al. 2016), this yields a potential
414 revenue of $\text{€ } 0.5 \text{ m}^{-3} \text{ year}^{-1}$ for the Control and $2.3 \text{ m}^{-3} \text{ year}^{-1}$ for the Sulphate digester at a value
415 of $\text{€ } 350 \text{ tonne}^{-1} \text{ P}$. At a value of $\text{€ } 1200 \text{ tonne}^{-1} \text{ P}$, this revenue increases to $\text{€ } 1.8 \text{ m}^{-3}$ & 7.9 m^{-3}
416 year^{-1} for the Control and Sulphate digester, respectively. This indicates that the deficit due to
417 the decrease in methane production can be at least partially covered by the revenue from
418 phosphorus recovery, but additional resource recovery strategies and/or advanced control of the
419 sulphate reduction process will be essential.

420 The integrated approach of this study that makes use of sulphate reduction during AD for
421 combined energy and phosphorus recovery faces additional challenges. First, the release of
422 phosphate into the liquid phase allows subsequent recovery, yet, phosphate recovery
423 technologies, such as struvite precipitation, result in additional costs and product quality issues.
424 For example, phosphorus recovery from pig manure through struvite crystallisation coincides
425 with an electricity cost of $\text{€ } 0.5 \text{ tonne}^{-1} \text{ pig manure}$ (De Vrieze et al. 2019, Flotats et al. 2011).

426 Second, the increase in H₂S in the biogas requires additional desulfurization, for which both
427 physicochemical and biological techniques can be used (Abatzoglou and Boivin 2009), to a
428 recommended value < 0.025 % (Weiland 2010) before it can be sent to the CHP unit. Third, the
429 presence of residual VFA in the liquid phase, with an average value of 4.9 ± 1.4 g COD L⁻¹
430 between day 89-119, necessitates additional (aerobic) effluent polishing.

431 These challenges can also be considered opportunities. The H₂S in the biogas can be recovered
432 as elemental sulphur *via* different techniques (Pandey and Malhotra 1999). The phosphate and
433 residual VFA can be recovered in a combined electrodialysis or membrane electrolysis
434 approach (Andersen et al. 2014, De Vrieze et al. 2018). Alternatively, the AD process can be
435 shifted from methanogenesis to direct VFA production, thus, avoiding biogas desulfurization
436 and focusing on the liquid phase. Hence, an integrated approach that combines the release and
437 recovery of phosphorus with other resource recovery strategies for the valorisation of organic
438 waste and side streams could find its way towards future full-scale applications.

439

440 **5. Conclusions**

441 The exploitation of sulphate reduction for the release of phosphorus from energy-rich sludge
442 increased the phosphate concentration in the liquid phase with a factor 4.5. The sulphate
443 reduction process pushed the anaerobic digestion process to an inhibited steady state, as
444 reflected in both operational and microbial parameters, with reduced, yet, stable methane
445 production rates. Phosphate can be recovered in an economically feasible way, but only in
446 combination with energy, organics or other resources for integrated valorisation of energy-rich
447 sludge.

448

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455

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667

668 **Tables:**

669 **Table 1** Main characteristics of the high-rate energy-rich A-sludge (n=3). TS = total solids, VS
670 = volatile solids, COD = chemical oxygen demand, VFA = volatile fatty acids, TAN = total
671 ammonia nitrogen, TKN = Kjeldahl nitrogen, FW = fresh weight.

Parameter	Unit	A-sludge
pH	-	5.64 ± 0.01
TS	g TS kg ⁻¹ FW	35.4 ± 1.6
VS	g VS kg ⁻¹ FW	28.4 ± 1.3
Total COD	g COD kg ⁻¹ FW	40.1 ± 2.1
Conductivity	mS cm ⁻¹	2.04 ± 0.02
Total VFA	mg COD kg ⁻¹ FW	0 ± 0
TAN	mg N kg ⁻¹ FW	28 ± 0
TKN	mg N kg ⁻¹ FW	1571 ± 225
Total P	mg P kg ⁻¹ FW	659 ± 28
COD:N ratio	-	25.5 ± 3.9
COD:P ratio	-	60.8 ± 4.1
TS:VS ratio	-	1.25 ± 0.08
COD:VS ratio	-	1.41 ± 0.10

672

673 **Table 2** Main characteristics of the inoculum sludge (n=3). TSS = total suspended solids, VSS
674 = volatile suspended solids, COD = chemical oxygen demand, TAN = total ammonia nitrogen,
675 VFA = volatile fatty acids, FA = free ammonia nitrogen.

Parameter	Unit	Inoculum
pH	-	7.48 ± 0.07
TSS	g TSS L ⁻¹	44.6 ± 0.07
VSS	g VSS L ⁻¹	23.7 ± 0.11
Conductivity	mS cm ⁻¹	10.51 ± 0.02
Total VFA	mg COD L ⁻¹	0 ± 0
TAN	mg N L ⁻¹	572 ± 26
FA ¹	mg N L ⁻¹	18 ± 1

676 ¹The free ammonia (FA) content was calculated based on the TAN concentration, pH and
677 temperature in the full-scale installation (Anthonisen et al. 1976).

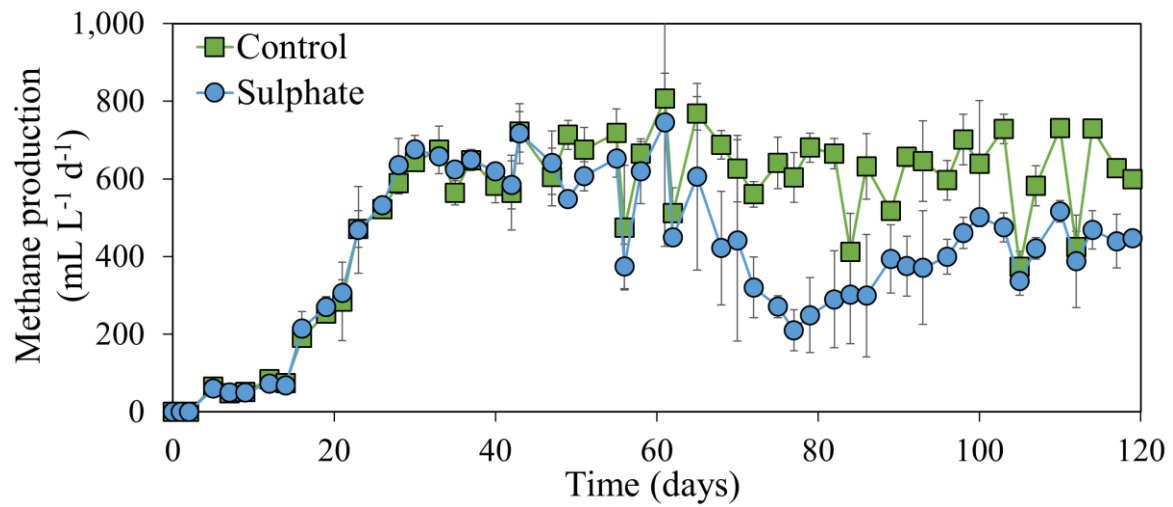
678

679 **Table 3** Overview of the key OTUs with their relative abundance in the bacterial community
680 that show a significantly different relative abundance (DESeqDataSetFromMatrix, $P < 0.0001$)
681 between the Sulphate and Control digesters.

OTU	Genus	Relative abundance (%)	
		Sulphate	Control
Otu00025	<i>Pseudomonas</i>	3.5 ± 4.3	0.0 ± 0.0
Otu00057	<i>Rhodoferrax</i>	2.0 ± 0.6	0.2 ± 0.2
Otu00198	<i>Desulfovibrio</i>	0.8 ± 0.6	0.0 ± 0.0
Otu00219	<i>Desulfobulbus</i>	0.1 ± 0.1	0.0 ± 0.0
Otu00393	<i>Desulfomicrobium</i>	0.3 ± 0.2	0.0 ± 0.0

682

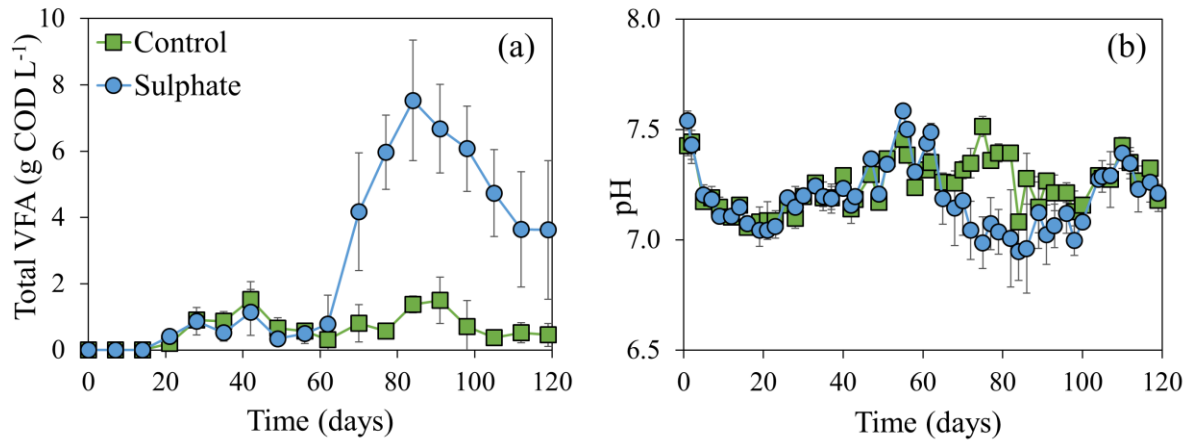
683 **Figures:**



684

685 **Figure 1** Methane production in function of time in the Control and Sulphate digester. Average
686 values of the biological replicates (n=3) are presented, and the error bars represent standard
687 deviations.

688



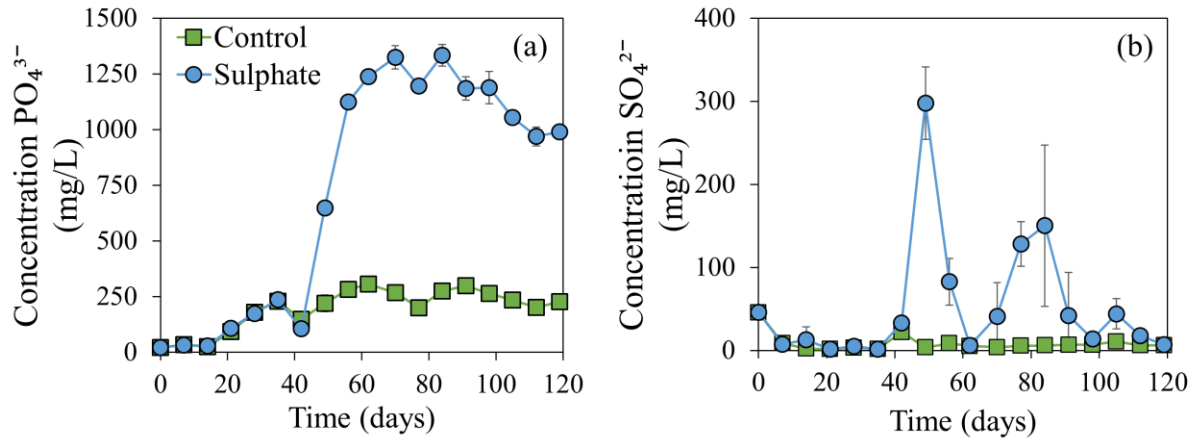
689

690 **Figure 2** Total volatile fatty acid (VFA) concentration (a) and pH (b) in function of time in the

691 Control and Sulphate digester. Average values of the biological replicates (n=3) are presented,

692 and the error bars represent standard deviations.

693



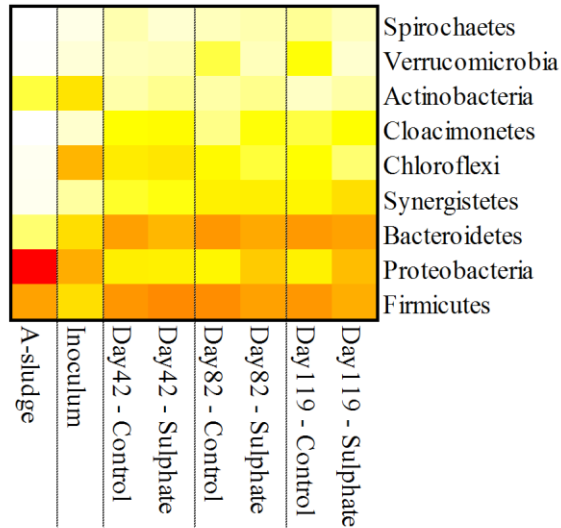
694

695 **Figure 3** Phosphate (a) and sulphate (b) concentration in function of time in the Control and

696 Sulphate digester. Average values of the biological replicates (n=3) are presented, and the error

697 bars represent standard deviations.

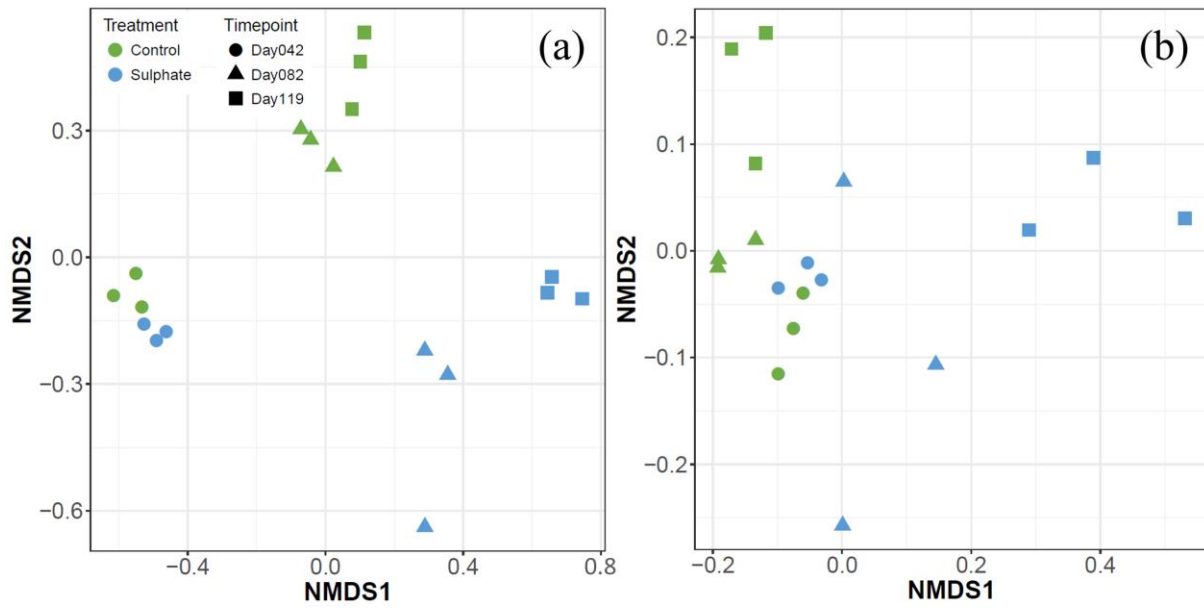
698



699

700 **Figure 4** Heatmap showing the relative abundance of the bacterial community at the phylum
701 level in the A-sludge feedstock, the inoculum and on day 42, 82 and 119 for both digesters.
702 Weighted average values of the biological replicates are presented. The colour scale ranges
703 from 0 (white) to 60% (red) relative abundance.

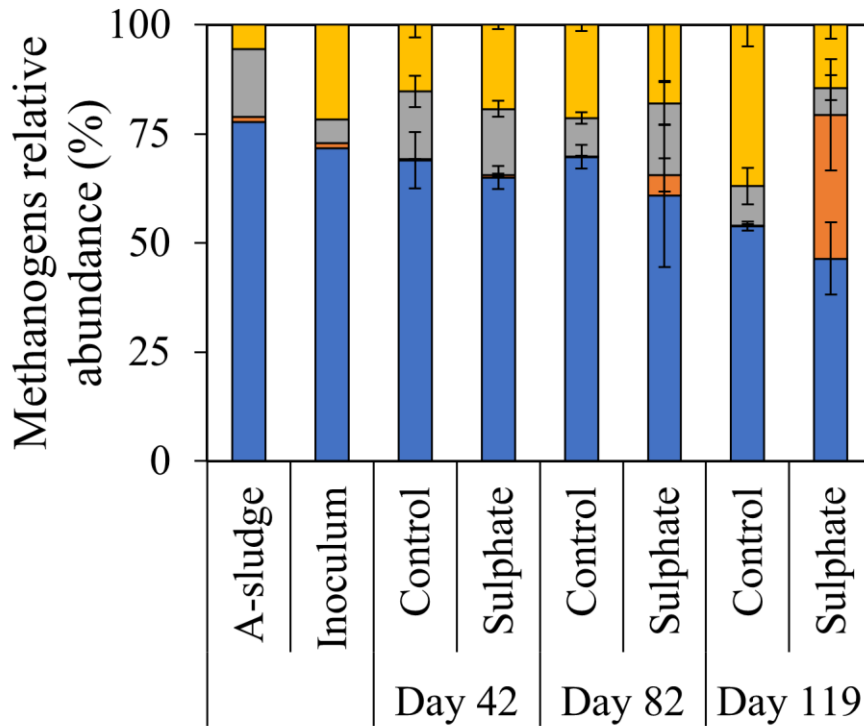
704



705

706 **Figure 5** Non-metric multidimensional distance scaling (NMDS) analysis of the Bray-Curtis
707 distance measure of the bacterial (a), based on amplicon sequencing data at OTU level (stress
708 = 0.059), and methanogenic (b) community (stress = 0.072), based on real-time PCR data.
709 Different colours and symbols are used for different digesters and timepoints, respectively.

710



711

712 **Figure 6** Relative abundance (%) of the Methanosaetaceae (blue, ■), Methanosarcinaceae
713 (orange, ■), Methanobacteriales (grey, ■) and Methanomicrobiales (yellow, ■) in the
714 methanogenic community of the A-sludge feedstock, the inoculum and on day 42, 82 and 119
715 for both digesters. Average values of the biological replicates (n=3) are presented, and the error
716 bars represent standard deviations.