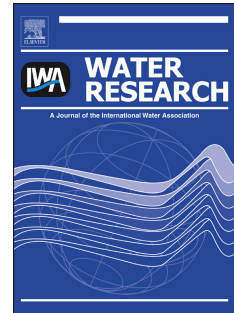


Accepted Manuscript

Title: The cost of a large-scale hollow fibre MBR

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PII: S0043-1354(10)00441-0

DOI: [10.1016/j.watres.2010.06.054](https://doi.org/10.1016/j.watres.2010.06.054)

Reference: WR 8109

To appear in: *Water Research*

Received Date: 31 March 2010

Revised Date: 18 June 2010

Accepted Date: 22 June 2010

Please cite this article as: Verrecht, B., Maere, T., Nopens, I., Brepols, C., Judd, S. The cost of a large-scale hollow fibre MBR, *Water Research* (2010), doi: 10.1016/j.watres.2010.06.054

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1 **The cost of a large-scale hollow fibre MBR**

2

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11

12

13 **Abstract**

14 A cost sensitivity analysis was carried out for a full scale hollow fibre membrane
15 bioreactor to quantify the effect of design choices and operational parameters on
16 cost. Different options were subjected to a long-term dynamic influent profile and
17 evaluated using ASM1 for effluent quality, aeration requirements and sludge
18 production. The results were used to calculate a net present value (NPV),
19 incorporating both capital expenditure (capex), based on costs obtained from
20 equipment manufacturers and full scale plants, and operating expenditure (opex),
21 accounting for energy demand, sludge production and chemical cleaning costs.

22

23 Results show that the amount of contingency built in to cope with changes in
24 feedwater flow has a large impact on NPV. Deviation from a constant daily flow
25 increases NPV as mean plant utilisation decreases. Conversely, adding a buffer tank
26 reduces NPV, since less membrane surface is required when average plant
27 utilisation increases. Membrane cost and lifetime is decisive in determining NPV: an
28 increased membrane replacement interval from 5 to 10 years reduces NPV by 19%.
29 Operation at higher SRT increases the NPV, since the reduced costs for sludge
30 treatment are offset by correspondingly higher aeration costs at higher MLSS levels,
31 though the analysis is very sensitive to sludge treatment costs. A higher sustainable
32 flux demands greater membrane aeration, but the subsequent opex increase is offset
33 by the reduced membrane area and the corresponding lower capex.

34

35 **Keywords** *Membrane bioreactor, cost sensitivity, life cycle, biokinetics,*
36 *aeration*

37 **List of symbols and abbreviations**

38	ASM1	Activated sludge model no. 1
39	BSM1 LT	Long term benchmark simulation model no. 1
40	Capex	Capital expenditures, in Euro
41	CAS	Conventional activated sludge plant
42	CIP	Cleaning in place
43	COP	Cleaning out of place
44	d_f	Hollow fibre outside diameter, in m
45	EQI	Effluent quality index, in kg PU.d ⁻¹
46	HF	Hollow fibre
47	HRT	Hydraulic retention time, in h
48	i	Discount rate, in %
49	J_0	Intercept of the J vs. U curve, in l.m ⁻² .h ⁻¹
50	J_x	Flux, in l.m ⁻² .h ⁻¹
51	$L_{membrane}$	Length of the membrane module, in m
52	L_{tank}	Tank length, in m
53	m	Slope of the J vs U curve
54	MBR	Membrane bioreactor
55	NPV	Net present value, in Euro
56	Opex	Operational expenditures, in Euro.year ⁻¹
57	PE_{sludge}	Pumping energy required per unit of sludge, in kWh.m ⁻³
58	P_{sludge}	Power required for sludge pumping, in kW
59	PU_x	Pollution unit for effluent component x, in kg.d ⁻¹
60	Q_E	Effluent flow, in m ³ .d ⁻¹
61	Q_I	Influent flow, in m ³ .d ⁻¹
62	Q_{MR}	Membrane recirculation flow, in m ³ .d ⁻¹
63	Q_{NR}	Nitrate recirculation flow, in m ³ .d ⁻¹
64	Q_W	Wastage flow, in m ³ .d ⁻¹
65	SAD_m	Specific aeration demand per unit of membrane area, in Nm ³ .m ⁻² .h ⁻¹
66	SAD_p	Specific aeration demand per unit of permeate, dimensionless
67	SRT	Solids retention time, in d
68	U	In-module air upflow gas velocity, in m.s ⁻¹
69	W_{tank}	Tank width, in m
70	Δh	Head loss, in m
71	α	Clean-to-process water correction factor for oxygen transfer
72	β_x	Weighting factor for effluent component x, dimensionless
73	ξ_B	Blower efficiency, dimensionless
74	ξ_p	Pump efficiency, dimensionless
75	ρ_{sludge}	Sludge density, in kg.m ⁻³
76	φ	Module packing density, in m ⁻¹

77 1. Introduction

78 Over the last two decades, implementation of membrane bioreactors (MBRs) has
79 increased due to their superior effluent quality and low plant footprint (Judd, 2008).
80 However, they are still viewed as a high-cost option, both with regards to capital and
81 operating expenditure (capex and opex), mainly due to membrane installation and
82 replacement costs and higher energy demand compared to conventional activated
83 sludge systems. However, quantification of such impacts is constrained by availability
84 of credible data.

85

86 An overview of literature investment cost data (McAdam and Judd, 2006, Fig. 1) over
87 a range of reported plant sizes reveals costs to increase exponentially with
88 decreasing plant size, and that a large variation in required capex arises according to
89 assumptions made and costs included. DeCarolis *et al.* (2004) provided a
90 comprehensive overview of costing data in terms of capex and opex, both for the
91 MBR system alone (based on quotes from four leading suppliers), and for the
92 complete installation (based on preliminary plant design and assumptions about the
93 location-specific contribution of land costs, contractor overheads, engineering, legal
94 costs, etc). Côté *et al.* (2004) compared capex and opex of an MBR to a conventional
95 activated sludge (CAS) system with tertiary filtration for effluent reuse purposes,
96 demonstrating an integrated MBR to be less expensive than a combination of CAS
97 and tertiary filtration - a conclusion subsequently corroborated by Brepols *et al.*
98 (2009) for German wastewater plants. The latter authors showed energy demand to
99 increase for plants with significant in-built contingency, since the average plant
100 utilisation is low. This has recently led Maurer (2009) to introduce the specific net
101 value (SNPV), which takes into account the average plant utilisation over its lifetime
102 and so reflects the cost per service unit.

103

104 Notwithstanding the above, no in-depth analysis has been produced quantifying the
105 impact of key design and operating parameters on both capex and opex over the
106 lifetime of an installation. This paper aims to determine both absolute values of capex
107 and opex and their sensitivity to various influencing parameters such as contingency
108 (to provide robustness to changes in feedwater flow and composition), membrane
109 replacement, net flux, and hydraulic and solids residence time (HRT and SRT). The
110 approach taken is to evaluate the impact of representative dynamic flow and load
111 conditions using ASM1 (Henze *et al.*, 2000) on effluent quality, sludge production and
112 aeration demand, based on various MBR process designs. Dynamic simulation

113 results can then be used as input for specific cost models for both capex and opex,
114 generated using representative heuristic and empirical available cost data. Opex for
115 energy demand (Maere *et al.*, 2009), added to sludge treatment and disposal and
116 chemical cleaning costs, can then be combined with capex to produce the NPV. This
117 then allows the impact of design and operation parameter selection to be quantified.

118 ***Insert Figure 1: Specific investment vs. installed plant capacity, based on literature***
119 ***data (adapted from McAdam and Judd, 2006)***

120

121 **2. Materials and methods**

122 **2.1 Long term influent**

123 The 87 week long BSM1 LT dynamic influent file (Gernaey *et al.*, 2006) was used to
124 evaluate the different plant designs. It includes all phenomena typically observed in a
125 year of full-scale WWTP influent data. Average influent flow (Q_{in}) was $20,851 \text{ m}^3 \cdot \text{d}^{-1}$,
126 while the maximum instantaneous flow was $59,580 \text{ m}^3 \cdot \text{d}^{-1}$. The first 35 weeks of
127 influent data were used to initialise the models; the remaining influent data covering a
128 period of one year (52 weeks) were used for evaluation.

129 **2.2 Biological process model**

130 Figure 2 depicts the generic nitrifying-denitrifying plant upon which all further design
131 options were based. The ASM1 biokinetic model was selected to study the impact of
132 design and operational parameters on biological performance. Since no consensus
133 exists on updating biokinetic values for an MBR, the default ASM1 biokinetic
134 parameter values, as reported in Henze *et al.* (2000), were used throughout.
135 Simulations were performed using the WEST[®] simulation and modelling platform
136 (Vanhooren *et al.*, 2003).

137

138 ***Insert Figure 2: Schematic overview of the generic nitrifying/denitrifying MBR design***

139

140 Biological tank volumes were determined by a required minimum HRT at average
141 influent flow conditions of 8 hours, or a minimum HRT at maximum flow conditions of
142 4 hours, whichever was the largest, and the default SRT value was 25 days. These
143 design conditions are within reported trends for large MBR in Europe (Itokawa *et al.*,
144 2008). The anoxic fraction represented 40% of tank volumes. Sludge recirculation
145 was carried out from the membrane tank to the aerobic tanks was taken as four times

146 the average feed flow: $Q_{MR} = 83,404 \text{ m}^3 \cdot \text{d}^{-1}$. Internal recirculation from the aerobic
147 tanks to the anoxic tanks was three times the feed flow $Q_{int} = 62,553 \text{ m}^3 \cdot \text{d}^{-1}$.

148

149 The membrane tank volume, included in the total aerobic volume, was calculated
150 based on a conservative packing density of 45 m^2 membrane area per m^3 of tank
151 volume, which is at the lower end of values reported (Judd and Judd, 2010). The
152 number of membrane tanks required was based on the design parameters for a large
153 scale plant (Brepols *et al.*, 2008), one membrane tank required per $10,000 \text{ m}^2$
154 membrane area, allowing sufficient flexibility in operation and cleaning.

155

156 The required buffer tank volume was dictated by:

- 157 • an assumed maximum buffer tank HRT of 2 days - based on the maximum flow
158 from the buffer tank equating to the difference between the conservative net flux
159 and the maximum sustainable flux, corresponding to 40% of plant design flow;
- 160 • the combination of plant and buffer tank required to cope with storm flows without
161 bypass.

162 Taking these constraints into account, the maximum size of the buffer tank was equal
163 to 80% of the daily design plant flow.

164

165 **2.3 Capital costs**

166 To evaluate capital investment costs, pricing information (Table 2) was obtained from
167 manufacturers or based on costs provided by end-users for similar items of
168 equipment at full scale MBR plants (Brepols, 2010). Assumptions made were as
169 follows:

170

171 **Membranes** A net design flux of $20 \text{ l} \cdot \text{m}^{-2} \cdot \text{h}^{-1}$ (LMH) was used for calculating
172 membrane area, while the maximum sustainable flux was assumed to be 40%
173 higher, i.e. 28 LMH, which can be considered conservative based on literature values
174 (Judd and Judd, 2010; Garcés *et al.*, 2007). A regime of 10 min filtration followed by
175 30 s backwashing resulted in an instantaneous flux of 22.1 LMH and maximum
176 instantaneous flux of 30.9 LMH. HF membrane costs were assumed to be $\text{€}50 \cdot \text{m}^{-2}$
177 (Brepols *et al.*, 2010).

178

179 **Tanks** Tank building costs were based on costs of $\text{€}220 \cdot \text{m}^{-3}$ tank volume (Brepols *et*
180 *al.*, 2010).

181

182 **Plant equipment** A 6mm coarse screening step followed by a 0.75 mm fine screen
183 was chosen as a representative pre-treatment for HF membranes (De Wilde *et al.*,
184 2007a). Screens were sized to treat the maximum instantaneous flow to the plant,
185 with 50% redundancy, ensuring that the whole flow could be treated by 2 sets of fine
186 and coarse screens with one set on standby.

187

188 To size the membrane blowers, SAD_m was assumed constant at $0.3 \text{ Nm}^3 \cdot \text{m}^{-2} \cdot \text{h}^{-1}$. The
189 number of blowers for membrane aeration installed was based on the number of
190 membrane tanks, with one standby blower. The biology blowers were sized based on
191 the maximum aeration demand to maintain DO at $2 \text{ mg} \cdot \text{l}^{-1}$ over the final 365 days of
192 simulation, assuming 50% standby capacity and a maximum design temperature of
193 $20 \text{ }^\circ\text{C}$.

194

195 Biomass recirculation, permeate pumps and anoxic zone mixers were sized based
196 on those typical of a large scale plant, with one standby in each case. One agitator
197 per 450 m^3 of anoxic tank volume was assumed. Costs of land, civil engineering,
198 other electrical equipment and construction were excluded, these being location
199 specific.

200

201 **2.4 Operational costs**

202 Operational costs were determined using the approach of the control strategy
203 evaluation benchmark community (Copp *et al.*, 2002), which was extended by Maere
204 *et al.* (2009) for MBR applications. The opex analysis was limited to energy demand,
205 sludge treatment and disposal, and chemical usage for membrane cleaning.

206

207 **2.4.1 Energy demand**

208 The individual contributions to energy demand are described below, and a Germany-
209 specific energy cost of $\text{€}0.0942 \cdot \text{kWh}^{-1}$ used throughout (Energyref - Christoph??).

210

211 **Aeration energy** The influence of MLSS concentration (via the α -factor) and aerator
212 type (fine and coarse bubble) on oxygen transfer was computed using the dedicated
213 aeration model of Maere *et al.* (2009), combining several literature findings (Metcalf
214 and Eddy, 2003; Henze *et al.*, 2008; Verrecht *et al.*, 2008; Krampe and Krauth, 2003;
215 Germain *et al.*, 2007; Stenstrom and Rosso, 2008).

216

217 Based on typical practically measured values for blower outlet pressure (106300 Pa;
218 for a typical aerator depth of 5 m and allowing for losses incurred in the pipework)
219 and a blower efficiency ξ_B of 0.60, a value of 0.025 kWh.Nm⁻³ air was determined for
220 the aeration energy demand, corresponding well with literature values (Verrecht *et al.*,
221 2008) and data from blower manufacturers. The average total aeration energy in
222 kWh.d⁻¹ was obtained by summing blower power consumption for both membrane
223 and biology blowers and integrating over the 365 day simulation period (Maere *et al.*,
224 2009).

225

226 **Pumping energy** Sludge pumping requirements, for internal recirculation (Q_{int} ,
227 m³.d⁻¹), membrane recirculation (Q_{MR} , m³.d⁻¹) and wastage (Q_W , m³.d⁻¹) (Insert Figure
228 2), were determined from the expression of Maere *et al.* (2009), using a power
229 requirement of 0.016 kWh.m⁻³ of sludge pumped which was calculated from
230 assuming a simple linear dependency of P_{Sludge} (Power required for sludge pumping)
231 on sludge flow and assuming a total headloss Δh of 3m and a pump efficiency ξ_p of
232 50%. To calculate additional pumping energy for permeate pumping and
233 backwashing, the expression provided by Maere *et al.* (2009) was applied.

234

235 **Mixing energy** A typical constant mixing power requirement of 8 W per m³ of anoxic
236 tank volume was used (Metcalf and Eddy, 2003), with no supplementary mechanical
237 mixing required for the aerobic, membrane and buffer tanks.

238

239 **2.4.2 Sludge production**

240 Sludge production (in kg.d⁻¹) was calculated using the expressions of Copp *et al.*
241 (2002), adapted for MBR use by Maere *et al.* (2009). Reported costs for sludge
242 handling and disposal vary from €43.tnDS⁻¹ (Rossi *et al.*, 2002), which accounts for
243 chemicals, labour, treatment and disposal, to €259.tnDS⁻¹ (Stensel and Strand,
244 2004), based on costs for collection, thickening, digestion, dewatering, reuse, but
245 excluding haulage. Sludge handling cost figures across a broad range of values were
246 thus considered.

247

248 **2.4.3 Chemical consumption**

249 A typical membrane cleaning protocol and frequency based on literature data
250 (Brepols *et al.*, 2008; Judd and Judd, 2010) was assumed to provide chemical
251 consumption data. The protocol comprised a weekly clean in place (CIP) with 500
252 ppm NaOCl and 2000 ppm citric acid, and a cleaning out of place (COP) with 1000

253 ppm NaOCl and 2000 ppm citric acid, conducted twice yearly. Representative prices
254 for bulk chemicals were obtained from chemical suppliers.

255 **2.5 Effluent quality evaluation**

256 Evaluation of effluent quality was based on the approach of Copp *et al.* (2002), which
257 quantifies the pollution load to a receiving water body in a single parameter, the
258 effluent quality index (EQI), in kg pollution units.d⁻¹ (kg PU.d⁻¹). A larger EQI thus
259 indicates worse effluent quality. The average EQI was determined through integrating
260 the expressions of Copp (2002) over the evaluation period, using the weighting
261 factors β_x as reported by Vanrolleghem *et al.* (1996).

262 **2.6 Net present value calculation**

263 The net present value was calculated for a plant lifetime of 30 years, taking into
264 account all capital and operational expenditures during the plant lifetime:

$$265 \quad NPV = \sum_{t=0}^{29} \frac{(capex)_t + (opex)_t}{(1+i)^t} \quad (1)$$

267
268 A membrane life of 10 years was assumed, corresponding to two complete
269 membrane refits during the projected plant lifetime, based on recently reported trends
270 (De Wilde *et al.*, 2007b). Long term inflation was assumed to be 3%, while a discount
271 rate i of 6% was used, comparable to values used by Côté *et al.* (2004).

272 **3. Results and discussion**

273 **3.1 Effect of contingency: changes in feedwater flow and strength**

274 **3.1.1 Hybrid plant vs. plant designed for maximum flow**

275 Table 1 shows a breakdown of costs for two extreme scenarios:

- 276 a) the MBR part of a 'hybrid' plant (i.e. an MBR parallel to a CAS plant; the MBR is
277 designed to treat a constant daily flow, while excess flow is treated by the CAS
278 plant, that is not taken into account in this analysis); and
279 b) a plant designed to cope with maximum flow conditions (peak flow = 3 x average
280 flow).
281

282
283 The results illustrate that deviating from the ideal 'hybrid' plant scenario leads to
284 severe plant under-utilization, and a resulting cost penalty manifested in a 59%
285 increased NPV value over that of the hybrid plant, despite treating the same

286 cumulative flow over the plant life. The EQI is 3.8% lower for the 'hybrid' plant, due to
287 the constant HRT of 8h, while for the plant designed for maximum flow the HRT can
288 be as low as 4h during peak flows.

289

290 ***Insert Table 1: Capex, opex and resulting NPV for an MBR treating steady state***
291 ***influent, as part of a hybrid plant, and a MBR, designed for maximum flow without***
292 ***buffer tanks.***

293

294 Figure 3 shows a breakdown of the energy demand for the same two plants. The
295 values obtained are in line with those reported for full scale plants (Garcés *et al.*,
296 2007; Brepols *et al.*, 2009). The average energy demand for the 'maximum flow'
297 plant is ~54% higher, mostly due to under-utilisation of the available membrane
298 capacity and the resulting excess aeration. This illustrates that effective control
299 strategies where membrane aeration as applied in proportion to flow conditions could
300 generate significant opex savings.

301 ***Insert Figure 3: Breakdown of energy demand for a) the MBR part of a hybrid plant***
302 ***(Average total energy demand = 0.7 kWh.m⁻³) and b) a plant designed for maximum***
303 ***flow (Average total energy demand = 1.08 kWh.m⁻³)***

304

305 The analysis shows NPV and operational efficiency of MBRs to be very susceptible
306 to the extent of built-in contingency, which is mostly determined by the changes in
307 feedwater flow such as during storm events. An example of this is the 48 MLD
308 (megalitres per day) Nordkanal plant in Germany (Brepols *et al.*, 2009), which was
309 designed to treat a peak flow that is 3-4 times higher than the average flow. The plant
310 also has 33% more membrane surface installed than required to treat the peak flow,
311 a requirement under German regulations. Consequently, mean fluxes at the plant are
312 only 8 LMH and specific energy consumption for the MBR is 0.5-1.8 kWh.m⁻³.
313 Conversely, the hybrid MBR plant at Ulu Pandan in Singapore is designed to
314 continuously treat a flow of 23 MLD, leading to very efficient operation and energy
315 consumption as low as ~0.4 kWh.m⁻³ for the MBR part of the hybrid plant, mainly due
316 to continuous improvement in membrane aeration protocols (Qin *et al.*, 2006, 2007;
317 Seah *et al.*, 2009). Thus, provided there is a constant demand for high quality effluent
318 for reuse, the hybrid plant is the most favoured option. This can be retrofitted to an
319 existing CAS, provided full effluent disinfection is not required (Lesjean *et al.*, 2009;
320 Mulder, 2008).

321

3.1.2 Buffer tank vs. extra membranes

Adding buffering capacity for flow equalisation permits a smaller plant design with a reduced membrane surface requirement and so higher average plant utilization. Figure 4 shows the influence of buffer tank size on mean plant utilisation, and contrasts the resulting membrane and buffer tank costs. For practical reasons, the buffer tank size is constrained by an HRT of below 2 days (corresponding to 80% of design flow). Since no influent can bypass the plant under storm conditions the combined capacity of the buffer tank and MBR plant must cope with the maximum flow. Figure 4 shows that over the buffer tank size range considered, the cost of adding a buffer tank is only partially offset by the cost savings from a reduction in required membrane surface area due to increased average plant utilization (Figure 5). The EQI and NPV trends are both determined by the constraints on tank size imposed by an HRT of 8h at average flow, or a minimum HRT of 4h at maximum design flow. Addition of a buffer tank with the maximum acceptable size results in a NPV decrease from €30.2 million to €27 million, or a saving of 10.5%, due to decreased opex (-21%), which is partly offset by an increase in capex (+32%) (Figure 6). A maximum NPV saving of 11.8% can be achieved through addition of buffer tank with the most economical size (i.e 1.2 d HRT; at average plant utilization of 47.9%). Effluent quality, as indicated by EQI, is largely unaffected and deteriorates by a maximum of 4% over the buffer tank size range considered (Figure 5).

Insert Figure 4: Influence of buffer tank size on plant utilisation, and a comparison of the resulting costs for membranes and buffer tank

The cost of land required for the buffer tank is excluded from this NPV analysis. However, provided the additional land required for the buffer tank has a projected value of less than €3.2m, it is always beneficial to build a buffer tank. Assuming a total plant footprint equaling 2.5 times the combined footprint required for the biotanks and buffer tanks (Brepols *et al.*, 2010), a plant with the maximum sized buffer tank (2d HRT) requires 9,715 m² extra land compared to a plant without buffer tank. Land costs would have to increase to €324 per m² before addition of a buffer tank becomes economically unviable. This value is at least 32% higher than typical reported values for industrial land in the Germany, which range between €17 and €247 per m² (Ref for industry land – Christoph??). Assuming a CAS to incur 2.7 times the footprint of an MBR (Brepols *et al.*, 2010), a combined MBR with the maximum sized buffer tank would be ~10% larger than a CAS treating the same flow.

358

359 **Insert Figure 5: Influence of average plant utilisation on net present value and effluent**
360 **quality index**

361

362 **3.2 Influence of operational and design parameters**

363 The influence of operational and design parameters on NPV and EQI was evaluated
364 for the plant design with the most economically sized buffer tank, thus providing the
365 lowest NPV (Table 2). Table 3 displays the variation in NPV and EQI resulting from
366 changing parameter values for operation, design and costings within given ranges
367 pertaining to full scale plants.

368

369 **Insert Figure 6: Influence of size of buffer tank on capex and opex**

370 **Insert Table 2: MBR design parameters and base case costs for the study of**
371 **operational and design parameters**

372 **Insert Table 3: Sensitivity of NPV and EQI on design and operational parameters and**
373 **costs. % Change in NPV and EQI is compared with the base conditions as described in**
374 **Table 2**

375

376 **Influence of SRT** A shorter design SRT decreases capex due to decreased installed
377 aerobic tank blower capacity at the lower MLSS concentrations and the resulting
378 decreased aeration demand. However, the cost for the process blowers is less than
379 2% of total capex (Table 1), so the potential influence is negligible. The reduction in
380 NPV is attributed to the effect of SRT on opex. At a conservative sludge treatment
381 and disposal cost of €150.m³ of dry solids, energy consumption accounts for 78-85%
382 of opex, sludge treatment and disposal for 12-19%, and chemical cleaning about 3%.
383 The decreased aeration demand at lower MLSS concentrations and shorter SRT
384 thus outweighs the costs incurred by increased sludge production. This would seem
385 to corroborate recent trends of working at lower MLSS concentrations, particularly in
386 the US (Trussell *et al.*, 2006, 2007), but is contrary to the conclusions of Yoon *et al.*
387 (2004). The latter study ignored membrane aeration, thus underestimating the total
388 opex since membrane aeration contributes significantly to total energy demand
389 (Figure 3).

390

391 The influence of SRT is sensitive to sludge treatment and disposal costs. As sludge
392 management costs increase, the cost incurred by sludge treatment and disposal

393 starts to outweigh the opex reduction from decreased energy demand at lower SRT.
 394 Table 3 also shows that effluent quality requirements place a lower limit on the SRT
 395 operating range, since EQI deteriorates as SRT decreases. Selection of SRT is thus
 396 based on available sludge processing facilities on site and end disposal costs, as
 397 well as the desired effluent quality. Operation at lower SRT and MLSS values may
 398 also lead to higher permeability decline rates (Trussel *et al.*, 2006), mitigating against
 399 lower SRT operation.

400

401 **Influence of HRT** Longer HRTs increase capex due to the larger tank volume
 402 required, but this is partially offset by lower opex at lower MLSS concentrations
 403 (10,000 and 6,000 mg/l average MLSS concentrations in aerobic tank at 6 and 10h
 404 HRT respectively). The impact on NPV is thus negligible compared to, say, the
 405 influence of contingency or choice of SRT. The effect on EQI is more pronounced: an
 406 increase in average HRT from 6 to 10 hours improves effluent quality by 9%. A larger
 407 MBR thus provides better effluent quality, without detriment to NPV provided land
 408 costs are not excessive.

409

410 **Influence of anoxic fraction** Increasing or decreasing the anoxic fraction of total
 411 tank volume has a negligible effect on NPV (Table 3), but a large impact on EQI.
 412 Increasing the anoxic fraction from 30 to 50% improves EQI by 18% due to improved
 413 denitrification.

414

415 **Influence of membrane aeration and sustainable flux** Membrane aeration energy
 416 contributes significantly to opex (Verrecht *et al.*, 2008; Seah, 2009; Brepols *et al.*,
 417 2009), as confirmed by Figure 3. Membrane aeration energy can be related to SAD_p ,
 418 the specific aeration demand per unit permeate volume. Extensive pilot studies
 419 regarding the impact of membrane aeration and sustainable flux (Guglielmi *et al.*
 420 2007, 2008) suggest a neo-linear relationship between sustainable flux J and U , the
 421 in-module air flow velocity in $\text{m}\cdot\text{s}^{-1}$ (Verrecht *et al.*, 2008). For HF geometry,
 422 calibrating against two full scale plants (Verrecht *et al.*, 2008), the correlation
 423 between J and U can be expressed as:

424

$$425 \quad J = m \cdot \frac{SAD_m \cdot L_{membrane}}{\left(\frac{1}{\phi} - \frac{d_f}{4}\right)} + J_0 \quad \text{for } J < J_{sust,max} (\text{l}\cdot\text{m}^{-2}\cdot\text{h}^{-1}) \quad (2)$$

$$426 \quad J = J_{sust,max} (\text{l}\cdot\text{m}^{-2}\cdot\text{h}^{-1}) \quad \text{for } SAD_m > SAD_{m,max} (\text{Nm}^3\cdot\text{m}^{-2}\cdot\text{s}^{-1}) \quad (3)$$

427

428 where J is the flux through the membrane, in $\text{m}^3 \cdot \text{m}^{-2} \cdot \text{h}^{-1}$ and SAD_m the specific
 429 aeration demand per unit membrane area in $\text{Nm}^3 \cdot \text{m}^{-2} \cdot \text{h}^{-1}$, m the slope of the J vs U
 430 curve (247, according to Verrecht *et al.*, 2008), L the membrane module length (1.8
 431 m); φ the module packing density (300 m^{-1}), d_f the hollow fibre outside diameter
 432 (0.002 m) and J_0 the intercept of the J vs. U curve ($5 \text{ l} \cdot \text{m}^{-2} \cdot \text{h}^{-1}$). Thus:

433

$$434 \quad SAD_p = \frac{SAD_m}{J} \quad (4)$$

435

436 where a minimum SAD_p is required to maintain a sustainable flux (2), but increasing
 437 SAD_p beyond $SAD_{m,max}$ has no impact on the sustainable flux (3) and a higher
 438 maximum sustainable flux demands a higher SAD_p . When considering the influence
 439 of sustainable flux and SAD_p on NPV (Figure 7), higher sustainable fluxes lead to
 440 lower NPVs, indicating that the higher operational costs are offset by lower capital
 441 expenditures which can mainly be attributed to the reduction in membrane capacity
 442 required. An increase in sustainable flux from 15 to $30 \text{ l} \cdot \text{m}^{-2} \cdot \text{h}^{-1}$ results in a decrease
 443 in NPV of 9% (at minimum required SAD_p ; Table 3). It can thus be concluded that
 444 higher sustainable fluxes are beneficial to NPV, despite the higher aeration demand
 445 and associated increase in opex, provided $J \leq J_{sust,max}$ and J_0 takes a positive value.

446

447 **Insert Figure 7: Influence of SAD_p on net present value for a range of sustainable fluxes**

448

449 **Energy cost** For an annual energy price rise of 4%, in line with the historical average
 450 (EIA, 2009), a 5.7% increase NPV arises over the base case for inflation-linked
 451 energy costs. A 'worst case' of a 7% annual increase, corresponding to a doubling of
 452 energy prices roughly every 10 years, increases NPV by 30%.

453

454 **Membrane replacement and cost** As shown in Table 1, membrane costs make up
 455 47-57% of total capex, while the other process equipment combined contributes
 456 about 20%. Analysis of component lifetime cost impacts is thus most sensitive to
 457 membrane life and costs. A 'worst case' membrane lifetime of 5 years (i.e. 6
 458 membrane replacements in the projected plant lifetime of 30 years) results in a 23%
 459 increase in NPV compared to the base cost assuming membrane replacement every
 460 10 years. A halving of membrane costs every 10 years, on the other hand, reduces
 461 NPV by 9.2%, whereas an increase in initial membrane cost from $\text{€}20 \cdot \text{m}^{-2}$ to $\text{€}100 \cdot \text{m}^{-2}$

462 increases NPV by 50% for a 10 year membrane life and by 85% for a 5 year
463 membrane lifetime.

464

465 Since membrane replacement is critical in determining NPV, it is unsurprising that
466 considerable attention has been paid to optimisation of membrane lifetime by
467 operating under a sustainable regime and developing adequate cleaning strategies
468 (Brepols *et al.*, 2008). There is increasing evidence that MBR membrane life can
469 reach, or even exceed, a decade for large plants. The Zenon plants at Rodingen (3.2
470 MLD PDF) and Brescia (42 MLD PDF) are successfully operating with membrane
471 modules which are from 2000 and 2002 years respectively, and the Kubota plant at
472 Porlock still operates with 40% of the panels originally installed in 1997 (Judd and
473 Judd, 2010); predicted replacement intervals of up to 13 years have been reported
474 (De Wilde *et al.*, 2007b).

475

476 **4. Conclusions**

477 A cost sensitivity analysis, using dynamic simulation results, with respect to design
478 and operational parameters for an MBR over the lifetime of the plant has revealed:

479

- 480 1. The contingency provided for changes in feedwater flow and composition impacts
481 significantly on net present value (NPV). The analysis shows that any deviation
482 from the ideal 'hybrid' plant, where the MBR treats a constant influent stream,
483 leads to plant under-utilisation and a resulting cost penalty manifested as an
484 increase of up to 58% in NPV for a plant designed for three times the mean flow.
- 485 2. Addition of a buffer tank for flow equalisation increases average plant utilisation,
486 leading to more efficient operation and a resulting reduction in opex, whilst capex
487 can also be reduced according to the reduction in membrane area and MBR plant
488 size. In the example presented, a decrease in NPV of up to 11% with increased
489 average plant utilisation from 34 to 48% results.
- 490 3. Addition of a buffer tank is economically beneficial as long as the cost of land
491 required is less than the NPV saving achieved. In the example presented,
492 addition of a buffer tank is economically viable for increased land costs below
493 €324.m², an excessive value for industrial land. An MBR with the maximum sized
494 buffer tank (2d HRT) has a footprint approximately 10% greater than that of a
495 conventional activated sludge plant.
- 496 4. An increased SRT at constant tank volume increases the NPV since a greater
497 aeration demand is incurred at higher MLSS concentrations. Whilst sludge
498 production is concomitantly reduced, the resulting cost savings do not fully offset

- 499 the increased energy costs. However, results are very sensitive to sludge
500 treatment and disposal costs. The effect of HRT on NPV is minimal, if land costs
501 are negligible, but a higher average HRT improves effluent quality.
- 502 5. Higher sustainable fluxes provide a decreased NPV. Although the opex is
503 increased due to the higher aeration demand, this is offset by the reduction in
504 capex and membrane replacement costs since less membrane area is required.
505 An increase in sustainable flux from 15 to 30 LMH decreases NPV by 9%.
- 506 6. The future trend in energy costs is a determining factor for NPV: a doubling of
507 energy costs every ten years increases the NPV by 30%.
- 508 7. A membrane lifetime of 5 years results in an NPV 23% higher compared to a 10
509 year membrane replacement interval, for a constant membrane cost of €50.m². If
510 initial membrane costs increase five-fold from €20 per m², NPV increases by 85%
511 for a 5 year membrane lifetime and by 50% for a 10 year membrane life.

512 Acknowledgements

513 The authors are grateful to MOSTforWATER N.V. (Kortrijk, Belgium) for providing the
514 WEST® modelling software, and would also like to thank Thames Water for the
515 resources provided by them for this paper. Thomas Maere is supported by the
516 Institute for Encouragement of Innovation by means of Science and Technology in
517 Flanders (IWT).

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Table 1: Capex, opex and resulting NPV for an MBR treating steady state influent, as part of a hybrid plant, and a MBR, designed for maximum flow without buffer tanks.

	<i>Unit</i>	<i>MBR part of a hybrid plant</i>	<i>Plant designed for maximum flow</i>
<i>Average plant influent flow</i>	m ³ .d ⁻¹	20,851	20,851
<i>Maximum flow to the MBR</i>	m ³ .d ⁻¹	20,851	59,580
<i>Total tank volume</i>	m ³	6,949	9,930
<i>Average plant utilisation</i>	%	100%	34%
<i>Effluent Quality Index</i>	kg PU.d ⁻¹	5,035	5,236
<i>COD_{average}</i>	mg.l ⁻¹	29.7	30.15
<i>NH₄-N_{average}</i>	mg.l ⁻¹	0.46	0.43
<i>NO₃-N_{average}</i>	mg.l ⁻¹	10.4	9.55
TOTAL CAPEX	Euro	4,634,387	7,844,684
<i>Screens</i>	%	11.8	8.4
<i>Membranes</i>	%	46.9	56.5
<i>Tank construction</i>	%	33.0	27.9
<i>Biology blowers</i>	%	1.4	0.8
<i>Membrane blowers</i>	%	1.5	1.6
<i>Permeate pumps</i>	%	1.5	2.2
<i>Mixing equipment</i>	%	1.9	1.4
<i>Recirculation pumps</i>	%	2	1.2
TOTAL OPEX	Euro/year	618,602	891,373
<i>Energy</i>	%	79.6	84.1
<i>Sludge treatment and disposal</i>	%	17.9	12.3
<i>Chemicals</i>	%	2.5	3.6
NET PRESENT VALUE	Euro	19,047,870	30,209,875

Table 2: MBR design parameters and base case costs for the study of operational and design parameters

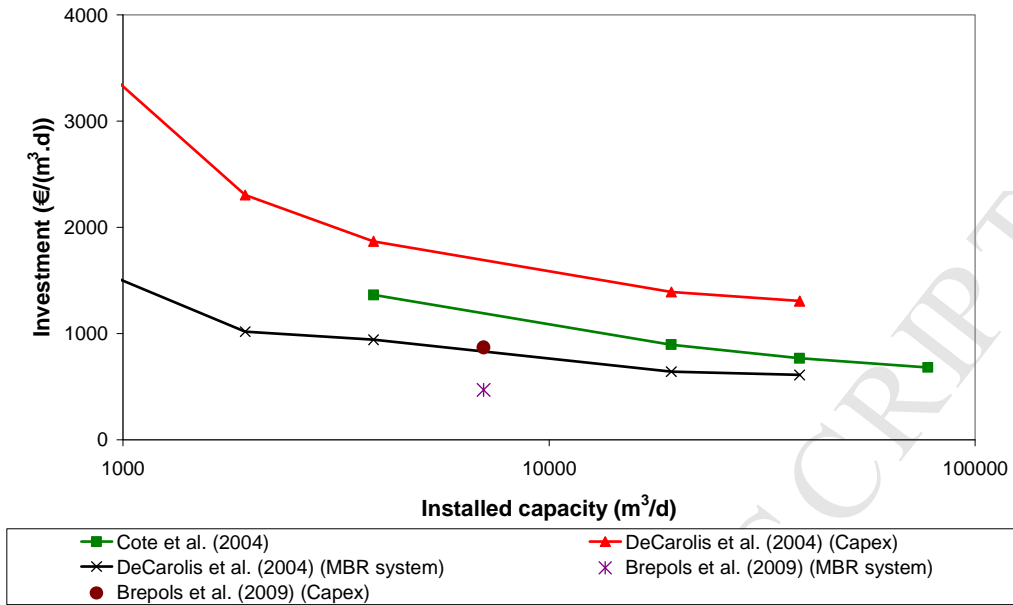
Parameter	Units	Value	Reference	Units	Value
<i>Assumptions for capex calculation</i>				<i>Base design, EQI and NPV</i>	
Membrane cost	€·m ⁻²	50	Judd & Judd, 2010	Design capacity	m ³ ·d ⁻¹ 30,416
Tank civil cost	€·m ⁻³ tank volume	220	Brepols, 2010b	Maximum plant capacity*	m ³ ·d ⁻¹ 42,582
Screens – 0.75 mm	€·m ⁻³ ·d ⁻¹ capacity	3.1 – 5.6**	Manufacturers	Total tank volume	m ³ 7,097
Screens – 6mm	€·m ⁻³ ·d ⁻¹ capacity	0.9 – 2.1**	Manufacturers	Membrane area	m ² 63,366
Blowers	€·Nm ⁻³ ·h ⁻¹ capacity	4 – 4.3**	Manufacturers	SRT	d 23.8
Permeate pumps	€·m ⁻³ ·h ⁻¹ capacity	58.8	Manufacturers; Brepols, 2010b		
Biomass recirculation pumps	€·m ⁻³ ·h ⁻¹ capacity	12.1	Manufacturers; Brepols, 2010b	Buffer tank size	m ³ 14,530
Mixing equipment	€·m ⁻³ tank volume	27.8	Brepols, 2010b	Maximum flow out of buffertank*	m ³ ·d ⁻¹ 12,166
<i>Assumptions for opex calculation</i>				Max HRT in buffer tank	d 1.2
Energy cost	€·kWh ⁻¹	0.0942	- ref Christophe	Effluent quality index	kg PU·d ⁻¹ 5,430
Sludge treatment cost	€·ton ⁻¹ of DS	150	-	NH ₄ -N	mg·l ⁻¹ 0.52
Citric acid 50%	€/ton	760	Brepols, 2010b	NO ₃ -N	mg·l ⁻¹ 10.7
NaOCl 14%	€/m ³	254	Brepols, 2010b	COD	mg·l ⁻¹ 30.1
<i>Assumptions for NPV calculation</i>				Net present value	Million Euro (M€) 26.7
Membrane life	Year	10	Judd & Judd, 2010		
Inflation	%	3%	-		
Discount rate	%	6%	-		

* As determined by the design requirement that maximum sustainable flux = 140% of design flux

** Depending on size of installed equipment

Table 3: Sensitivity of NPV and EQI on design and operational parameters and costs. % Change in NPV and EQI is compared with the base conditions as described in Table 2

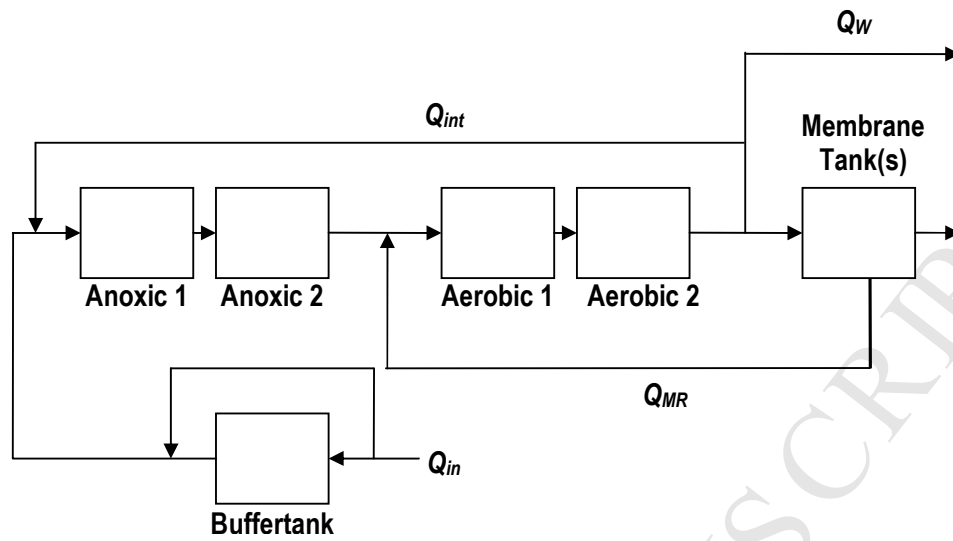
	Net present value		EQI	
	Million euro (M€)	% change	kg PU.d ⁻¹	% change
<i>Solids retention time (SRT)</i>				
9.5 days	26.4	-1.1%	5,835	+7.5%
47.6 days	27.8	+4.4%	5,172	-4.7%
<i>Hydraulic residence time (HRT)</i>				
6 hours	26.3	-1.3%	5,628	3.7%
10 hours	26.8	+0.5%	5,214	-4.0%
<i>Sustainable flux at membrane aeration</i>				
15 l.m ⁻² .h ⁻¹ at SAD _p = 15.3	29.1	9.2%	5,551	+2.2%
30 l.m ⁻² .h ⁻¹ at SAD _p = 19.1	26.5	-0.5%	5,295	-2.5%
<i>Buffer tank</i>				
0 days HRT (No buffertank)	30.2	+13.4	5,236	-3.6%
2 days HRT (Maximum considered)	27.1	+1.6%	5,401	-0.5%
<i>Anoxic tank volume</i>				
f _{anox} = 30%	26.7	+0.1%	6,313	+16.3
f _{anox} = 50%	26.6	-0.3%	5,146	-5.2
<i>Energy prices</i>				
Rising by 4% annually	28.2	+5.7%	5,430	0%
Rising by 7% annually	34.7	+30.0%	5,430	0%
<i>Sludge treatment costs (excluding hauling)</i>				
43 Eur.ton ⁻¹ of DS	25.2	-5.6%	5,430	0%
300 Eur.ton ⁻¹ of DS	28.8	+7.9%	5,430	0%
<i>Membrane costs</i>				
20 Eur.m ⁻² membrane surface	22.4	-15.8%	5,430	0%
100 Eur.m ⁻² membrane surface	33.7	+26.4%	5,430	0%
Membrane costs – halving every ten years	24.2	-9.3%	5,430	0%
Membrane life – 5 years	32.8	+23.1%	5,430	0%



1

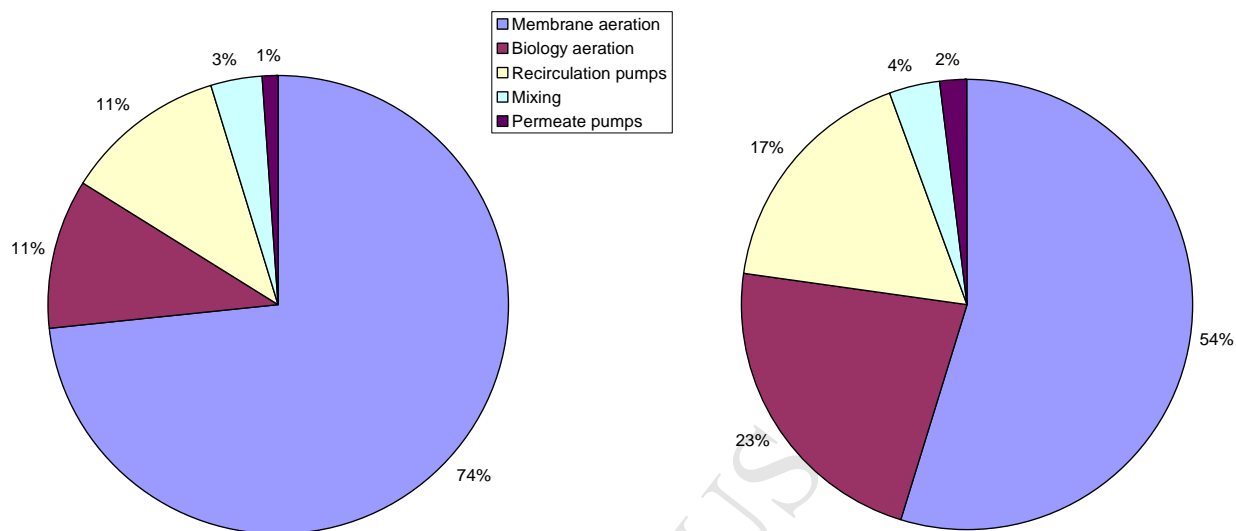
2 Figure 1: Specific investment vs. installed plant capacity, based on literature data (adapted
 3 from McAdam and Judd, 2006)

4



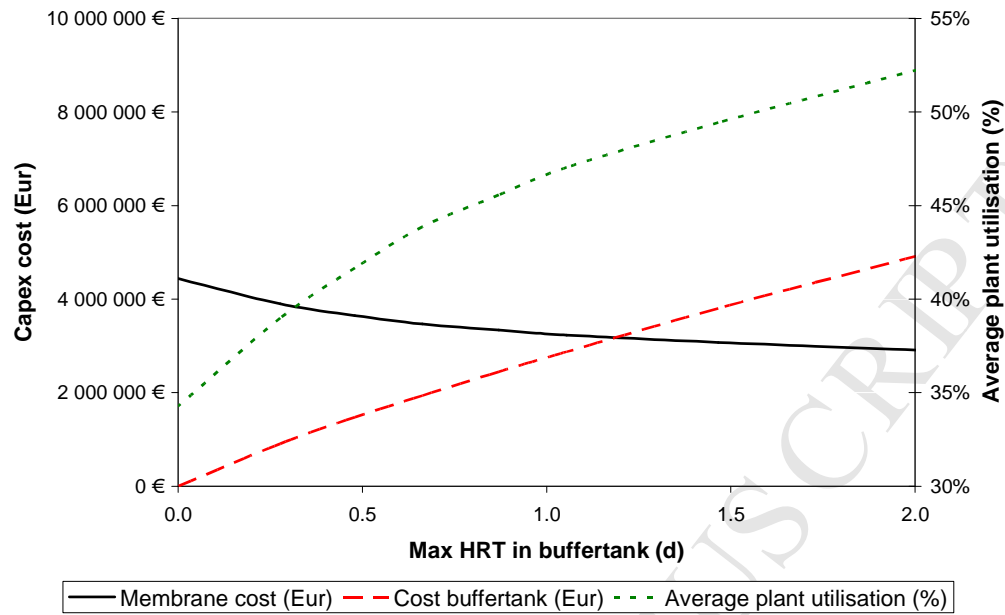
5

6 **Figure 2: Schematic overview of the generic nitrifying/denitrifying MBR design**

7
8

9 **Figure 3: Breakdown of energy demand for a) the MBR part of a hybrid plant (Average**
10 **total energy demand = 0.7 kWh.m⁻³) and b) a plant designed for maximum flow**
11 **(Average total energy demand = 1.07 kWh.m⁻³)**

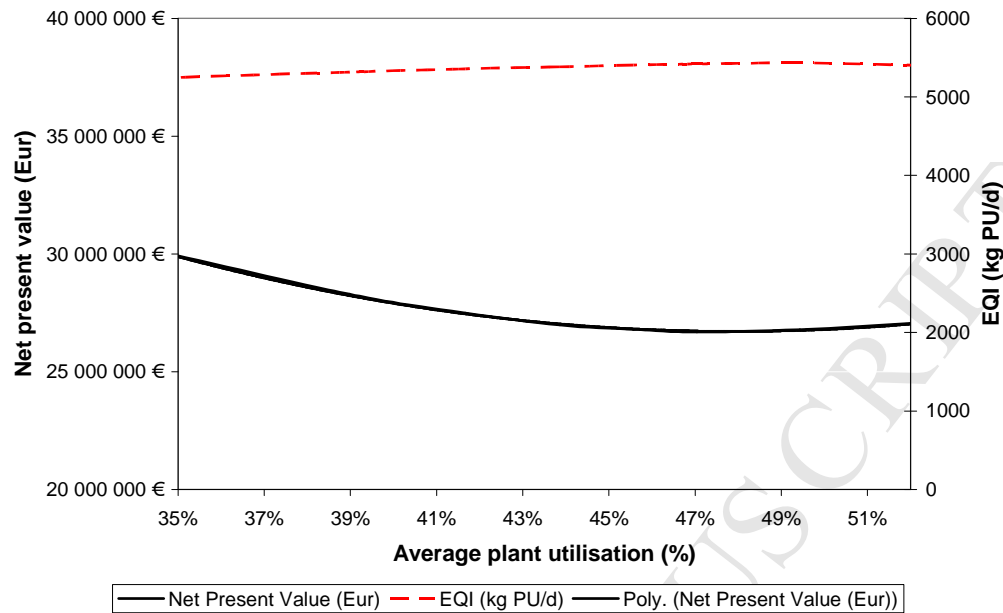
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14 Figure 4: Influence of buffer tank size on plant utilisation, and a comparison of the resulting
 15 costs for membranes and buffer tank

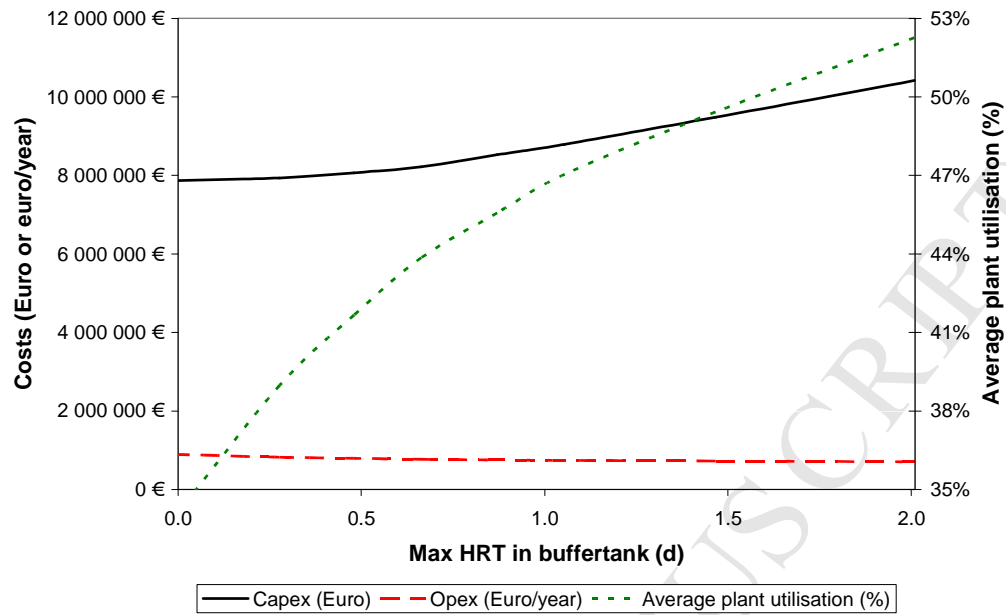
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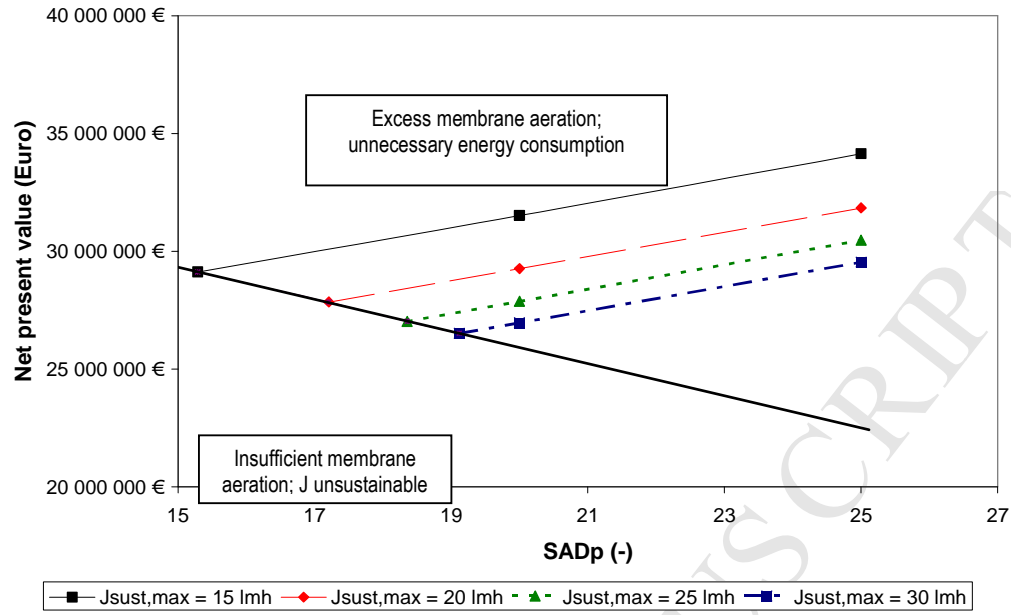
18 Figure 5: Influence of average plant utilisation on net present value and effluent quality index

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21 Figure 6: Influence of size of buffer tank on capex and opex



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Figure 7: Influence of SAD_p on net present value for a range of sustainable fluxes