Supporting Information

Techno-economic Assessment of Optimised Vacuum Swing Adsorption for Post-Combustion CO₂ Capture from Steam-Methane Reformer Flue Gas

Sai Gokul Subraveti¹, Simon Roussanaly^{2,*}, Rahul Anantharaman², Luca Riboldi², and Arvind Rajendran^{1,*}

¹Department of Chemical and Materials Engineering, University of Alberta, 12th floor, Donadeo Innovation Centre for Engineering (ICE), 9211-116 Street, Edmonton, Alberta T6G1H9, Canada ²SINTEF Energy Research, NO-7465, Trondheim, Norway *Corresponding authors. E-mail: Simon.Roussanaly@sintef.no (Simon Roussanaly), arvind.rajendran@ualberta.ca (Arvind Rajendran)

S1 Baseline MEA-based CO₂ capture



Figure S1: Detailed process flow diagram of the MEA-based CO_2 capture process for the hydrogen production plant with CO_2 capture [1].

Parameter	Without CCS	With CCS
Natural Gas to feedstock (t h^{-1})	51.66	51.66
Natural Gas to fuel (t h^{-1})	26.59	26.59
Natural Gas LHV (MJ kg^{-1})	46.49	46.49
Total Energy Input (MW)	1010	1010
H_2 to battery limit (t h ⁻¹)	18.77	18.77
H_2 to battery limit (Nm ³ H ₂ h ⁻¹)	208700	208700
Total energy in H_2 product (MW)	626	626
Gross power output from Steam cycle (MW_e)	123.8	91.6
H_2 plant and co-generation power consumption (MW _e)	-3.5	-3.5
$\rm CO_2$ capture plant (MW _e)	-	-6.7
CO_2 conditioning plant (MW _e)	-	-18.3
Net Power output (MW_e)	120.3	63.1
Total energy in H_2 product compared Total Energy Input (%)	61.9	61.9
Total energy in H_2 and electricity produced compared Total Energy Input (%)	73.8	68.2
Emissions $(kg_{CO_2} Nm^{-3} H_2)$	0.994	0.100
Levelised Cost of Hydrogen ($c \in Nm^{-3} H_2$)	12.20	18.07
CO_2 avoidance cost ($\in/t_{CO_2,avoided}$)	-	66.6
CO_2 capture cost ($\in/t_{CO_2,avoided}$)	-	30.1

Table S1: Key performances of hydrogen production plant without and with MEA-based CCS [2].

S2 Adsorbent Materials

	Zeolite 13X [3]	UTSA-16 [4]	IISERP MOF2 [5]
CO ₂			
$q_{\rm sb} \ ({\rm mol} \ {\rm kg}^{-1})$	3.09	4.08	3.29
$q_{\rm sd} \ ({\rm mol} \ {\rm kg}^{-1})$	2.54	1.29	1.89
$b_0 \; (\mathrm{m}^3 \; \mathrm{mol}^{-1})$	8.65×10^{-7}	2.52×10^{-7}	9.39×10^{-8}
$d_0 \; (\mathrm{m}^3 \; \mathrm{mol}^{-1})$	2.63×10^{-8}	1.75×10^{-9}	5.23×10^{-7}
$\Delta U_{\rm b} \ ({\rm J} \ {\rm mol}^{-1})$	-36641	-32800	-31135
$\Delta U_{\rm d} \ ({\rm J} \ {\rm mol}^{-1})$	-35690	-35040	-31135
N_2			
$q_{\rm sb} \ ({\rm mol} \ {\rm kg}^{-1})$	3.09	1.33	3.29
$q_{ m sd} \ ({ m mol} \ { m kg}^{-1})$	2.54	1.77	1.89
$b_0 \; (\mathrm{m}^3 \; \mathrm{mol}^{-1})$	2.69×10^{-6}	9.17×10^{-5}	2.55×10^{-7}
$d_0 \; (\mathrm{m}^3 \; \mathrm{mol}^{-1})$	2.69×10^{-6}	9.42×10^{-9}	2.55×10^{-7}
$\Delta U_{\rm b} \ ({\rm J} \ {\rm mol}^{-1})$	-15710	-7500	-11890
$\Delta U_{\rm d} \ ({\rm J} \ {\rm mol}^{-1})$	-15710	-27760	-11890

Table S2: Dual-site Langmuir isotherm parameters.

S3 Technical Modelling of Vacuum Swing Adsorption

S3.1 Model Equations

Component mass balance

$$\frac{\partial y_{i}}{\partial t} + \frac{y_{i}}{T}\frac{\partial P}{\partial t} - \frac{y_{i}}{P}\frac{\partial T}{\partial t} = \frac{T}{P}D_{L}\frac{\partial}{\partial z}\left(\frac{P}{T}\frac{\partial y_{i}}{\partial z}\right) - \frac{T}{P}\frac{\partial}{\partial z}\left(\frac{y_{i}P}{T}v\right) - \frac{RT}{P}\frac{1-\varepsilon}{\varepsilon}\frac{\partial q_{i}}{\partial t}$$
(S1)

Total mass balance

$$\frac{1}{P}\frac{\partial P}{\partial t} - \frac{1}{T}\frac{\partial T}{\partial t} = -\frac{T}{P}\frac{\partial}{\partial z}\left(\frac{P}{T}v\right) - \frac{RT}{P}\frac{1-\varepsilon}{\varepsilon}\sum_{i=1}^{n_{\rm comp}}\frac{\partial q_i}{\partial t}$$
(S2)

Linear driving force model

$$\frac{\partial q_{\mathbf{i}}}{\partial t} = k_{\mathbf{i}}(q_{\mathbf{i}}^* - q_{\mathbf{i}}) \tag{S3}$$

Mass transfer coefficient (macropore controlled)

$$k_{\rm i} = \frac{c_{\rm i}}{q_{\rm i}^*} \frac{15\varepsilon_{\rm p} D_{\rm p}}{r_{\rm p}^2} \tag{S4}$$

Column energy balance

$$\left[\frac{1-\varepsilon}{\varepsilon}\left(\rho_{s}C_{p,s}+C_{p,a}\sum_{i=1}^{n_{comp}}q_{i}\right)\right]\frac{\partial T}{\partial t}=\frac{K_{z}}{\varepsilon}\frac{\partial^{2}T}{\partial z^{2}}-\frac{C_{p,g}}{R}\frac{\partial P}{\partial t}-\frac{C_{p,g}}{R}\frac{\partial}{\partial z}\left(vP\right)-\frac{1-\varepsilon}{\varepsilon}C_{p,a}T\sum_{i=1}^{n_{comp}}\frac{\partial q_{i}}{\partial t}+\frac{1-\varepsilon}{\varepsilon}\sum_{i=1}^{n_{comp}}\left(\left(-\Delta H\right)\frac{\partial q_{i}}{\partial t}\right)$$
(S5)

Pressure drop (Ergun's equation)

$$-\frac{\partial P}{\partial z} = \frac{150}{4} \frac{1}{r_{\rm p}^2} \left(\frac{1-\varepsilon}{\varepsilon}\right)^2 \mu v + \frac{1.75}{2} \frac{1}{r_{\rm p}} \left(\frac{1-\varepsilon}{\varepsilon}\right) \rho |v|v \tag{S6}$$

Ideal gas law

$$c_{\rm i} = \frac{y_{\rm i}P}{RT} \tag{S7}$$

Step	z=0	z=L
	$v _{z=0} = v_{\text{feed}}$	$P _{z=L} = P_{H}$
Adsorption	$D_{\mathrm{L}} \left. \frac{\partial y_i}{\partial z} \right _{z=0} = -v _{z=0} \left(y_{i,\mathrm{feed}} - y_i _{z=0} \right)$	$\left. \frac{\partial y_i}{\partial z} \right _{z=\mathrm{L}} = 0$
	$\frac{\partial T}{\partial z}\Big _{z=0} = -\varepsilon v\Big _{z=0} \rho_{g} C_{p,g} (T_{\text{feed}} - T _{z=0})$	$\left. \frac{\partial T}{\partial z} \right _{z=L} = 0$
	$\left. \frac{\partial P}{\partial z} \right _{z=0} = 0$	$v _{z=L} = v_{\text{vac.pump}}$
Blowdown	$\left. \frac{\partial y_i}{\partial z} \right _{z=0} = 0$	$\left. \frac{\partial y_i}{\partial z} \right _{z=\mathrm{L}} = 0$
	$\left. \frac{\partial T}{\partial z} \right _{\mathbf{z}=0} = 0$	$\left. \frac{\partial T}{\partial z} \right _{z=L} = 0$
	$v _{\rm z=0} = v_{\rm vac.pump}$	$\left. \frac{\partial P}{\partial z} \right _{z=\mathrm{L}} = 0$
Evacuation	$\left. \frac{\partial y_i}{\partial z} \right _{z=0} = 0$	$\left. \frac{\partial y_i}{\partial z} \right _{z=\mathrm{L}} = 0$
	$\left. \frac{\partial T}{\partial z} \right _{\mathbf{z}=0} = 0$	$\left. \frac{\partial T}{\partial z} \right _{z=L} = 0$
Light Product Pressurisation	$\left. \frac{\partial P}{\partial z} \right _{z=0} = 0$	$v _{z=L} = \frac{v_{ADS}P_{ADS} _{z=L}}{P _{z=L}}$
	$\left. \frac{\partial y_i}{\partial z} \right _{z=0} = 0$	$D_{\mathrm{L}} \left. \frac{\partial y_i}{\partial z} \right _{\mathrm{z}=\mathrm{L}} = -v _{\mathrm{z}=\mathrm{L}} \left(y_{i,\mathrm{feed}} - y_i _{\mathrm{z}=\mathrm{L}} \right)$
	$\frac{\partial T}{\partial z}\Big _{z=0} = 0$	$\frac{\partial T}{\partial z}\Big _{z=L} = -\varepsilon v\Big _{z=L} \rho_g C_{pg} (T_{\text{feed}} - T _{z=L})$

Table S3: Boundary conditions for the 4-step VSA cycle.

Parameter	Value
Column Properties	
Particle diameter, $d_{\rm p}$ (mm)	1.5
Column void fraction, $\epsilon_{\rm B}$ (-)	0.37
Particle void fraction, $\epsilon_{\rm P}$ (-)	0.35
Tortuosity, τ (-)	3
Operating Conditions	
Adsorption pressure, $P_{\rm H}$ (bar)	1.02
Inlet feed composition, y_{CO_2}/y_{N_2} (-)	0.2/0.8
Inlet feed temperature, T_{feed} (K)	298.15
Physical Properties	
Adsorbent density, $\rho_{\rm s}$ (kg m ⁻³)	
Zeolite 13X	1130.0 [3]
UTSA-16	1171.0 [4]
IISERP MOF2	937.7 [5]
Molecular diffusivity, $D_{\rm m} \ ({\rm cm}^2 \ {\rm s}^{-1})$	0.16
Fluid viscosity, μ (cP)	0.0172
Specific heat capacity of adsorbent, $C_{p,s}$ (J kg ⁻¹ K ⁻¹)	
Zeolite 13X	1070.0
UTSA-16	1070.0
IISERP MOF2	1070.0
Specific heat capacity of gas phase, $C_{p,g}$ (J mol ⁻¹ K ⁻¹)	30.7
Specific heat capacity of adsorbed phase, $C_{p,a}$ (J mol ⁻¹ K ⁻¹)	30.7
Inside heat transfer coefficient, $h_{\rm in}$ (J m ⁻² K ⁻¹ s ⁻¹)	0
Outside heat transfer coefficient, h_{out} (J m ⁻² K ⁻¹ s ⁻¹)	0
Effective gas thermal conductivity, K_z (J m ⁻¹ K ⁻¹ s ⁻¹)	0.09
Universal gas constant, R (m^3 Pa mol ⁻¹ K ⁻¹)	8.314

Table S4: VSA simulation parameters.

S3.2 Design of Unit Train

The procedure proposed by Khurana and Farooq [6] was used to determine the column scheduling. Each train comprises minimum number of columns and vacuum pumps necessary for a continuous operation. The minimum number of columns per train was calculated as follows:

$$N = \operatorname{ceiling}\left(\frac{\sum_{i=\operatorname{steps}} t_i}{t_{\operatorname{ADS}}}\right) \tag{S8a}$$

 t_i represents the duration of step *i* in the cycle. The minimum number of blowdown/evacuation vacuum pumps required is given by,

$$N_{\rm V,j} = \text{ceiling}\left(\frac{t_j}{t_{\rm ADS}}\right) \quad j = \text{blowdown/evacuation}$$
(S8b)

If sum of the individual steps in a cycle is not a multiple of the adsorption time, an idle step has to be included after evacuation step so that the bed profiles are least affected [6]. The duration of an idle step was calculated as follows:

$$t_{\rm IDLE} = N t_{\rm ADS} - \sum_{i=\rm steps} t_i \tag{S8c}$$

S3.3 Parallel Trains

A single VSA train might not be sufficient to treat the large volume of flue gas. Hence, several trains of VSA units in parallel are required to capture 90% CO_2 [6,7]. The number of parallel trains can be calculated as:

$$M = \operatorname{ceiling}\left(\frac{\dot{F}_{\text{flue}}}{\dot{F}_{\text{train}}}\right) \tag{S9}$$

Here \dot{F}_{flue} is the total flue gas flow rate in kmol h⁻¹ and \dot{F}_{train} is the average molar flow rate of the feed to each train in kmol h⁻¹. It is worth mentioning that the inlet pressure varies over the duration of the adsorption step owing to the constant velocity boundary condition at the feed end. Therefore, the average molar flow rate of feed to each train was calculated based on an integral average of the molar flow rate over the duration of the adsorption step (shown in Eq. S10) and then, used to calculate the number of parallel trains.

$$\dot{F}_{\text{train}} = \frac{1}{t_{\text{ADS}}} \int_0^{t_{\text{ADS}}} \dot{F} dt \tag{S10}$$

S4 Technical Modelling of peripheral units

The implementation of CO_2 capture using VSA technology requires several peripheral units extending from flue gas pre-treatment to CO_2 conditioning. In this section, the technical modeling related to each component unit is discussed below.

Flue Gas Cooling and Drying: The wet flue gas was first cooled to 313.15 K by a direct contact cooler and then dehydrated using a molecular sieve 3Å [8].

Compressors: Single-stage compressors were modeled as an isoentropic compression process. The motor efficiency was assumed to be 100%. The energy consumption was calculated as follows:

$$E_{\rm C} \left({\rm J}_{\rm e} \right) = \frac{1}{\eta_{\rm C}} \frac{\gamma}{\gamma - 1} \int_{t=0}^{t=t_{\rm ADS}} QP \left[\left(\frac{P}{P_{\rm ref}} \right)^{\frac{\gamma - 1}{\gamma}} - 1 \right] \mathrm{d}t$$
(S11a)

Here $\eta_{\rm C}$ is the compression efficiency which was assumed to be 80%, γ is the adiabatic constant obtained from a linear regression as a function of CO₂ composition (see Fig. S2), P is the pressure, $P_{\rm ref}$ is the reference pressure of flue gas, $t_{\rm ADS}$ is the adsorption step time and Q is the volumetric flow rate of the feed mixture.

Vacuum Pumps: The energy consumption by a vacuum pump was modeled as an isentropic expansion process as given by,

$$E_{\rm V} \left(\mathbf{J}_{\rm e} \right) = \frac{1}{\eta_{\rm V}} \frac{\gamma}{\gamma - 1} \int_{t=0}^{t=t_{\rm step}} QP \left[\left(\frac{P_{\rm atm}}{P} \right)^{\frac{\gamma - 1}{\gamma}} - 1 \right] \mathrm{d}t \tag{S11b}$$

In the above equation, t_{step} is the step duration of blowdown/evacuation step, η_{V} is the vacuum pump efficiency.

Heat Exchangers: Two identical counter-current heat exchangers were considered to cool the dry flue gas after compression to 298.15 K. The design was evaluated based on the cooling duty and the log-mean temperature difference (LMTD) for the counter-current flow. The dry flue gas represents the hot side of the heat exchangers while the cooling water is the cold side. To determine the cooling duty, input and output stream characteristics of the hot dry flue gas were used. While the mass flow rate, input and output temperatures of the dry flue gas were known,

the specific heat capacity was obtained from the National Institute of Standards and Technology (NIST) REFPROP v.9 database [9]. The mass flow rate of the cooling water was then determined by dividing the cooling duty by the heat capacity [9] and an allowable temperature increase of the cooling water. The inlet and outlet temperatures of cooling water were set to 283.15 K and 291.5 K, respectively. The heat exchanger area ($A_{\rm EX}$) was obtained using,

$$A_{\rm EX} = \frac{\dot{Q}_{\rm EX}}{U_{\rm EX} \rm LMTD} \tag{S12}$$

where Q_{EX} is the cooling duty (W) and U_{EX} is the overall heat transfer coefficient which is assumed to be around 1000 W m⁻² K⁻¹ for all process heat exchangers [10].

 CO_2 Conditioning: The CO₂ after capture undergoes compression from 1 bar, 298.15 K to offshore pipeline transport conditions at 200 bar and 318.15 K. The CO₂ conditioning before pipeline transport was modelled as a four-stage compression system with intercoolers and a pump to deliver the CO₂ at desired pressure in Aspen HYSYS. The readers are referred elsewhere [2] for detailed modeling of CO₂ conditioning.

 CO_2 Transport and Storage: The costs of the transport and storage are assessed using the iCCS tool developed by SINTEF Energy Research [11] and previously documented [12–14]. The transport cost model relies on the pipeline cost model developed by Knoope et al. [15] and the storage cost model relies on the Zero Emission Platform for Zero Emission Fossil Fuel Power Plants [16].



Figure S2: Linear dependence of adiabatic constant (γ) as a function of CO₂ mole fraction. Note that the γ values were obtained from NIST database [9].



Figure S3: Optimal cycle schedules for all three adsorbents.

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