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Analysis and modelling of innovative technologies on natural
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ABSTRACT

Renewable energy sources (RESs), such as wind, solar and biomass, are the keystone of the energy policy of the EU to fulfil the target of a carbon neutral economy by 2050 [1]. However, the integration of a significant share of RESs pose significant challenges to the EU energy systems as it requires, on the one hand, storing large energy volumes to match intermittent renewable supply with the pattern of energy consumption and, on the other hand, transporting renewable energy from where it can be most efficiently and feasibly produced to where it is consumed.

To overcome such challenges, it will be effective to consider the reuse of the existing natural gas (NG) infrastructure. The latter will play a crucial role in the development of a decarbonized energy system based on a large usage of RESs due to its widespread presence and its capacity to provide a cost-effective option for transporting and storing large amounts of energy for long-term period exploiting the NG transportation and distribution networks, as well as the storage complexes of the existing NG infrastructure.

This thesis aims to address, with a multi-thematic approach, the issue of innovative uses of, as well as the development of innovative technologies on, NG transport and distribution networks. To this aim, a number of five case studies were investigated to: i) evaluate the fault management strategies for NG distribution networks to minimize the disservice and to define possible structural improvement measures; ii) assess the technical feasibility of the Power to Gas concept to store intermittent RESs; iii) analyse the impact of hydrogen injection on NG networks; iv) evaluate the effectiveness of the equilibrium gasification models as a modelling tool for the design and optimization of biomass gasification systems integrated into polygenerative plants coupled with energy networks.

Results of this thesis provide useful insights to researchers, designers and policy makers, filling some of the gaps highlighted in the existing scientific literature in all the analysed areas.

TABLE OF CONTENTS

ABSTRACT.....	I
List of Figures.....	VI
List of Tables.....	X
INTRODUCTION.....	1
CHAPTER 1. ANALYSIS OF RESILIENCE OF NATURAL GAS NETWORKS	7
1.1 Defining resilience for Critical Infrastructures.....	7
1.2 Title Literature review on resilience analysis of natural gas networks	9
1.3 <i>Case study #1: A resilience analysis simulation tool for natural gas distribution networks</i>	11
1.3.1 Theory and methods.....	11
1.3.2 Application of the resilience analysis simulation tool to a real natural gas distribution network.....	20
1.3.3 Results and discussion.....	23
CHAPTER 2. ANALYSIS OF POWER TO GAS TECHNOLOGY.....	29
2.1 Power to Gas concept.....	29
2.2 State of the art technologies.....	31
2.2.1 Electrolysis process.....	31
2.2.1.1 Alkaline electrolysis.....	32
2.2.1.2 Proton exchange membrane electrolysis.....	34
2.2.1.3 Solid oxide electrolysis.....	36
2.2.2 Methanation process.....	39
2.2.2.1 Fundamentals.....	39
2.2.2.2 Process concepts.....	40
2.2.2.2.1 Adiabatic fixed-bed reactors.....	41
2.2.2.2.1.1 Process with recycle.....	41
<i>TREMP process</i>	41
<i>Lurgi process</i>	42
<i>HICOM process</i>	44
2.2.2.2.1.2 Process without recycle.....	45

<i>RMP process</i>	45
<i>ICI process</i>	47
<i>VESTA process</i>	48
2.2.2.2.2 Isothermal fixed-bed reactors	48
<i>Linde process</i>	49
2.2.2.2.3 Fluidized-bed reactors	50
<i>Comflux process</i>	51
2.2.2.2.4 Three-phase reactors.....	53
2.3 <i>Case study #2: SNG generation via Power to Gas technology: plant design and annual performance assessment</i>	55
2.3.1 Material and methods	55
2.3.1.1 PtSNG Plant description	55
2.3.1.2 Wind power source	56
2.3.1.3 Plant modelling	59
2.3.1.3.1 Electrolysis Unit (ELUNIT Hierarchy).....	60
2.3.1.3.2 Methanation Unit (METUNIT Hierarchy)	61
2.3.1.3.3 Storage Unit (STORAGE Hierarchy).....	63
2.3.1.3.4 SNG Upgrading Unit (SNGUP Hierarchy).....	63
2.3.1.4 Storage control strategy.....	63
2.3.2 Results and discussion.....	65
2.3.2.1 Performance parameters.....	65
2.3.2.2 Plants sizing and performance assessment at nominal conditions...	66
2.3.2.3 Dynamic operation and annual mass and energy balance.....	69
2.4 <i>Case study #3: Heat recovery from a PtSNG plant coupled with wind energy</i>	72
2.4.1 Material and methods	72
2.4.1.1 PtSNG Plant description	72
2.4.1.2 Heat recovery management.....	75
2.4.1.2.1 Thermal storage	76
2.4.1.2.2 Electrical storage	77
2.4.1.3 Heat recovery system sizing criteria	79
2.4.1.4 Performance parameters.....	80
2.4.2 Results and discussion.....	81
2.4.2.1 Plant mass and energy balance results	81
2.4.2.2 Thermal energy requirement	85
2.4.2.2.1 Thermal storage results.....	87

2.4.2.2.2	Electrical storage results	87
2.4.2.3	Annual performance results	90
CHAPTER 3.	ANALYSIS OF IMPACT OF HYDROGEN INJECTION NATURAL GAS NETWORKS	92
3.1	Challenges to injecting hydrogen into natural gas networks	92
3.2	Literature review on analysis of the effects of injection on natural networks	94
3.3	<i>Case study #4: Impact of hydrogen injection on natural gas networks measurement</i>	95
3.3.1	Theory and methods	95
3.3.1.1	Thermodynamic properties	95
3.3.1.2	Volume conversion factor	98
3.3.1.3	Ultrasonic static gas meter measuring principles	99
3.3.2	Results and discussion	101
3.3.2.1	Impact of H ₂ injection on the thermodynamic properties	101
3.3.2.2	Impact of H ₂ injection on the volume conversion factor	104
3.3.2.3	Impact of H ₂ injection on the ultrasonic domestic gas meters	108
3.3.3	Quality measurements of NH ₂ NG mixtures	111
CHAPTER 4.	ANALYSIS OF RELIABILITY OF THERMODYNAMIC EQUILIBRIUM GASIFICATION MODELS	113
4.1	Biomass gasification process overview	113
4.2	Biomass gasification modelling	115
4.2.1	Literature review on thermodynamic equilibrium modelling	117
4.3	<i>Case study #5: Reliability of thermodynamic equilibrium gasification models for selected biomass types and compositions</i>	120
4.3.1	Material and methods	120
4.3.2	Results and discussion	123
CONCLUSIONS	134
Appendix A	141
Appendix B	144
References	147

LIST OF FIGURES

Figure 1.1 – Schematic representation of the recovery strategy model.....	12
Figure 1.2 – Mass and energy conservation laws in network nodes and meshes [31].	17
Figure 1.3 – Simulation scheme for different NG network faults.	19
Figure 1.4 – The investigated NG distribution network of Battipaglia.	22
Figure 1.5 – Distribution of citizens.	23
Figure 1.6 – a) Pressure reduction units location in the network; b) Detail of natural gas distribution network in the city centre.	24
Figure 1.7 – Hourly trend of pressure and flowrate at standard conditions.....	25
Figure 1.8 – Average pressure trend in the 21 investigated scenarios.....	25
Figure 1.9 – Scenario id. 9468: a) Number of disrupted nodes; b) Percentage of disrupted nodes in respect of total low-pressure nodes.	27
Figure 1.10 – Number of disrupted nodes under different failure scenarios.	27
Figure 1.11 – Hourly trend of disrupted citizens under scenario 9648.....	28
Figure 2.1 – Scheme of the operating principle of an alkaline electrolysis cell [52].	34
Figure 2.2 – Scheme of the operating principle of a proton exchange membrane electrolysis cell [52].	36
Figure 2.3 – Scheme of the operating principle of a solid oxide electrolysis cell [52].	38
Figure 2.4 – Classification of methanation reactors.	41
Figure 2.5 – TREMP process scheme [60].	42
Figure 2.6 – Lurgi process scheme [60].	43
Figure 2.7 – Lurgi process as installed in Great Plains Synfuels Plant [62].....	44
Figure 2.8 – HICOM process scheme [60].	45
Figure 2.9 – RMP process scheme [60].	46
Figure 2.10 – ICI process scheme [60].	47
Figure 2.11 – VESTA process scheme.	48
Figure 2.12 – Isothermal reactor scheme [64].	50
Figure 2.13 – Linde process scheme [60].	50
Figure 2.14 – Multiple feed fluidized-bed reactor scheme [60].	51

Figure 2.15 – Fluidized bed methanation reactor by Bituminous Coal Research Inc. [60].	51
Figure 2.16 – Comflux process scheme [60].	52
Figure 2.17 – Three-phase methanation concept [60].	54
Figure 2.18 – PtSNG plant layout.	55
Figure 2.19 – Power duration and cumulative electric energy production.	57
Figure 2.20 – Distribution of the electric energy production and the duration time for different power ranges.	57
Figure 2.21 – Electric energy consumption (MWh _{AC} /year) and stored energy factor.	58
Figure 2.22 – The flowsheet of PtSNG plant integrated model.	59
Figure 2.23 – Flowsheet of PEM electrolysis unit model.	60
Figure 2.24 – Flowsheet of the methanation unit model.	62
Figure 2.25 – Flowsheet of SNG upgrading unit model.	63
Figure 2.26 – Specific energy consumption of the electrolysis module in the load range.	70
Figure 2.27 – PtSNG plant layout.	74
Figure 2.28 – Conceptual schemes and energy flows of the two proposed waste heat recovery systems: a) Thermal storage; b) Electrical storage.	76
Figure 2.29 – Schematic diagram of thermal storage system.	77
Figure 2.30 – Schematic diagram of electrical storage system.	78
Figure 2.31 – Functional scheme of the ORC.	79
Figure 2.32 – Results of the statistical analysis and distribution of the standby periods.	86
Figure 2.33 – Diathermic oil circuit for waste heat recovery from the methanation unit, sized for coupling with ORC power unit.	88
Figure 2.34 – Results for the ORC power unit	89
Figure 3.1 – Measurement principle of the ultrasonic gas meters.	100
Figure 3.2 – Trend of the thermophysical properties as a function of the H ₂ content in NG mixtures: a) relative density, b) specific heat capacity, c) higher calorific value, d) Wobbe Index.	103
Figure 3.3 – Trend of the sound speed as a function of the H ₂ content in NG mixtures.	103
Figure 3.4 – Trend of the compressibility factor as a function of the H ₂ content in NG mixtures: a) ISO 12213–2, b) ISO 12213–3, c) AGA NX 19 and d) AGA NX 19 Mod.	105

Figure 3.5 – Trend of the volumetric correction factor as a function of the H ₂ content in NG mixtures: a) ISO 12213–2, b) ISO 12213–3, c) AGA NX 19 and d) AGA NX 19 Mod.....	108
Figure 3.6 – Average errors of indications of G4 ultrasonic gas meters tested with different gas mixtures as a function of relative flow rates Q/Q _{max}	110
Figure 4.1 – Comparison of syngas mole fraction composition between results of calibrated and not-calibrated analytical and Aspen models and experimental data relating to hazelnut shell [152]: a) CH ₄ mole fraction; b) H ₂ mole fraction; c) CO mole fraction; d) CO ₂ mole fraction; e) N ₂ mole fraction.	126
Figure 4.2 – Comparison of average gasification bed temperature between the analytical model results and experimental data relating to hazelnut shell [152].	126
Figure 4.3 – Comparison of syngas mole fraction composition between results of calibrated and not-calibrated analytical and Aspen models and experimental data relating to rubber wood [153]: a) CH ₄ mole fraction; b) H ₂ mole fraction; c) CO mole fraction; d) CO ₂ mole fraction; e) N ₂ mole fraction.	127
Figure 4.4 – Comparison of syngas mole fraction composition between results of calibrated and not-calibrated analytical and Aspen models and experimental data relating to wood pellets [154]: a) CH ₄ mole fraction; b) H ₂ mole fraction; c) CO mole fraction; d) CO ₂ mole fraction; e) N ₂ mole fraction.	128
Figure 4.5 – Comparison of syngas mole fraction composition between results of analytical and Aspen models and experimental data relating to corn cobs [142]: a) CH ₄ mole fraction; b) H ₂ mole fraction; c) CO mole fraction; d) CO ₂ mole fraction; e) N ₂ mole fraction.	128
Figure 4.6 – Comparison of syngas mole fraction composition between results of calibrated and not-calibrated analytical and Aspen models and experimental data relating to rice husks [142]: a) CH ₄ mole fraction; b) H ₂ mole fraction; c) CO mole fraction; d) CO ₂ mole fraction; e) N ₂ mole fraction.	129
Figure 4.7 – Comparison of syngas mole fraction composition between results of analytical and Aspen models and experimental data relating to coniferous wood [155]: a) CH ₄ mole fraction; b) H ₂ mole fraction; c) CO mole fraction; d) CO ₂ mole fraction; e) N ₂ mole fraction.....	130
Figure 4.8 – Comparison of average gasification bed temperature between the analytical model results and experimental data relating to coniferous wood [155].	131
Figure A.1 – Flowsheet of the methanation process model.....	141
Figure A.2 – Flowchart of the storage control strategy.	143

Figure B.0.1 – Flowsheet of CO₂ methanation unit..... 144

LIST OF TABLES

Table 1.1 – Fault classification in a distribution NG network.....	15
Table 1.2 – Battipaglia’s NG distribution network data.....	21
Table 2.1 – Construction characteristics and operating parameters of electrolysis technologies [51,53].....	38
Table 2.2 – Operation parameters and gas composition of TREMP process [60].....	42
Table 2.3 – Operation parameters and gas composition at pilot plant of the Lurgi process [60].....	43
Table 2.4 – Operation parameters and gas composition of the HICOM process [60].	45
Table 2.5 – Operation parameters and gas composition of the RMP process [60]....	46
Table 2.6 – Operation parameters and gas composition of the ICI process [60].....	47
Table 2.7 – Experimental conditions for the Thyssengas pilot plant [60].....	52
Table 2.8 – Three-phase methanation reactors and operational conditions [60].	54
Table 2.9 – Common operating data of PtSNG plants.	67
Table 2.10 – Mass and energy balances under nominal conditions.....	68
Table 2.11 – Annual operation time.	70
Table 2.12 – Annual mass and energy balance.....	71
Table 2.13 – Operational data of PtSNG plant.	82
Table 2.14 – Mass and energy balances under nominal conditions.....	83
Table 2.15 – Thermal powers available from the methanation unit and SNG inlet and outlet temperatures in the heat exchangers.	84
Table 2.16 – Annual operation time of 1MW PtSNG plant.	85
Table 2.17 – Thermal energy requirements of the methanation unit.....	85
Table 2.18 – Results of the statistical analysis and distribution of the standby periods.	87
Table 2.19 – Operating conditions and performance of ORC unit.....	88
Table 2.20 – Annual mass and energy balance and annual performances.	90
Table 3.1 – Regression coefficients for chemical compound [88].	97
Table 3.2 – Characteristics of the gas mixtures used for the metrological tests.....	101
Table 3.3 – Molar composition of the investigated gas mixtures [90].	102

Table 3.4 – Molar composition of the investigated gas mixture.	104
Table 3.5 – Compressibility factor values evaluated employing the ISO 12213-2, ISO 12213-3, AGA NX 19 and AGA NX 19 Mod.	106
Table 3.6 – Average errors and average errors of indication drifts as well as WMEs and WMEs drifts of G4 ultrasonic gas meters tested with different gas mixtures. .	109
Table 3.7 – Metrological assessment of average error of indication drift of ultrasonic G4 gas meters with 2E/H5 and 2E/H0.	110
Table 4.1 – Ultimate and proximate analysis of biomass used for the validation of models.	124
Table 4.2 – Gasifier operating conditions of the experimental tests used for the validation of models.	124
Table 4.3 – Comparison of syngas mole fraction composition between results of analytical and Aspen models and experimental data relating to wood pellets [140].	129
Table 4.4 – Comparison of syngas mole fraction composition between results of analytical and Aspen models and experimental data reported in [140].	130
Table A.1 – Simulation results compared with experimental data of ADAM I.	141
Table B.0.1 – Parameters of the reaction rate by Xu and Froment [155].	146

INTRODUCTION

To move towards a low-carbon economy and to comply with commitments under the Paris Agreement, the European Union (EU) has set ambitious binding climate and energy targets for 2030. By 2050, the EU aims for 80 – 95 % of emissions reduction targets, but also declare its aspiration to meet higher ambition levels, by adopting the 2018 EU Long Term Greenhouse Gas Emissions (GHGs) reduction strategy with the view to achieve net-zero GHGs emissions by 2050, as announced in the plan for European Green Deal.

Renewable energy sources (RESs), such as wind, solar and biomass, are the keystone of the energy policy of the EU to fulfil the target of a carbon neutral economy by 2050 [1]. According to the European Directive 2018/2001 on the promotion of the use of energy from renewable sources, Member States shall collectively ensure that the share of energy from RESs in the EU's gross final consumption of energy is at least 32% by 2030 [2]. Such a percentage has been recently judged to be not sufficient to reach the goals set in the EU Long Term Strategy. Therefore, the European Commission has started to review the old regulatory packages relating to climate change. It is expected that the new renewable energy directive will set an increased target, in which 40% of the European final energy consumption should come from RESs by 2030 [3].

However, the integration of a significant share of RESs pose significant challenges to the EU energy systems as it requires, on the one hand, storing large energy volumes over weeks, months and seasons, to match intermittent renewable supply with the pattern of energy consumption and, on the other hand, transporting renewable energy from where it can be most efficiently and feasibly produced to where it is consumed, encompassing both long-distance transports.

To overcome such challenges, it will be effective to consider the reuse of the existing natural gas (NG) infrastructure. The latter will play a crucial role in the

development of a decarbonized energy system based on a large usage of RESs due to its widespread presence and its capacity to provide a cost-effective option for transporting and storing large amounts of energy for long-term period exploiting the NG transportation and distribution networks, as well as the storage complexes of the existing NG infrastructure.

NG infrastructure has historically been designed to bridge long distances between production points and consumption ones at low cost and with low energy losses. The current European NG transportation system is wide-ranging and well-established, in addition to having a transmission capacity that greatly exceeds electricity transmission grid capacity. Similarly, European NG distribution system is extensive and existing. Within the EU almost half of household end-energy consumers are connected to the gas distribution network. On EU average, the volume delivered via the distribution grid to households is around four times higher for gas than for electricity. Also, the storage capacity of European NG infrastructure is larger than electricity storage volume. Thus, due to its inherent features, NG infrastructure can help to handle the challenges of transport and storage of RESs in the short and medium term.

On the other hand, the existing NG infrastructure is suited for the transport of renewable gases, such as biomethane, hydrogen and substitute natural gas. The use of existing NG infrastructure for the injection and the transport of renewable gases, mixed or, in the future, also pure, as in the case hydrogen, can help to achieve the goals of the decarbonisation and the integration between the electricity and gas sectors, as indicated in the 2020 EU Strategy for Energy System Integration.

An interlinkage between the gas and electricity systems based on synergies between these two energy carriers (i.e., Power to gas) will provide benefits to the whole energy system, in particular:

- i. integration of RESs: maximised integration of renewable energy, especially non-programmable RESs;
- ii. flexibility: growing shares of RESs electricity generation make the electricity system exposed to weather patterns; this creates the need for back up and short- and long-term flexibility technologies working in the base load and ensuring the hourly to seasonal balance between production and

- demand; gas decarbonisation technologies can bring the required flexibility for balancing the variable input from wind and photovoltaic power generation units;
- iii. sector coupling: the electricity system and the gas system should be seen as complementary; the electric system allows for the production of large quantities of renewable energy but has challenges with regards providing long-term energy storage, handling peak production and consumption as well as facilitating long-distance transportation;
 - iv. sectoral integration: hydrogen and Power to Gas (PtG) will bring renewable energy into other sectors where substantial amounts of energy are required and where some processes are otherwise difficult or expensive to electrify;
 - v. cybersecurity: risks could be higher in an all-electricity system; the two energy systems together will offer more resilience, more time to react and better options to recover when needed;
 - vi. robust system: an integrated energy system will provide a higher level of security of supply and better integration of renewables in general, due to the capabilities on long term, seasonal gas storage and peak production and demand.

Depending on the evolution of hydrogen, biomethane and synthetic natural gas supply potential and user demand, NG infrastructure functioning will change, depending also on Member States' choice of technologies (e.g., P2G, biomethane, etc.) best serving their national needs and circumstances. The future EU energy system will have to combine the specificities of each Member State or region intending to make best use of their potential, whilst achieving the decarbonisation targets in a cost-effective manner.

In the above-described context, it is clear how NG gas infrastructure can play a central role to meet decarbonisation targets, obtain flexibility, storage options, cross-border transportation capacities and security of supply in the most efficient way. Nevertheless, to effectively promote an optimal management and an innovative use of the existing NG infrastructure, as well as the development of innovative technologies on natural gas transportation and distribution networks, the support for technologies to

mature will be needed, especially as regards the following research topic, which will be all analysed within the framework of the present research:

- i. methods and tools for the optimized management of NG transport and distribution networks;
- ii. impact of the injection of renewable gases into NG transport and distribution networks;
- iii. technologies for the production of renewable gases.

Aim of the work

The present research aims to address some of the gaps in the current scientific literature concerning the subject of innovative uses of NG transport and distribution networks, which have been described in the above introductory section. The different topics have been approached from different perspectives. For this reason, the chapters of the thesis have been compiled in a thematic way and consistently with the candidate's research experiences.

In the following, an overview about the research outline together with the thesis structure will be given in order to allow a clear understanding during the reading of the present work.

Research outline and thesis structure

The thesis is organized in four main thematic chapters, each containing an introductory section in which the scientific relevance of the research topic and the gaps in current scientific literature are highlighted.

Chapter 1 addresses the problem of NG networks management in the event of fault occurrence through the application of resilience concept. An overview about the definition of resilience for Critical Infrastructures and the state of the art of methodologies for the resilience assessment of NG networks and the analysis of a case study (*case study #1*) is provided. In *case study #1*, a novel a resilience assessment methodology for NG distribution networks able to simulate fault conditions leading to quality-of-service issues and to disservice is proposed and described. The proposed methodology is experimented in case of a real distribution network and the results is presented and discussed.

Chapter 2 is dedicated at evaluating the technical feasibility of the PtG concept to store intermittent RESs. After the description of the current state of the art of core technologies in the PtG process, two different case studies are analysed and discussed. In *case study #2*, a PtSNG plant coupled with a RES facility is designed and investigated by evaluating nominal and annual efficiencies for different sizes of the electrolysis unit. Specifically, a detailed and comprehensive simulation model for PtSNG system analysis under steady state and dynamic operation is provided and useful criteria for the PtSNG systems design and sizing are defined. In *case study #3*, a novel thermal management strategy for a PtSNG plant and how it impacts on the annual plant performances is analysed.

Chapter 3 presents an overview of the effect of hydrogen injection on NG networks. The analysis of impact of hydrogen injection on the reliability of measuring devices that currently run on the existing NG infrastructures, both from a theoretical and an experimental point of view, is proposed as *case study #4*. In detail, a comprehensive analysis of the changes of thermo-physical properties, when gases with different hydrogen content are used, is carried out, as well as of the theoretical effects on the accuracy of volume conversion at standard conditions. The impact of the hydrogen injection on the accuracy of static ultrasonic domestic gas meters is also experimentally investigated and the related results is presented and discussed. Finally, the impacts of hydrogen injection on NG quality measurements are discussed from a theoretical point of view.

Finally, *Chapter 4*, is aimed at evaluating the reliability of the equilibrium gasification models currently available in the scientific literature to establish whether they can be employed to build new control and optimization schemes and operating maps of biomass gasification systems integrated into polygenerative plants coupled with energy networks. In particular, the main objective consists in the assessment of thermodynamic equilibrium models as a function of biomass type and composition to better understand in which conditions of practical interest such models can be applied with acceptable reliability. To this aim, a thermodynamic equilibrium model using a simulation tool implemented is described (*case study #5*) and applied to the simulation

INTRODUCTION

of different biomasses gasification, employing experimental data available in the scientific literature.

CHAPTER 1. ANALYSIS OF RESILIENCE OF NATURAL GAS NETWORKS

1.1 Defining resilience for Critical Infrastructures

European Union Critical Infrastructures (CIs) include those physical resources, services, and information technology facilities, networks, and infrastructure assets which, if disrupted or destroyed, would have a serious impact on the health, safety, security or economic well-being of citizens or the effective functioning of governments [4]. CIs can be damaged, destroyed or disrupted by deliberate acts of terrorism, natural disasters, negligence, accidents or computer hacking, criminal activity, and malicious behaviour. To save the lives and property of people at risk, any disruption or manipulation of CIs should be, to the extent possible, brief, infrequent, manageable, geographically isolated, and minimally detrimental to the welfare of the state and its citizens.

It is therefore necessary that such infrastructural systems are highly reliable and safe. In other words, it is imperative that CIs are highly resilient to potential threats. In order to develop action plans that efficaciously assure the resilience of CIs, it is crucial to define the concept of resilience to make it operational and subsequently, to quantify it using a suitable metric [5].

The term resilience is derived from the Latin word *resiliens – ěntis*, the present participle of *resilire* meaning "bounce back". The concept was, for the first time in academic research, employed by Holling to denote the ability of ecological systems to absorb change and disturbance maintaining the same relationships between populations or state variables within the systems themselves [6]. Since that time, several definitions of the notion of resilience have been proposed in a wide range of disciplines ranging from environmental research to materials science and engineering, psychology, sociology, and economics [7]. Even though the consensus view is that the definition introduced by Holling have significantly influenced the common usage of the term resilience, a univocal definition applicable indiscriminately to the various

29 research areas has not proposed still now. But the notion changes according to the
30 discipline and application domain [5,8].

31 In the engineering domain, the concept of resilience implies different attributes
32 depending on the specific system analysed and the scope of analysis, as can be deduced
33 by comparing the different definitions from engineering literature, policy directives
34 and academic community.

35 Bruneau et al. [7] defined the seismic resilience of physical systems as the ability
36 to reduce the occurrence of a shock, to absorb a shock if it occurs and to recover
37 quickly after a shock, identifying four main attributes of resilient systems: i)
38 robustness; ii) rapidity; iii) resourcefulness; iv) redundancy. The US Department of
39 Homeland Security denoted the resilience as the capacity of a system or network to
40 resist, absorb, recover from, or successfully adapt to adversity or a change in
41 conditions [9]. The National Infrastructure Advisory Council (NIAC) similarly stated
42 that the effectiveness of a resilient infrastructure depends upon its ability to anticipate,
43 absorb, adapt to and/or rapidly recover from a potentially disruptive event [10]. The
44 American Society of Mechanical Engineers (ASME) defined the engineering
45 resilience as the capability of a system to sustain external and internal disruptions
46 without discontinuity of performing the system's function or if the function is
47 disconnected, to fully recover the function rapidly [11]. According to Haines [12], the
48 resilience of a system is the ability to withstand a major disruption within acceptable
49 degradation parameters and to recover within an acceptable time and composite costs
50 and risks. Vugrin et al. [13] defined the resilience of a system to a particular disruptive
51 event (or set of events) as the ability to efficiently reduce both the magnitude and
52 duration of the deviation from targeted system performance levels. Alessandri and
53 Filippini [14] regarded the resilience of network of interconnected systems as a sort of
54 structural property, more precisely the ability to resist to internal drift and cascading
55 failures, and recover back to the initial operation state. Lately, Hollnagel et al. [15]
56 introduced the notion of engineering resilience defining it as the ability of a system to
57 keep, or recover quickly to, a safe stable state, allowing it to continue operations during
58 and after a major mishap or in the presence of continuous significant stresses. Erker et
59 al. [16] presented the concept of regional energy resilience as the preservation and
60 protection of the basic functions of living (i.e., housing, working, supply and mobility)
61 within an energy region despite an immediate disruptions.

62 Based on all of the abovementioned definitions, resilience can be denoted as the
63 response of a system to a disruptive event and its behaviour throughout the response
64 time. For the infrastructure systems, the response to a disruption is represented by the

65 state of service supplied by the system. Thus, in this case, resilience describes the
66 extent of disruption across multiple dimensions including quality, time, and geography
67 of service provision. The state of service resulting from a disruptive event is
68 determined by the nature of disruptive event itself, design of the system, and mode in
69 which the system is operated. Such factors impact on the time of disruption and rate
70 of service decrease, as well as the duration of disruption and the rate and extent of
71 recovery. A key aspect of resilience emphasized by some authors is the speed of
72 recovery. In this sense, it is deemed that the increasing speed of recovery is associated
73 with a greater resilience of the system. Moreover, it can be observed that some
74 definitions, such as the one proposed by ASME [11], underline the need of system to
75 restore the performance level previous to the disruptive event; conversely, other
76 definitions do not entail that the system returns to the pre-disaster state. Lastly, note
77 that resilience as a concept is often related to other concepts as reliability, robustness,
78 recoverability, sustainability, redundancy.

79 The growing interest in the CIs resilience, both within academic field and
80 infrastructure system management, denotes that resilience is considered a promising
81 concept to better understand and improve the performance of complex and highly
82 inter-connected infrastructures that are indispensable for the functioning of our
83 modern society and the products and services it depends on. Contextually, the system's
84 complexity, the exchange of goods, services, information and people and the risk of
85 disruptive events have increased. This continues to rise calls for a methodological
86 approach that in a consistent and comprehensive manner can contribute to successfully
87 master future challenges.

88 **1.2 Title Literature review on resilience analysis of** 89 **natural gas networks**

90 The literature on resilience analysis of NG pipeline systems is not extensive. As
91 reported by Liu and Song [17], such topic has not been widely investigated and more
92 emphasis on risk and vulnerability assessment of NG pipeline network systems rather
93 than resilience one has given in the academic field.

94 Esposito et al. [18] investigated the physical impact of L'Aquila 2009 earthquake
95 on the performance of the local medium and low pressure NG distribution networks.
96 To this aim, the authors assessed the resilience of the system as the integral area of
97 functionality recovery curve defined by the number of users supplied/disrupted from
98 the time of shock to that at which the pre-event level of performance is fully recovered.

99 Ellison et al. [19] assessed the resilience of the NG network to a potential NG outages
100 in a ShakeOut scenario earthquake. The consequences to the overall infrastructure
101 from the failure of components have been analysed using a network model. Carvalho
102 et al. [20] analysed the response of the European NG pipeline system to supply
103 shortages in case of human conflict, geopolitical crises, terrorist attacks and natural
104 disasters. A resilient response strategy has been presented and its effectiveness has
105 been quantified for several scenarios. In particular, a network has been considered
106 resilient to a scenario if the vectors of country throughput per capita for the scenario
107 and the baseline scenario are similar. Cimellaro et al. [21] proposed a performance
108 assessment methodology for NG distribution networks, which includes the restoration
109 phase right after an extreme event to evaluate the resilience index of the entire network.
110 The proposed resilience index has been defined as the normalized shaded area
111 underneath the functionality curve of the system analysed, which has been determined
112 as a combination of the normalized gas flow rate and the total length of the network in
113 service before and after the disruptive event. Golará et al. [22] developed a
114 comprehensive model for NG networks resilience estimation under natural disasters
115 such as earthquakes and fires, or artificial hazards such as deliberate attacks on the
116 system. To evaluate the resilience of the NG network, an index measuring delivery
117 importance of the system has been suggested. In particular, the delivery importance
118 has been quantified through the amount of NG distributed to fixed nodes in a given
119 time period. Wadhawan et al. [23] investigated the resilience of the NG pipeline
120 systems under different cyber-physical attack scenarios using a time-related index that
121 is the time required by a network to reach the failure state and to define the system
122 performance afterward a cyber-attack. Hauser et al. [24] evaluated the resilience of
123 the northeast German NG pipeline system during critical situations. For this purpose,
124 a detailed model of pipelines, storages, border points and exit points of the system has
125 been developed and a cost minimization approach has been used to investigate demand
126 utilisation of NG infrastructure and curtailment of demand on two representative days.
127 Kyriakidis et al. [25] employed a simulation approach to quantify a resilience metric
128 of a NG distribution network and, more generally, a critical infrastructure.
129 Specifically, two recovery metrics of resilience has been examined: the first metric
130 consists in the time to recover to 100 % level of service, whereas the second one
131 consists in the integral loss of service. The results have indicated that the former is a
132 rather poor measure, and it does not represent well the integral loss of service. Su et
133 al. [26] developed a supply resilience analysis based on an integrated dynamic model
134 that has been used to evaluate the ability of NG pipeline networks to maintain stable

135 supply of gas for customers. The proposed method allows evaluating the capacities of
136 the overall system and customers to withstand disturbances and recover to an
137 equilibrium state. Marino and Zio [27] developed an original resilience analysis
138 framework for a complex NG pipeline transmission network considering the
139 cybernetic interdependence of the physical gas pipeline network with the SCADA
140 system. In this study, the resilience of the NG networks has been quantified using the
141 robustness and recoverability indices that are based on the integration of the area under
142 a resilience function between different time intervals.

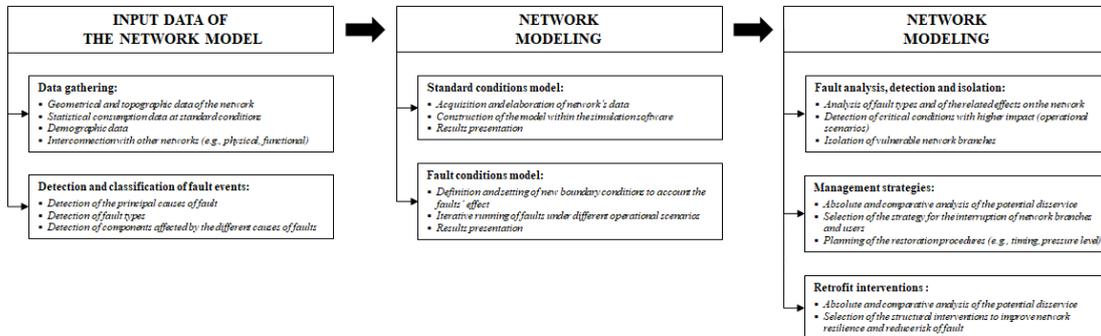
143 Based on the literature review, one of challenges integral to assess the resilience of
144 complex infrastructural systems, such as NG pipeline systems, is the development a
145 holistic methodology able to take into account multiple factors, such as geometrical
146 and topographic characteristic of the system, type of the fault, consumption scenario,
147 recovery strategy etc., that impact on the state of service resulting from an
148 unpredictable event. In the following paragraph, a resilience assessment methodology
149 for acquiring, simulating, and processing the impact of failures on NG distribution
150 networks. The aim of the proposed methodology is to evaluate the behaviour and
151 fragility of distribution networks, under different fault scenarios and network
152 structures, as well as to evaluate the fault management strategies to minimize the
153 disservice and to define possible structural improvement measures.

154 ***1.3 Case study #1: A resilience analysis simulation tool*** 155 ***for natural gas distribution networks***

156 **1.3.1 Theory and methods**

157 The proposed resilience analysis methodology is based on a software framework
158 that is able to assess the impacts of these faults on the network itself, starting from
159 possible faults that could affect the quality of service of a NG distribution network.
160 Specifically, the methodology allows to: i) evaluate the behaviour and fragility of the
161 network under different fault scenarios and assets (e.g., summer, winter, day, night);
162 ii) identify possible critical areas with end-users out of services (e.g., too high or too
163 low pressures, flow reduction, flow interruption) or danger to citizens; iii) evaluate the
164 operational recovery strategies to minimize the inefficiency necessary for the drafting
165 of intervention plans to guarantee a safer and more efficient service; iv) evaluate
166 selectively any structural improvements of the network to reduce the risk of faults and
167 to improve its resilience (e.g., increase in the redundancy of components, the meshing

168 of the network, the segmentation of the control). Figure 1.1 represents the proposed
 169 methodology.



170

171

Figure 1.1 – Schematic representation of the recovery strategy model.

172 One of the inputs of the framework is represented by the geo-referenced map of the
 173 network containing both structural (e.g., position, length and diameter of the pipes)
 174 and technical information (e.g., material of the tubes, PRPs, valves). Indeed, a NG
 175 distribution network is constituted of a number of (generally underground) pipelines
 176 that starting from physical supply and/or interconnection points allow the delivery to
 177 residential, technological, and industrial end-users. From a structural point of view,
 178 NG distribution networks mainly consist of a series of City Gates plants (i.e., pressure
 179 regulation and measurement at the border with the transport network), highly branched
 180 distributed pipelines connecting adjacent nodes and supplying end-users, PRPs,
 181 redelivery points (RDPs, i.e., the physical network points, at which the distribution
 182 company redelivers gas to the user, and where gas is metered). In particular, the nodes
 183 represent: i) entry points where the gas enters into the system (City Gates); ii) points
 184 where the gas is delivered to end-users (RDPs); iii) points where a pipeline undergoes
 185 a change in diameter, or it is split into two or more branches. An important source of
 186 data of the framework is represented by the network GIS maps developed by the
 187 distribution operators as they use GIS maps for planning and to analyse the risk of
 188 actions (for example, laying underground cables, excavations and earthmoving) that
 189 can affect the functionality of the network [28]. However, GIS maps almost always
 190 represent exclusively the underground medium/low pressure distribution networks.
 191 Whereas, the last section of the above-ground redelivery network, on which the supply
 192 meters of end-users are installed, it is not represented in GIS tools. Consequently, the
 193 terminal nodes traced on the GIS maps represent groups of redelivery points (i.e., they
 194 represent group of end users) and, generally, they do not refer to single RDP except
 195 for large users. Therefore, one of the most critical points of the network modelling

196 activity is represented by the accurate representation of end-users and citizens served
197 by individual nodes. For this purpose, to measure the impact of a failure on citizens
198 served at each node of the network, an updated demographic model, developed in Italy
199 by ISTAT (Italian National Statistics Institute), was associated with the network GIS
200 Model, by overlapping the maps. Another relevant input for the simulation framework
201 is represented by the knowledge of load profiles in terms of distributed NG in the
202 network terminal nodes, which reliably describes the real end-user's consumption
203 trends. In general, it is possible to identify different kind of end users. Residential end
204 users use NG for heating/cooling, cooking and domestic hot water production (often
205 not all services at the same time) and the relative consumptions are generally measured
206 through traditional mechanical gas meters [29]. Commercial, technological, and larger
207 residential users (e.g., buildings with central heating systems) consume greater NG
208 volumes and they are often equipped with larger smart gas meter. Generally, they are
209 hybrid mechanical gas meters, static ultrasonic or capillary thermal mass meters,
210 which are becoming increasingly spread in urban distribution networks [30]. In fact,
211 in Italy, the Regulatory Authority for Energy Networks and the Environment
212 (ARERA) in 2008 launched a roll-out program setting the obligation to gradually
213 replace the old meters in the gas distribution networks with smart meters. To date, the
214 installation of new smart meters of a higher size than G10 has been completed, whereas
215 the use of residential G4 smart meters is not yet spread especially in medium and small
216 networks. The loads of the network can therefore be obtained from the statistical data
217 obtained from the smart meters (when installed) or, alternatively for non-remote users,
218 from forecast models used by the operators for the NG balance sheet. These forecasting
219 models are generally based on: i) at least 2-m readings per year to characterize the
220 intensity of user consumption; ii) climatic conditions (e.g., average temperature, wind
221 speed, solar radiation); and iii) a classification of end-users categories. These models
222 are based on real historical consumption data which allow to set standard consumption
223 profiles associated with different end-user categories, as a function of the intended use
224 of the gas (e.g., heating, cooling, cooking, technological or commercial use) and of the
225 expected annual consumptions. Standard load profiles can be obtained through
226 regression analysis (linear, exponential or polynomial) for which regressors are
227 represented by climatic or dichotomous parameters capable, as for example, to
228 consider differences between working days and holidays or through the use of sigmoid
229 functions whose shape fits to the typical gas consumption trend due to climatic
230 conditions [31,32]. Therefore, the estimation of daily consumptions of single users is
231 obtained per Equation (1.1).

$$NDMDC = CV \cdot SLP_d \cdot CF_d \quad (1.1)$$

232 In which CF_d is a corrective factor taking into account different parameters
 233 affecting gas consumptions (i.e., climatic condition, holiday), SLP_d is the standard
 234 consumption profile and CV is the customer value depending on the number of meter
 235 readings performed by the Distribution System Operator (DSO) at the single end user's
 236 premise (L_1 and L_n) according to Eq. (1.2).

$$CV = \frac{L_1 - L_n}{\sum_{d=1}^{365} SLP_d} \quad (1.2)$$

237 Although these methods are designed to be used for the NGN monthly balancing,
 238 their accuracy is greater when longer periods are considered. Furthermore, the forecast
 239 reliability increases significantly as a function of the available number of readings.

240 Generally, the available consumption forecast models allow obtaining quite reliable
 241 average daily load profiles, but not the hourly ones. To overcome this limitation, actual
 242 consumption measured by smart meter has been integrated with consumption
 243 prediction models. In this way, starting from the hourly data of the NG entering the
 244 network through the city gates and outgoing at the redelivery points (technological and
 245 residential users) measured with smart meters, it was possible to estimate the global
 246 hourly consumption at the redelivery points not measured with smart meter (average
 247 hourly value). This latter is obtained by calculating the difference between the NG
 248 distribution network (measured at the City Gates) and the sum of the hourly
 249 consumption of the largest commercial and industrial users equipped with smart
 250 meters.

251 Moreover, with the increasing development of smart grids it is essential to consider
 252 the interconnections of the gas network with other networks. In fact, the more NG
 253 distribution networks are interconnected with other networks (such as those of NG
 254 transmission, electricity, district heating, etc.), the more a failure in a network can
 255 propagate to the interconnected ones increasing the impact on end-users. Therefore,
 256 interdependency (i.e., interconnection and interaction) between networks should be
 257 carefully localized, qualified, and included in the modelling framework. The NG
 258 transmission networks are interconnected with the distribution networks through the
 259 regulation and measurement systems (the so-called City Gates) in which the main input
 260 parameters are represented by entering flowrates, pressure and temperature, gas quality
 261 and odorization. In addition, the NG transmission network can be interconnected with
 262 electricity networks through plants for the production of electricity from NG and/or

263 Power to Gas plants. Given the spread of district heating in cities, the interconnection
 264 of natural gas distribution networks with district heating networks is also highly
 265 critical. The district heating plant, generally located in a central position with respect
 266 to the users served, can be considered a technological-industrial delivery point
 267 connected to the distribution network and, due to the high consumption and the non-
 268 interruptible nature of the service, it represents a critical point to ensure adequate flow
 269 rates and pressure levels.

270 A key element of the proposed resilience assessment methodology is the
 271 classification of the faults that may occur in the NG distribution networks and/or in
 272 the interconnected networks and related potential impacts. Table 1.1 shows some of
 273 the main causes of failure and the related impacts, divided by type (e.g., failures from
 274 interactions with other networks, intrinsic failures of the distribution network, natural
 275 events and damages of third parties). The classification of Table 1.1 has been produced
 276 in collaboration with the gas distribution operator of the city of Salerno that shared
 277 with the author their experience and know-how. Anyway, the classification can be
 278 refined to reflect specific requirements of each single operator.

279 This analysis allows the network manager to partially disengage from the
 280 experiences of individual technicians and make this knowledge a collective strength
 281 of the company. For each event, the Network Operator may also report event statistics,
 282 analysis of incorrect behaviour, analysis of impacts, average recovery times, etc. These
 283 elements would allow the training of new technicians and to develop management
 284 strategies while maintaining a greater historical memory of the specificity and
 285 criticality of the network itself.

286 *Table 1.1 – Fault classification in a distribution NG network.*

Type	Fault	Impact on distribution NG pipeline system
External interconnected network failure	Technical failure: - NG transmission network failure - Electric breakdown - District heating network failure - ICT network failure	- Service interruption - Deliveries irregularities
Internal distribution network failure	Pipeline faults: - Defects and Cracks - External corrosion - Internal corrosion - Stress Corrosion Cracking Systems and components faults: - Failure of Valve and PRPs - Failure of control and discharge devices - Failures of Filters - Failure of Odorization system	- Low pressure - High pressure - Deliveries irregularities - Gas losses - Carbon monoxide generation - Fire - Explosion - Service interruption

	<ul style="list-style-type: none"> - Failures of ICT systems Management Fault: <ul style="list-style-type: none"> - Operative and maintenance procedures - Human fault 	
Natural failure cause	Extreme natural events: <ul style="list-style-type: none"> - Hydro-geological events (landslides, earthquakes, erosion, etc.) - Extreme meteorological events (floods, hurricanes, thunderbolts, ...) 	<ul style="list-style-type: none"> - Fire - Explosion - Service interruption
Third-party damage	Accidental events: <ul style="list-style-type: none"> - Overloads - Accidents (fire, explosion) - Third party damages - Intentional damage - Targeted attacks and sabotages - Cyber attacks 	<ul style="list-style-type: none"> - Fire - Explosion - Service interruption

287 Then, the next phase of the proposed framework is the network modelling phase.
288 For the modelling of the network pressure, the energy conservation law is used,
289 assuming isothermal conditions, as per Eq. (1.3):

$$\frac{dP}{dx} + \frac{\lambda}{d\rho} \frac{w|w|}{2} + \rho \cdot g \cdot \sin \alpha = 0 \quad (1.3)$$

290 where: i) P is the pressure (bar), λ is the pipe friction factor (dimensionless), d is the
291 inner diameter of the pipe (m), w is the gas speed ($\text{m}\cdot\text{s}^{-1}$), g is the gravity ($\text{m}\cdot\text{s}^{-2}$), ρ is
292 the gas density ($\text{kg}\cdot\text{m}^{-3}$), α is the angle of inclination ($^\circ$).

293 In the technical literature there are several fluid dynamic models for the calculation
294 of λ in closed conduits [32]. For the case study described in the next paragraph, the
295 author used Hagen-Poiseuille's model in the case of laminar flow and Colebrook-
296 White's model in the case of turbulent flow, as described in Eqs. (1.4) and (1.5),
297 respectively.

$$\lambda = \frac{64}{Re} \quad (1.4)$$

$$\frac{1}{\sqrt{\lambda}} = -2 \log \left(\frac{K_S}{3,71 \cdot d} + \frac{2,51}{Re} \frac{1}{\sqrt{\lambda}} \right) \quad (1.5)$$

298 Where Re is the Reynolds number (dimensionless) and K_S is the sand roughness
299 (mm).

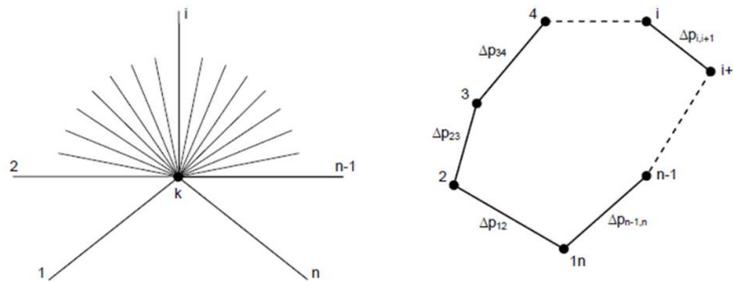
300 For the estimation of the pressure at the nodes and of the flow-rates through the
301 pipeline stretches, the principles of mass and energy conservation are used, that is the

302 first and second Kirchhoff's law [33], as represented in Figure 1.2 and in Eqs. (1.6)
 303 and (1.7).

$$\sum_{i=1}^N Q_{i,k} = 0 \quad (1.6)$$

$$\sum_{i=1}^{N-1} \Delta P_{i,i-1} = 0 \quad (1.7)$$

304 where: $Q_{i,k}$ is the flowrate entering or leaving the K -th node, $\Delta P_{i,i-1}$ is the pressure
 305 variation in the stretch between nodes i and $i+1$.



306
 307

Figure 1.2 – Mass and energy conservation laws in network nodes and meshes [34].

308 The solution of the above-described system of equations is carried out by the PSS-
 309 Sincal commercial simulator through the Hardy Cross iterative method of flow-rates
 310 balancing. This latter is a method widely used in the resolution of optimization
 311 problems concerning looped pipeline networks [35], the application of which requires
 312 the definition of an initial flow distribution within the pipelines, capable to meet the
 313 first Kirchhoff law in each node. Kirchhoff's second law for each closed loop must be
 314 meet at the end of the iterative calculation process. To this aim, the choice of the initial
 315 flow distribution does not have a significant influence on the convergence properties
 316 of the method [36]. At each iteration, in fact, the Hardy Cross method does not directly
 317 calculate the flow through the pipe segments but calculates a correction to be added
 318 algebraically to the flow obtained in the previous iteration. Since the sign of the flow
 319 determines the direction the gas travels through the pipeline, this algebraic sum
 320 operation requires the application of complex rules that represent the major flaw of
 321 this method. In fact, while accurate results even on meshed networks are expected, due
 322 to the complexity of the rules necessary for defining the signs of the corrections to be
 323 applied to the flows at each iteration, this method may require a not negligible number
 324 of iterations in the case of very large networks. There are methods in the literature

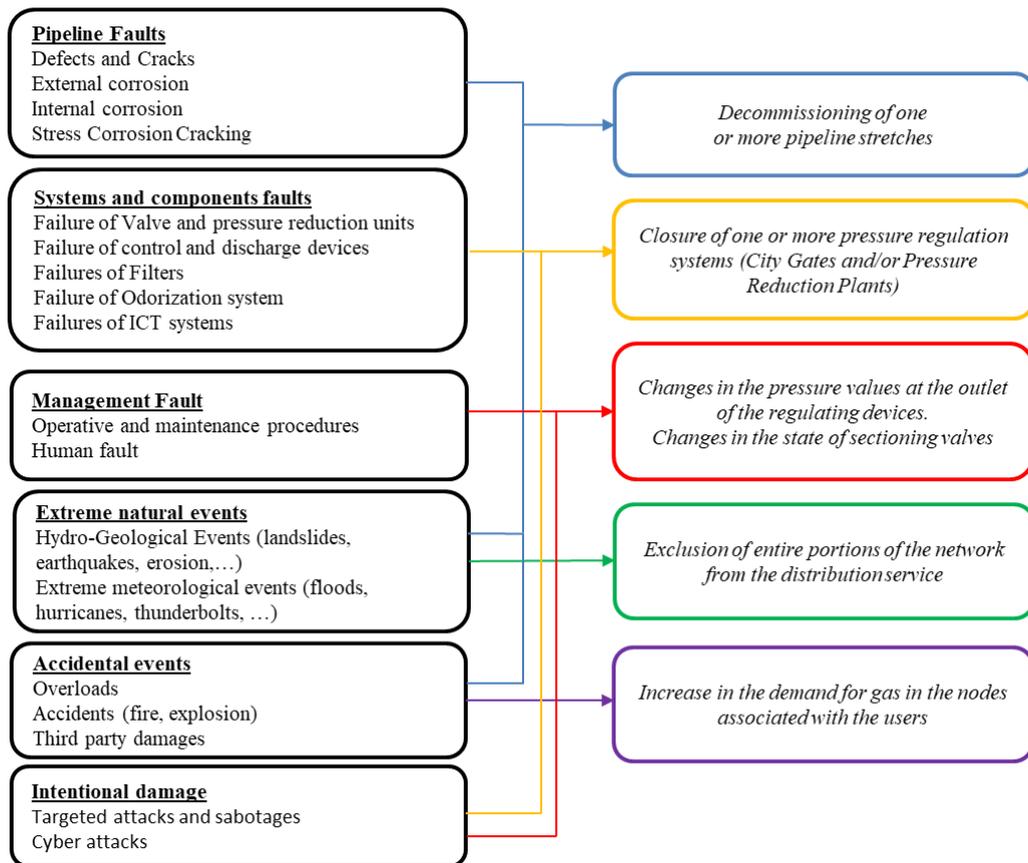
325 which, by directly calculating the flows at each iteration, show the advantage of
326 reducing the computational complexity of the problem [37]. In this work, once the gas
327 demand at each node has been set, which also determines the quantity introduced into
328 the network through the city gates, flows and pressures in the entire network have been
329 calculated through the system of equations just described. Thus, the uniqueness of the
330 obtained solution is guaranteed by the simultaneous application of the conservation
331 equation of mass and energy (i.e., Kirchhoff's laws), in fact, although in theory there
332 can be an indefinite number of flow distributions satisfying the conservation of mass,
333 there is a unique solution which simultaneously satisfies the conservation of energy in
334 each grid network [38].

335 The analysis performed is related to steady regime before and after the simulated
336 fault. Furthermore, although this method is strictly applicable to not compressible
337 fluids, in the investigated case study of natural gas distribution networks, pressure drop
338 in pipes is minor and thus the Hardy Cross method can be used without significant
339 errors [38]. Therefore, by applying the principle of energy and mass conservation in
340 the nodes and branches of the network, flow rates and pressures are obtained at the
341 relevant points of the network under standard operating conditions. On the other hand,
342 in the presence of network failures, variations in supply pressure, extensive or
343 localized inefficiencies are observed which affect the distribution service quality and
344 put the safety of users of the distribution service and citizens at risk. When a network
345 failure occurs, pressure drop in some nodes become likely. Consequently, if the
346 demand is kept constant, the nodes whose pressure is below a threshold limit (for
347 example 15 mbar) can be considered out of service because, due to the modified
348 network structure, it is not possible to safely supply the required flow at the specific
349 nodes. In this condition, the effectiveness of the distribution service is not guaranteed,
350 and the quality of the service will deviate from the reference level. Thus, adequate
351 recovery strategies must be promptly defined and adopted by the Network Operator.

352 In order to model the operation of the network under fault conditions summarized
353 in Table 1.1, corresponding simulation scenarios are developed (see Figure 1.3), by
354 means of a simulation tool for modelling the physical behaviour of the network.

355 Two approaches are currently available for simulating a gas distribution network:
356 i) pressure-driven approach and ii) demand-driven approach (which is the most widely
357 used). In this latter case, the model input is represented by the load profiles in each
358 node, while the simulated pressures in the nodes and the flowrates in single branches
359 represent the output. To allow the results of the numerical modelling to be used
360 effectively, it is necessary to post-process the output data according to the simulation

361 approach used. In the case of a demand driven approach, the simulation returns the
 362 values of the nodal pressures ensuring that imposed hourly demand of gas does not
 363 vary. On the other hand, in the case of pressure driven approach, the simulation returns
 364 the flowrates at the nodes, calculated as a function of the network pressure, starting
 365 from the input hourly demand.



366

367

Figure 1.3 – Simulation scheme for different NG network faults.

368 The abovementioned approaches allow assessing the network efficiency by varying
 369 physical parameters such as pressures and flowrates at specific points, and then
 370 calculating the corresponding quality level of the service. By changing the status of
 371 PRPs or of the pipelines in the model, it is therefore possible to simulate the operating
 372 mode under different faults, critical overload conditions or irregularity of the service.
 373 The proposed methodology does not allow simulating fires or explosions caused by
 374 gas leaks from the network, for example due to damage by third parties or natural
 375 events (e.g., earthquakes, landslides). In this case, the historical data related to fires
 376 and explosions recorded in the natural gas transmission and distribution networks [39–
 377 42] can be generally used, together with the safety concerns for users and citizens, as

378 well as that of network integrity. Starting from such knowledge, it is possible to
379 associate the corresponding occurrence frequencies to the types of accidents and to
380 assess the associated individual, social and economic risk [43,44].

381 The output of the resilience assessment strategy proposed here includes: a)
382 assessment of the resilience of the network with respect to the typical standard
383 operational conditions at different assets; b) identification of nodes and critical
384 stretches both for the quality of the service and for the safety of users and citizens.
385 Consequently, appropriate recovery operational strategies can be defined together with
386 prompt intervention plans, possibly also by SCADA systems for remote control and
387 command. The aforementioned strategies must also rely on: i) the identification of
388 structural adjustment interventions in the network to reduce the risk of breakdowns
389 and improve the resilience of the network itself (e.g., increase in the redundancy of the
390 components, meshing of the network, control segmentation), ii) scheduling of
391 predictive maintenance operations that require the interruption of network segments
392 or the temporary closure of pressure regulation devices.

393 **1.3.2 Application of the resilience analysis simulation tool to** 394 **a real natural gas distribution network**

395 The analysed case study is represented by an urban distribution network in Southern
396 Italy consisting of approximately 200 km of pipelines and supplying approximately
397 50,000 inhabitants in an area of about 57 km². The network is connected to the national
398 transmission network through two city gates where the gas is measured, odorized and
399 expanded up to about 5 bar, characteristic of the medium pressure branch of the
400 network (about 42% in length). A further reduction is subsequently performed from
401 the medium to the low-pressure network (which represents the remaining 58% in
402 network length) in 31 PRPs. The distribution network presents 17,096 redelivery
403 points (of which 48 industrial users) connected to 9372 nodes (of which 8195 in low
404 pressure).

405 Redelivery points include 76 users equipped with smart meters allowing a daily
406 metered (DM) analysis of consumptions. Indeed, these users are commonly defined as
407 daily metered since the daily consumption obtained by summing the hourly
408 measurements recorded by the meter is available for them. Furthermore, hourly
409 consumption data are also available, and these latter are useful for defining the demand
410 diagram which is one of the inputs of the simulations. Even though DM users are a
411 small percentage of the total redelivery points, they represent approximately 50% of

412 the consumed volume, since they are mainly technological users (i.e., small and
 413 medium industries and public offices also including few apartment buildings equipped
 414 with centralized heating system). In detail, 48 DM users are small and medium
 415 industrial users and the remaining 28 are public offices and apartment buildings
 416 equipped with centralized heating system (domestic use). An overview of the analysed
 417 urban network is given in Figure 1.4 and Table 1.2. In particular, Figure 1.5 shows a
 418 plan representation of the urban area supplied by the network.

419 For this study, three sets of data have been used: i) the geometrical characteristics
 420 of the network (i.e., diameters and length of single branches, reciprocal positions of
 421 the components); ii) total natural gas entering the network in a year on an hour basis;
 422 iii) DM readings (from smart gas meters) in a year on a daily basis; iv) DM readings
 423 (from smart gas meters) on hourly basis. The investigated distribution network was
 424 entirely transposed in the GIS environment, digitizing, and defining the information of
 425 all the physical elements of the system. Such data was imported into the PSS-SINCAL
 426 simulation software.

427

Table 1.2 – Battipaglia’s NG distribution network data.

NG distribution network data	
Low pressure network length, [km]	110
Medium pressure network length, [km]	80
Number of PRPs	31
Number of City Gates	2
Total number of Redelivery Points	17096
Number of residential customers	17048
Number of industrial users (Daily metered)	48
Number of supplied citizens	50464
Number of public offices and apartment buildings (Daily metered)	28

428 In order to define the required input loads for the numerical simulation, consisting
 429 of the hourly gas demand for 24 consecutive hours, the author built the hourly load
 430 diagram for each DM user equipped with Smart Gas Meter. To this aim, twenty-four
 431 continuous hours were simulated, obtaining the hourly pressure trend.

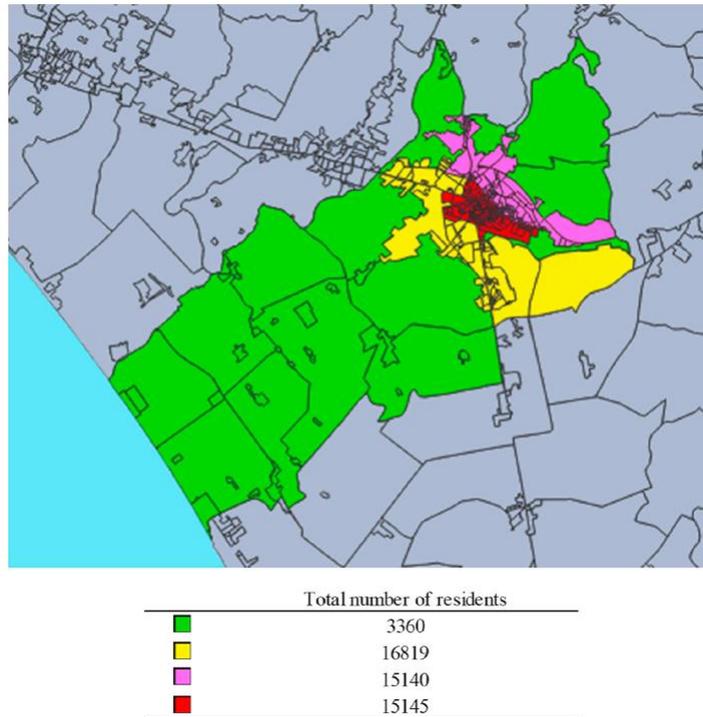


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Figure 1.4 – The investigated NG distribution network of Battipaglia.

434 Furthermore, aimed at assessing the effect of a fault in the network in terms of
435 disrupted citizens, author associated to single redelivery points the corresponding
436 average number of supplied citizens. This has been obtained by disaggregating the
437 available data from the 15th ISTAT population census at sub municipal level. Figure
438 1.5 shows the distribution of citizens in four sub urban areas in which the modelled
439 network is divided. Through the open-source software QGIS, these data have been
440 integrated with the node position. It can be also pointed out that ISTAT and single
441 municipalities identify homogeneous portions of territory in terms of demographic and
442 social characteristics. Therefore, assuming that the population is uniformly distributed
443 in these areas and the location of the citizens does not change over time during the day,
444 the approximate number of citizens can be attributed to each node. Furthermore, this
445 was aimed at enhancing the accuracy of the model, since the DSO provided
446 information only about the number of redelivery points.



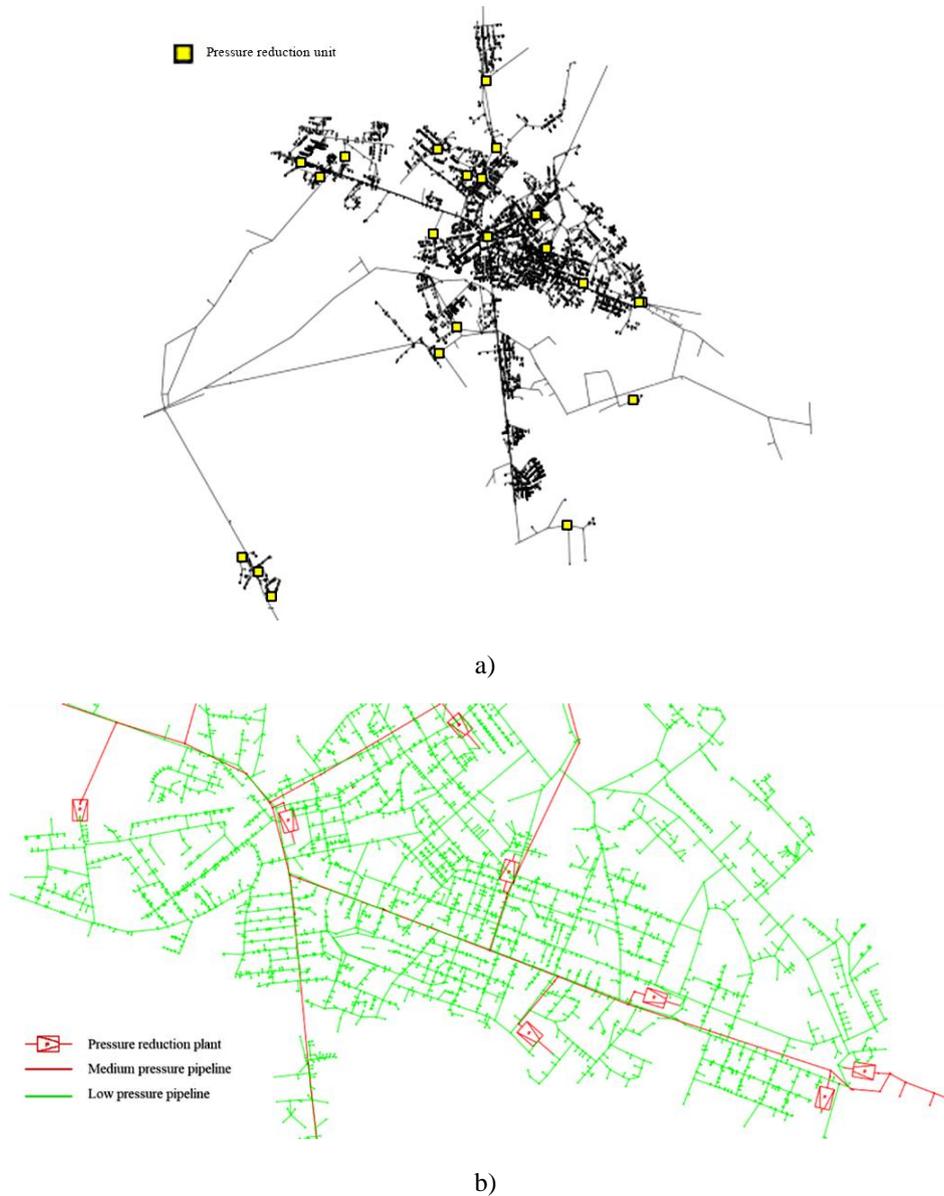
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Figure 1.5 – Distribution of citizens.

449 1.3.3 Results and discussion

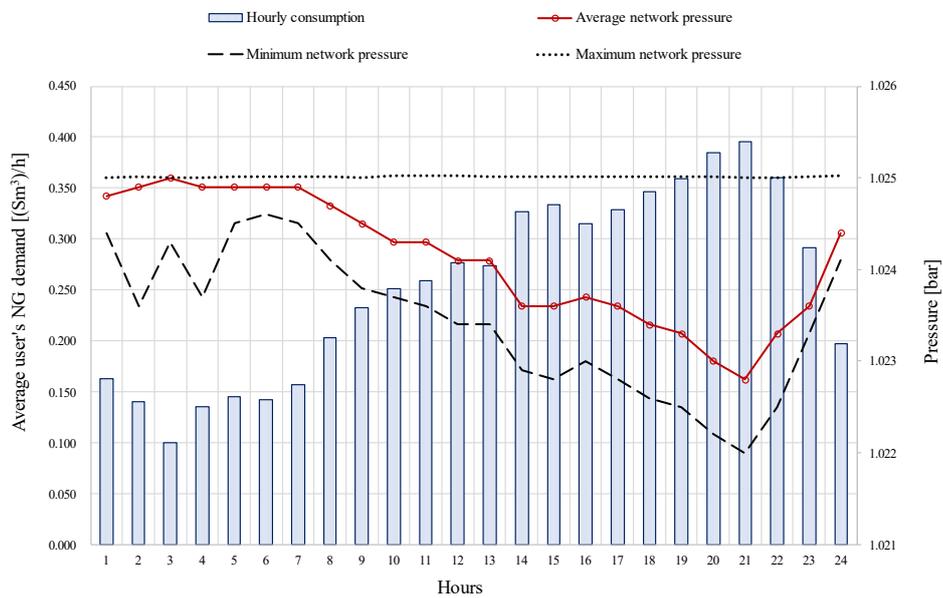
450 This section presents and discusses the results of the proposed network simulation
 451 framework. In particular, the analysis considers the load profiles related to the winter
 452 period that is crucial in terms of gas demand and simulates 24 h. The different
 453 simulation scenarios are obtained by considering 21 significant PRPs, identified
 454 according to their position in the network (see Figure 1.6), which in turn are set out of
 455 service for the whole simulation. The choice of the 21 meaningful PRPs was made by
 456 excluding the reduction groups dedicated to single users or to small groups of users
 457 located in peripheral areas of the network at medium pressure. These latter, in fact,
 458 represent trivial simulation scenarios since, the out of service of a PRP feeding a small
 459 number of isolated users, regardless of the surrounding conditions (i.e., hourly gas
 460 demand), always leads to the disruption of only this number of users. PRPs feeding
 461 isolated areas of the network have been considered as trivial cases since the number of
 462 disrupted customers is known a priori.



463 *Figure 1.6 – a) Pressure reduction units location in the network; b) Detail of natural gas distribution*
 464 *network in the city centre.*

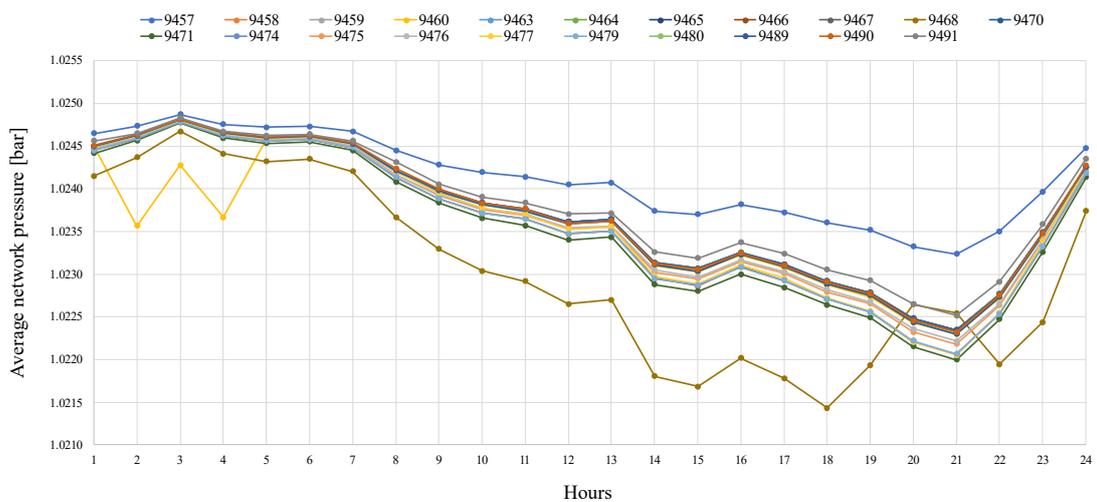
465 Figure 1.7 shows the trend of average, minimum and maximum pressure at
 466 redelivery points as the gas demand changes under standard operating conditions. It is
 467 clear that as the volumes consumed in the network increase, the average (minimum)
 468 pressure is normally between 1.0248 (1.0245) bar and 1.0228 (1.0220) bar.
 469 Furthermore, both the average pressure and the minimum tend to decrease in
 470 correspondence with the maximum absorption hours (evening operating hours) when
 471 space heating consumptions become more relevant and can lead to greater criticalities.

472 Figure 1.8 shows the average pressure of all nodes of the network during the 24 h
 473 when a single PRP is out of order. The graph shows that in the different scenarios, the
 474 average pressure ranges from 1.0228 to 1.0250 bar. In almost all scenarios, the
 475 dispersion is smaller during the first hours of the day (with limited hourly gas demand),
 476 whereas it increases as consumption increase. In general, the average hourly pressure
 477 lowers in all the analysed scenarios. In particular, the heaviest operating conditions
 478 have been recorded for the scenario in which PRP 9468 is faulty. This can be ascribed
 479 to a localized pressure drop, downstream of the PRP under failure.



480
 481

Figure 1.7 – Hourly trend of pressure and flowrate at standard conditions.



482
 483

Figure 1.8 – Average pressure trend in the 21 investigated scenarios.

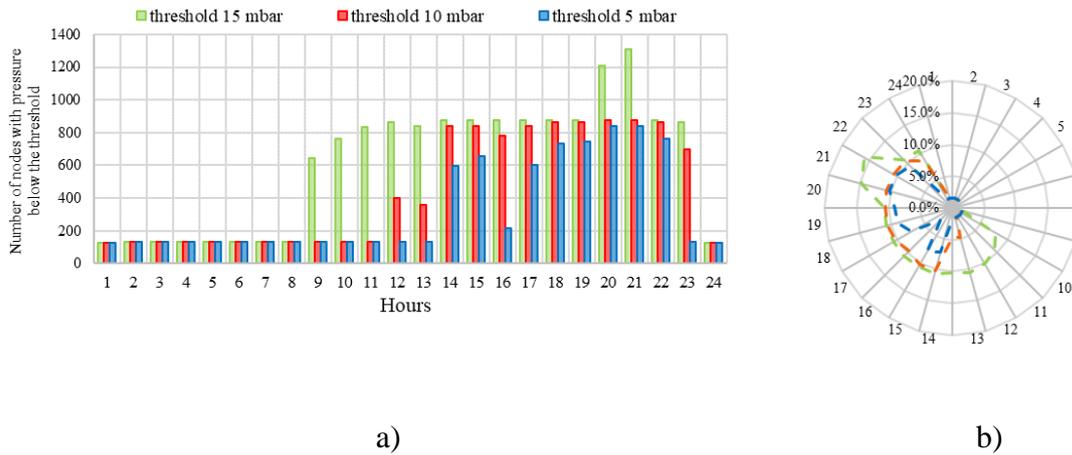
484 Whenever the pressure is not sufficient to grant the gas demand above a minimum
485 threshold, the redelivery points connected to the aforementioned nodes can be
486 considered out of service. As for example, Figure 1.9 shows the number of disrupted
487 nodes (i.e., presenting pressure level lower than the threshold respectively set equal to
488 15, 10 and 5 mbar) for the more critical failure scenario (id. 9468).

489 The most interesting results are those obtained with the threshold value of 15 mbar,
490 that is the value typically adopted by the DSO to guarantee the downstream gas
491 demand. In this case, it was observed that the number of disrupted nodes varies from
492 130 (i.e., 1.6% of the network nodes) at 8 a.m. up to 1300 nodes (i.e., 16% of nodes of
493 the network) at 9 p.m. when the peak of the gas demand occurs. Therefore, due to the
494 dependence of the pressure on the volumes withdrawn from the network, the hours
495 with higher gas demand are also the most critical for the level of service disruption
496 observed under failure. Therefore, prior and timely knowledge of the temporal
497 evolution of the failure in the network allows DSOs to guarantee the minimum level
498 of distribution service even during the emergency phase (that immediately follows the
499 failures of PRPs). DSOs may also schedule maintenance interventions involving
500 temporary out of service of a device by choosing a time slot that guarantees the lowest
501 disservice, thus limiting service interruptions.

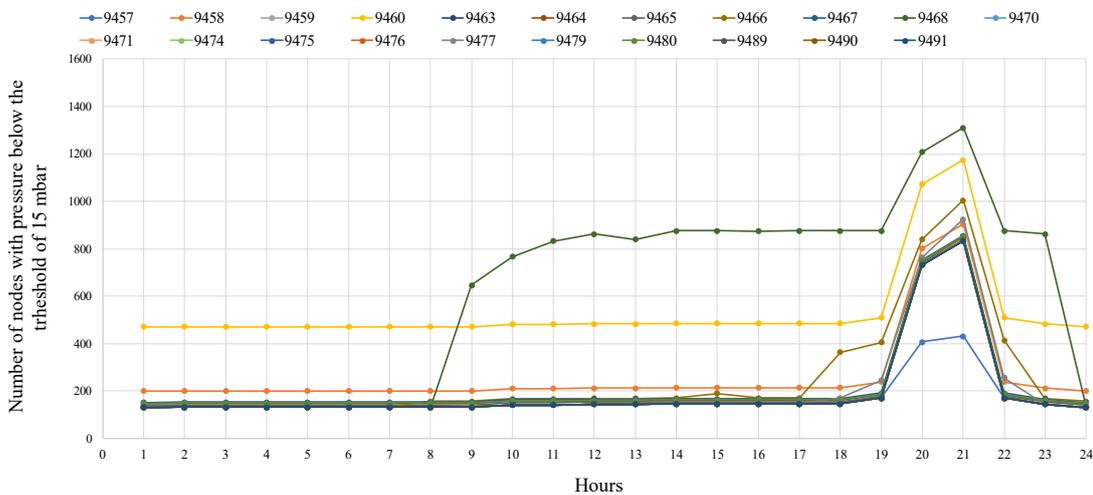
502 The results obtained with the threshold of 15 mbar are summarized in Figure 1.10.
503 In particular, the graph allows to identify the most critical scenarios (e.g., 9468) and,
504 consequently, to obtain useful information to plan structural network interventions to
505 increase its resilience. The dispersion of the number of nodes out of service in the
506 different simulated scenarios is due both to the position of PRPs and to the simulated
507 flow rates. Indeed, the most meshed sections of the network are less affected by the
508 failure (see scenario 9457). For a more thorough analysis of the effects of network
509 failure, the author also estimated the number of citizens not supplied under failure
510 conditions. In Figure 1.10 the hourly trend of disrupted citizens under scenario 9648
511 is reported. It can be pointed out that, when the demand increases (i.e., between 8 a.m.
512 and 9 a.m. see Figure 1.10), the number of citizens not supplied is greater than 5000
513 (about 10% of the total citizens). In the absence of restoration interventions, this figure
514 increases to a maximum of 10750 disrupted citizens (i.e., 21% of the total) between 8
515 p.m. and 9 p.m. and only after 11 p.m. the percentage of disrupted citizens will be less
516 than 2%.

517 The simulation results show how the DSO must adopt prompt countermeasures to
518 recover the failure. If the DSO fails to adopt proper recovery procedures and the
519 number of disrupted citizens becomes not acceptable, it will be necessary to limit the

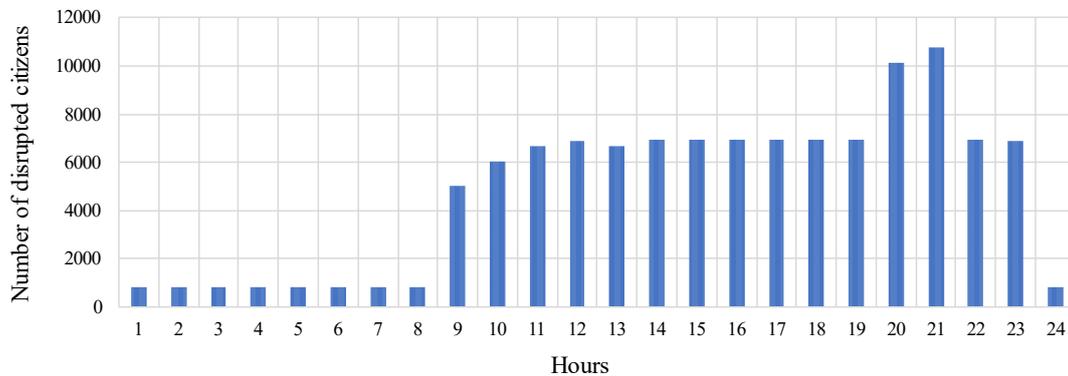
520 disservice by selectively reducing the gas demand of users. Indeed, the pressure of the
 521 network strongly depends on the gas demand (i.e., delivered volumes). To this end,
 522 different options for reducing the gas demand have been investigated by the author: i)
 523 10% (scenario # 1); ii) 15% (scenario # 2), iii) 20% (scenario # 3); iv) 25% (scenario
 524 # 4); v) 30% (scenario # 5); vi) 35% (scenario # 6); vii) 40% (scenario # 7).



525 *Figure 1.9 – Scenario id. 9468: a) Number of disrupted nodes; b) Percentage of disrupted nodes in*
 526 *respect of total low-pressure nodes.*



527
 528 *Figure 1.10 – Number of disrupted nodes under different failure scenarios.*



529

530

Figure 1.11 – Hourly trend of disrupted citizens under scenario 9648.

531 Figure 1.11 shows the results in terms of disrupted nodes, for each hour, in the
 532 different scenarios of simulated gas demand reduction. As for example, reducing the
 533 gas demand of 40%, the disservice in the critical hours (i.e., in the evening) is very
 534 limited. In particular, a 10% reduction in demand allows the disservice to be restored
 535 by 10 a.m. In the investigated case study, a homogeneous consumption reduction has
 536 been applied to all users in the network, which is a common emergency practice
 537 adopted by DSOs in presence of faults. The fact that there are no changes in the level
 538 of disservice following the imposed consumption reduction (i.e., hour 13 reduced from
 539 10 to 15% or hour 18 reduced from 20 to 25%) can be attributed to the presence, in the
 540 neighbourhood, of a particularly energy-intensive user that initially suffers little from
 541 the effects of a widespread reduction. In this case, an appropriate emergency
 542 management plan would allow to reduce consumption on few specific users. On the
 543 other hand, the decrease of consumption can be programmed taking into account the
 544 user typology, intervening selectively and guaranteeing the service to the users that
 545 require greater protection (e.g., hospitals, medical clinics, hospices). DSO could
 546 therefore identify users whose closure would lead to lower inconvenience and plan the
 547 recovery of the fault or define alternative solutions to meet the peak demand. For this
 548 purpose, the greater diffusion of smart meters (which in Italy are mandatorily equipped
 549 with a remotely controlled closure valve) will make it possible to identify the most
 550 energy-consuming interruptible users to contain the instantaneous gas demand thus
 551 limiting the level of disservice to other users and citizens.

552

553 **CHAPTER 2. ANALYSIS OF POWER TO** 554 **GAS TECHNOLOGY**

555 **2.1 Power to Gas concept**

556 Power to Gas (PtG) is one of the most promising storage technologies to support
557 RESs integration in the energy system [45]. Exploiting the NG transmission and
558 distribution networks, as well as the storage features of the existing NG infrastructure,
559 the synthetic gas (i.e., hydrogen or substitute natural gas) produced from surplus
560 renewable energy can be injected into the NG pipeline system, allowing to store large
561 amounts of energy for long-term periods, thus offering balancing and regulation
562 services to the electricity grid.

563 PtG concept involves the conversion of renewable electricity into gaseous hydrogen
564 through electrolysis process (Power to Hydrogen – PtH₂), or into substitute natural gas
565 (Power to Substitute Natural Gas – PtSNG) through CO₂ methanation process. The
566 hydrogen produced can be used locally in industrial processes, transported by a grid
567 for mobility applications, converted into methane or again into power by feeding fuel
568 cell systems, or can be stored. Hydrogen can be also injected directly into NG pipeline
569 systems, nevertheless certain limitations peculiar to the physical and chemical
570 properties of H₂ prevent it from being injected in the existing NG infrastructure in
571 larger proportions. Material durability, leakage and safety are some of the concerns
572 that hinder blending of higher percentage of H₂ in existing NG pipeline systems.
573 Moreover, the blending of H₂ also affects the quality of NG in terms of density, heat
574 value and Wobbe Index. Currently, H₂ can be injected into NG pipeline systems within
575 certain percentages varying between 0.1% and 12% vol. depending on national
576 standards [46,47]. Such limitations make the solution of PtSNG an enabling
577 technology toward the energy transition in the short and medium term, particularly in
578 Europe, where the NG infrastructure is extended and capillary. Besides, H₂ reacting
579 with carbon dioxide to form synthetic methane through the Sabatier process gives an
580 opportunity to valorise unavoidable CO₂ emissions. The gas mixture thus produced,
581 so called substitute natural gas or synthetic natural gas, has physical and chemical

582 properties comparable to NG one, therefore it can be injected into NG transportation
583 and distribution networks without limitations. Moreover, SNG has several advantages
584 over H₂ which deal with volumetric energy content and safety concerns [48,49].

585 The PtG technology allows to decarbonize the energy production and end-use
586 sectors, to store energy over long periods, to transport and distribute it even over long
587 distances, increasing the resilience and security of the energy system. Indeed, PtG
588 facilitates the penetration of non-programmable RESs allowing the storage of
589 electricity in 100% renewable energy carriers in place of conventional fossil fuel and
590 thus, supporting the decarbonization especially of those sectors difficult to electrify
591 such as part of industry and transport. Another benefit of PtG is the valorisation of the
592 existing NG infrastructure, avoiding electricity grid implementations, through the
593 electricity and gas networks integration.

594 Nevertheless, technical, economic and regulatory barriers hinder the spread of this
595 technology. PtG technological challenges include electrolyser development in
596 dynamic regimes, heat valorisation for catalytic methanation, integration of the
597 different technologies in an optimized PtG system. The more expensive green
598 hydrogen and SNG costs with respects to fossil gases represent a definite drawback,
599 to be overcome by government incentives, particularly in the assessment and
600 technological development step.

601 In the literature, several studies have been devoted to the economic analysis of the
602 PtG technology. Most studies agree that the biggest cost factor contributing to capital
603 expenditure is the electrolyser. The investment for the electrolysis unit changes in the
604 range of approximately 800 – 3000 €/kW_{el}, depending on the electrolysis technology
605 adopted [50]. Less information on the capital costs of the methanation unit is available
606 in literature. In [51], it has been reported that the investment costs of a 5MW plant and
607 a 110 MW are equal to 400 €/kW_{SNG} and 130 €/kW_{SNG}, respectively. Lehner et al. [52]
608 identified a range of 300 - 500 €/kW_{SNG}, comparing three different studies. Gassner et
609 al. [53] investigated a 14.8 MW CO methanation, operating at 15 bar, finding that the
610 investment costs for the methanation section are 175 €/kW, which is in agreement
611 with data from [51]. In [54], it has been estimated that the current costs of methanation
612 unit is about 1500 €/kW_{SNG}, but it is expected to decrease up to 1500 €/kW_{SNG} by
613 2030. In addition to capital costs of electrolysis and methanation units, it is also
614 necessary to take into account the investment cost of H₂ storage that according to Götz
615 et al. [50], it is the second most significant contributor to the investment of a PtG plant.
616 In [55], the authors evaluated the investment costs (i.e., costs of the main plant sections
617 and additional costs) for a 100 MW_{el} PtG plant equal to around 223 m€, 107 m€, and

618 54 m€ in 2020, 2030, and 2050, respectively. As regards the operational expenditures,
619 they are especially influenced the electricity price and potential CO₂ costs as well as a
620 possible heat and oxygen usage. The electricity costs vary depending on the field of
621 application and likewise, the electricity market is subject to constant change. The costs
622 for CO₂ capture vary widely, from about 5 to 350 €/t depending on the CO₂ source and
623 technology. In general, it can be assumed that the CO₂ required for methanation is
624 approximately 50 €/t. Selling oxygen can reduce the SNG production costs. This is
625 particularly advantageous in the case of oxygen is needed at the site of the PtG plant
626 (e.g., in wastewater treatment plants). The oxygen price is about 50 €/t for large
627 industrial use. Higher costs are incurred for small amounts in bottles (e.g., medical
628 oxygen) of around 5 €/kg on average. For the oxygen produced by an electrolyser, it
629 can assume a price of 50 €/t. The waste heat from electrolysis and methanation
630 processes can also be sold, reducing the operating costs and at same time increasing
631 the PtSNG plant efficiency. However, the integration of waste heat is highly dependent
632 on the existing framework conditions at the plant site and the technology used.
633 Generally, it can suppose that the waste heat can be sold at a price of about 55 €/MWh
634 [55]. Finally, according to the data reported in [50], the SNG production costs vary in
635 the range of 60 – 4 eurocents/kWh_{SNG}, that is not economically sustainable yet, if
636 compared with NG production costs (i.e., 2 – 3 eurocents/kWh).

637 In the following paragraphs, the current state of the art of core technologies in the
638 PtG process (i.e., electrolysis, methanation) is presented. Additionally, in order to
639 investigate the technical feasibility of the PtG concept to store intermittent RESs, two
640 different case studies have been analysed and discussed. In the former, a PtSNG plant
641 coupled with a RES facility has been designed and investigated by evaluating nominal
642 and annual efficiencies for different sizes of the electrolysis unit. In the latter, a novel
643 thermal management strategy for a PtSNG plant and how it impacts on the annual plant
644 performances have been studied.

645 **2.2 State of the art technologies**

646 **2.2.1 Electrolysis process**

647 Water electrolysis consists of circulating a direct current through water to separate
648 its molecules into hydrogen and oxygen. The current flows between two electrodes
649 separated and immersed in an electrolyte to raise the ionic conductivity. The
650 electrolysis process requires the implementation of a diaphragm or separator to avoid

651 the recombination of H_2 and O_2 generated at the electrodes. The electrodes, the
 652 diaphragm, and the electrolyte are the elements that configure the electrolytic cell.

653 The global electrolysis reaction taking place is:



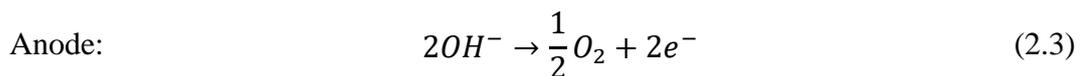
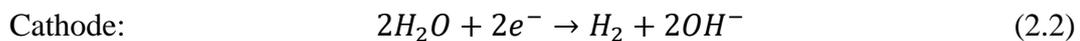
654 In the electrolysis process, the electrons are taken or released by the ions at the
 655 electrodes surface, generating a multiphasic gas-liquid-solid system. The reduction
 656 half reaction takes place at the cathode. The electrons flow to this electrode from the
 657 outside circuit and polarize it negatively. The oxidation half-reaction occurs at the
 658 other electrode, called anode. The electrons leave the anode to the outside circuit,
 659 polarizing it positively. Hydrogen is hence generated at the cathode and oxygen at the
 660 anode.

661 Currently, three electrolytic processes can be considered for the PtG process chain:
 662 i) alkaline electrolysis (AEL), ii) proton exchange membrane electrolysis (PEMEL),
 663 and iii) the solid oxide electrolysis (SOEL). The following sub- section describes the
 664 three electrolysis technologies considered.

665 **2.2.1.1 Alkaline electrolysis**

666 The alkaline electrolysis cell consists of two electrodes separated by a microporous
 667 membrane. This assembly is immersed in an aqueous alkaline electrolyte, generally
 668 composed of 20 – 40% wt. of potassium hydroxide (KOH) or, sometimes, of sodium
 669 hydroxide (NaOH) or sodium chloride (NaCl), to maximize its ionic conductivity.

670 The operating principle of an AEL cell is illustrated in Figure 2.1. Hydrogen gas
 671 evolves from the cathode, where water is reduced according to (2.2) yielding
 672 hydroxide anions that circulate across the diaphragm to the anode within the electric
 673 field established by an external power source. The hydroxide anions recombine on the
 674 anode surface according to (2.3) to produce oxygen.



675 A diaphragm separates the reaction zones from each other and prevents mixing of
 676 the produced gases. Hence, the diaphragm is permeable to water ions but impermeable
 677 to gases. In conventional electrolyzers asbestos is used as diaphragm material.
 678 Adjacent to both sides of the diaphragm electrodes are placed. The anode is usually

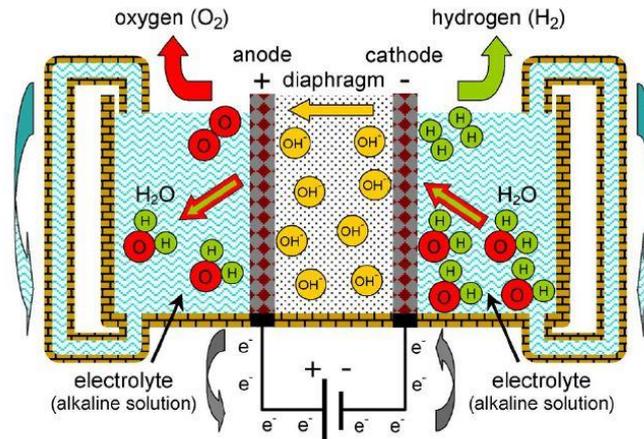
679 made of nickel or nickel coated steel, while the cathode is made of steel activated by
680 a coating with different catalysts [56].

681 Typical operating temperature range is 40 – 90 °C with relatively low current
682 densities of 0.2 and 0.4 A/cm² [57]. The operating pressure range is generally between
683 1 bar and 30 bar [46,58]. Interestingly, the maximum pressure is sufficiently high to
684 avoid one of the compression stages when hydrogen is to be stored as compressed gas.
685 Higher pressures do not offset because the investment costs increase significantly as
686 well as the risks of formation of hazardous gas mixtures [58]. The energy demand of
687 an alkaline electrolyzer is dependent on the characteristics of the electrodes and
688 operational conditions. Atmospheric electrolyzers have a specific energy demand of
689 about 4.1 – 4.5 kWh/Nm³ H₂ and pressurized electrolyzers have a demand of about 4.5
690 – 5 kWh/Nm³ H₂ [56]. The atmospheric electrolyzers have an efficiency up to 85%
691 (HHV) and the high-pressure ones have up to 78% (HHV). The purity levels of
692 hydrogen and oxygen can reach 99.5%, without auxiliary purification equipment
693 [57,59]. On the other hand, the water supplied to the electrolyser has to be adequately
694 pure to protect the electrodes and operate safely.

695 AEL is the most mature technology and broadly employed for large-scale industrial
696 application since decades [46,57]. AEL systems are reliable and safe, and exhibit
697 lifetimes that can reach up to 15 years [58]. The investment costs are relatively low
698 due to the avoidance of noble metals and mature stack components. However, low
699 current density and operating pressure negatively impact system size and hydrogen
700 production costs [60]. In the last years, the efficiency of the electrolyzers has been
701 improved with the aim of reducing operating costs associated with the electricity
702 consumption. The operating current densities have been increased with the aim of
703 decreasing the cell surface area to reduce the investment costs. These costs are almost
704 proportional to the electrolyser's cell surface area [56].

705 Current issues associated with AEL technology when applied in PtG applications
706 regard their limited minimal part-load capability. According to manufactures, the
707 production rate is in practice limited to 20% – 100% of the nominal range, in order to
708 prevent operating at conditions that could potentially lead to the formation of
709 flammable gas mixtures [46,61]. However, recently progresses have been reported.
710 New AEL designs would allow variable operation ranging from 5% to 100% of the
711 nominal capacity [45,62]. Additionally, dynamic operation is limited negatively
712 affecting system efficiency and gas purity, especially when AEL electrolyzers are
713 supplied by intermittent and fluctuating power sources. In fact, the response time is

714 typically seconds or minutes. They can take 30 – 60 min to restart the system following
 715 a shutdown [45,46].



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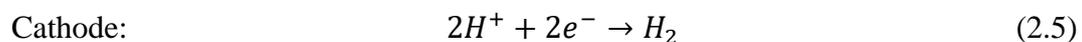
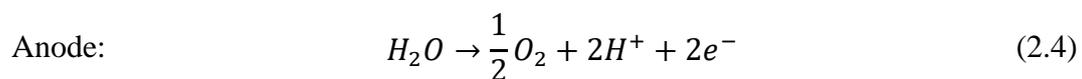
Figure 2.1 – Scheme of the operating principle of an alkaline electrolysis cell [58].

718 2.2.1.2 Proton exchange membrane electrolysis

719 In PEM electrolyzers, the electrolyte is a gas-tight thin polymeric membrane with a
 720 cross-linked structure and strongly acid character due to the presence of functional
 721 groups of the sulfonic acid (-SO₃H) type. These groups are responsible for the proton
 722 (H⁺) conducting ability of the materials through an ion exchange mechanism. The most
 723 commonly used membrane for water electrolysis is Nafion [58].

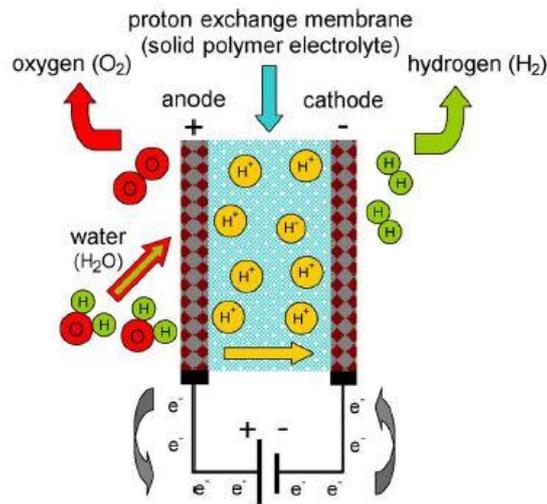
724 The modules of the proton exchange membrane water electrolyzers almost
 725 invariably adopt a bipolar configuration in which the electrical connection between
 726 cells is carried out by means of bipolar plates that also play an important role in the
 727 evacuation of the generated gases. The anode, cathode, and membrane set constitute
 728 the so-called membrane electrode assembly (MEA). The electrodes typically consist
 729 of noble metals such as platinum or iridium [58].

730 The operating principle of a PEMEL cell is shown in Figure 2.2. At the anode, water
 731 is oxidized according to (2.4) to produce oxygen, electrons, and protons that circulate
 732 across the membrane to the cathode where they are reduced according to (2.5) closing
 733 the circuit and producing hydrogen.



734 PEMEL electrolyzers operate at a temperature of around 80 °C and a pressure of up
735 to 15 bar [56]. Some models reach pressures up to 85 bar. This is due to the compact
736 character of the electrolysis modules as well as the structural properties of the MEAs,
737 which can endure big pressure differences between the electrode compartments. These
738 devices can operate at current densities that are quite higher than their alkaline
739 counterparts, achieving values even above 1.6 A/cm² [58]. Specific energy demand is
740 typically in the range of 6 – 8 kWh/Nm³ H₂, but it can also be below 6 kWh/Nm³ H₂
741 in large-scale systems with a production rate higher than 10 Nm³ H₂/hr. Their
742 efficiencies are in the range of 67 – 82% [56]. The hydrogen purity, higher than in
743 alkaline electrolyzers, is typically above 99.99 vol.%, without the need for any further
744 purification equipment [45]. Moreover, the very low gaseous permeability of the
745 polymeric membranes lowers the risk of formation of flammable mixtures; hence to
746 operate at very low current densities is permissible [58].

747 Compared to AEL, PEMEL technology is a relatively new technology. The first
748 commercial PEMEL electrolyzer was available for purchase in 1978 [45,46]. Because
749 of the use of noble materials (i.e., platinum group metals) as catalysts, the cost has
750 been estimated to be at least double (i.e., > 2000 €/kW) that of the AEL technology.
751 Moreover, the lifetime is shorter than that of the AEL systems and falls within the 5 –
752 20 year range [45]. However, PEMEL is considered to be the best choice for PtG plant
753 due to its ability to operate under variable power feeding regimes [62]. This is owing
754 to the fact that the proton transport across the polymeric membrane responds quickly
755 to power fluctuations. On the contrary, the ionic transport in liquid electrolytes in AEL
756 electrolyzers shows a greater inertia [56]. Indeed, PEMEL electrolyzers can be
757 completely ramped up and down in just a few seconds, additionally, they can work in
758 the 0 – 100% range and a cold start can be completed in just minutes.

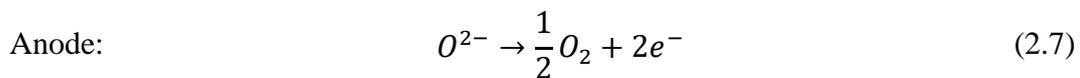
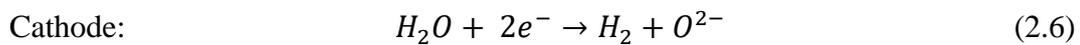


759

760 *Figure 2.2 – Scheme of the operating principle of a proton exchange membrane electrolysis cell [58].*761 **2.2.1.3 Solid oxide electrolysis**

762 Solid oxide electrolyzers constitute an advanced concept of water, or rather steam,
 763 electrolysis at high temperature (i.e., up to 1000 °C), which results in higher
 764 efficiencies compared to AEL or PEMEL electrolyzers [56].

765 The operating principle of a SOEL cell is illustrated in Figure 2.3. Both steam and
 766 recycled hydrogen are fed to the cathode, where water is reduced to produce hydrogen
 767 according to (2.6). The oxide anions generated in the cathode pass through the solid
 768 electrolyte to the anode, where they recombine according to (2.7) forming oxygen and
 769 closing the circuit with the released electrons.



770 These reactions evolve with the electrodes in contact with a gas or vapor phase,
 771 which is in clear contrast with the processes taking place on the electrodes of the AEL
 772 or PEMEL electrolyzers and makes the issue of maximizing the interfacial area in
 773 contact between the electrodes and the gaseous chemical species challenging. This is
 774 the reason of the porous character of the SOEL's electrodes. The solid oxide electrolyte
 775 is typically made up of a gas-tight thin film of yttria (Y_2O_3) stabilized zirconia (ZrO_2)
 776 (YSZ) [46]. This material exhibits good ionic conductivity at the prevailing high
 777 operating temperatures through a mechanism that involves the oxygen vacancies

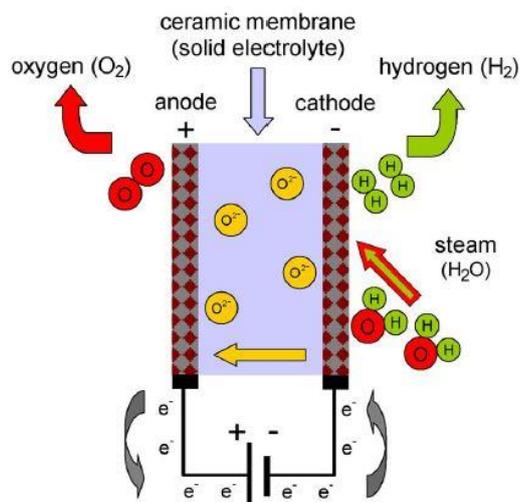
778 present in the mixed oxide lattice. The cathode is a cermet usually consisting of nickel
779 and YSZ. The anode is commonly a composite of YSZ and perovskites such as
780 lanthanum manganites (LaMnO_3), ferrites (LaFeO_3) or cobaltites (LaCoO_3) partially
781 substituted with strontium in order to promote structural and electronic defects that
782 increase the electrocatalytic activity [58].

783 Steam electrolysis emerged aiming to reduce the energy demand and so, the
784 operating costs of conventional liquid water electrolysis. Indeed, the electrical energy
785 demand significantly decreases as the temperature increases. This means that the
786 proportion of the required energy that can be supplied in the form of heat increases
787 with electrolysis temperature. Specifically, up to 40% of the energy required to
788 produce hydrogen via steam electrolysis can be supplied as heat at 1000 °C and the
789 electricity demand can be reduced by up to 25% [56]. Therefore, those electrolyzers
790 are expected to reach higher efficiency than that of AEL and PEMEL electrolyzers
791 [46,62][46,50].

792 Currently, SOEL electrolyzers are at a research and development stage. They are
793 not yet widely commercialised, but systems have been developed and demonstrated
794 on laboratory scale and individual companies are aiming to bring this technology to
795 market [57]. However, the main current obstacle for the commercialization of SOEL
796 technology are the fast material degradation and the limited long-term stability of the
797 electrolysis cells due to high temperature operation. Furthermore, the high temperature
798 level means that the product stream from the electrolyser is a mixture of hydrogen and
799 steam, requiring additional processing; thereby further increasing the capital costs
800 [46,56]. Thus, current research is focussed on stabilising existing component
801 materials, developing new materials, and lowering the operation temperature to 500 –
802 700 °C to enable the commercialisation of this technology [57].

803 As regard the SOEL technology application into the PtG process chain, it could lead
804 to a higher overall efficiency of the process if combined with exothermal reactions
805 (e.g., CO_2 methanation). Nevertheless, SOEL systems are not stable against fluctuating
806 and intermittent power sources [46,56].

807 The construction characteristics and operating parameters of the above-described
808 electrolysis technologies are listed in Table 2.1.



809

810

Figure 2.3 – Scheme of the operating principle of a solid oxide electrolysis cell [52].

811

Table 2.1 – Construction characteristics and operating parameters of electrolysis technologies

812

[57,59].

	AEL	PEMEL	SOEL
State of development	Commercial	Commercial	Laboratory
Electrolyte	20 – 40 %wt. KOH	Nafion	YSZ
Cathode	Ni, Ni-Mo alloys	Pt, Pt-Pd	Ni/YSZ
Anode	Ni, Ni-Co alloys	RuO ₂ , IrO ₂	LSM/YSZ
Charge carrier	OH ⁻	H ₃ O ⁺ /H ⁺	O ²⁻
Cell temperature [°C]	40 – 90	20 – 100	750 – 1000
Cell pressure [bar]	<30	<80	<25
Cell voltage [V]	1.8 – 2.4	1.8 – 2.2	0.91 – 1.3
Current density [A/cm ²]	0.2 – 0.4	0.6 – 3.0	0.3 – 2.0
Production rate [m ³ /h H ₂]	up to 760	up to 450	-
Stack energy consumption [kWh _{el} /m ³ H ₂]	4.2 – 5.9	4.2 – 5.5	>3.2
System energy consumption [kWh _{el} /m ³ H ₂]	4.5 – 6.6	4.2 – 6.6	>3.7
Gas purity [%]	>99.5	99.99	99.9
Lower load [%]	10 – 40	0 – 10	>30
System Response	Seconds	Milliseconds	-
Cold-start time [min]	<60	<20	>60
Stack Lifetime [h]	60000–90000	20000–60000	<10000
Replacement stack [year]	8 – 12	-	-
Capital Cost [€/kW _{el}]	1000 – 1200	1860 – 2320	>2000

813 **2.2.2 Methanation process**

814 For the production of SNG, further processing of the H₂ and CO₂ via the Sabatier
 815 reaction is required. The chemism of CO₂ methanation reaction is known for more than
 816 one century and chemical methanation processes have been state of the art for several
 817 decades. They have been and still are applied to produce SNG from synthesis gas
 818 derived from coal or biomass. Gas purification in chemical or petrochemical industries
 819 is another widely used application of the methanation process. Although methanation
 820 is technological mature in these fields of application, specific differences and
 821 challenges arise when used as process step within the PtG concept. This paragraph
 822 gives an overview of the state of the art of methanation processes and the specifics for
 823 the application of this technology as part of PtG.

824 **2.2.2.1 Fundamentals**

825 Methanation of CO₂ is an exothermic process through which H₂ and CO₂ react to
 826 form CH₄ and H₂O. The reaction is described by described by the following
 827 stoichiometric equation:



828 The most widely accepted mechanism of the CO₂ methanation reaction is the
 829 combination of the endothermic Reversed Water Gas Shift (RWGS) reaction as an
 830 intermediate reaction (2.9) and the exothermic methanation of the CO (2.10) produced
 831 from the RWGS reaction:



832 Low temperatures, typically between 200 °C and 550 °C depending on the used
 833 catalyst, and high pressures favour thermodynamically the conversion of CO₂ to CH₄.
 834 In a simple one-stage process and with an operating pressure of 20 bar and keeping the
 835 temperature at 450 °C, the carbon dioxide conversion reaches approximately 93% and
 836 an increase of the operating pressure does not significantly improve the methane yield
 837 [63].

838 The product gas leaving the reactor contains steam, CO and unconverted educts
 839 beside the product CH₄. The product composition can be influenced by the
 840 methanation process concept, the reaction parameters and also by the reactor types

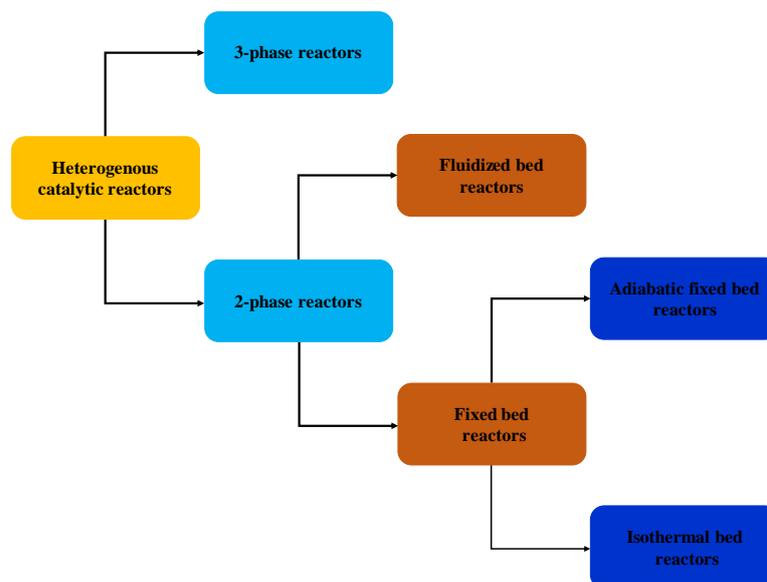
841 used. Additionally, the applied catalyst influences kinetics, conversion rate and
842 selectivity of the process.

843 Catalytic active substances for the hydrogenation of CO₂ or CO are group VIII
844 metals, i.e., the Fe-group, the Co-group and the Ni-group. Mainly due to the reasonable
845 cost and satisfactory performance in terms of conversion rates and selectivity, Ni based
846 catalysts are widely utilized for methanation processes today. Usually, silica-based
847 carriers are used, but also zeolites or metal carriers are known. Basically, catalysts are
848 sensitive to poisons, which may result in catalyst deactivation. Typical catalyst poisons
849 are heavy metals, but also sulphur compounds or oxygen [64].

850 2.2.2.2 Process concepts

851 Due to the highly exothermic nature of CO₂ methanation reaction, a significant issue
852 in the methanation process is the temperature control as a means to prevent
853 thermodynamic limitation and catalyst sintering. To solve such issue, several steady
854 state reactor concepts have been developed [46].

855 In Figure 2.4, the classification of reactors employed in the methanation process is
856 depicted. Specifically, all developed chemical methanation reactors involve the use of
857 a solid catalyst and for this reason, they belong to the category of heterogenous
858 catalytic reactors. The latter can be classified into 2-phase reactors and 3-phase ones,
859 which differ from the 2-phase systems in the application of a liquid energy carrier to
860 remove the heat released by the methanation reaction. The 2-phase reactors, in turn,
861 can be categorized into fixed bed reactors (i.e., adiabatic fixed-bed reactors, isothermal
862 fixed-bed reactors) and fluidized bed reactors.



863

864

*Figure 2.4 – Classification of methanation reactors.***2.2.2.2.1 Adiabatic fixed-bed reactors**

866 Fixed-bed reactors are state of the art for methanation process in the PtG plant
867 [62,65]. Fixed-bed methanation utilizes catalysts, in pellet form, some millimetres in
868 size which are dumped randomly into the reactor forming a preferably homogeneous,
869 static catalyst bed [64]. Thus, the contacting of the catalyst particles by the gas tends
870 to be quite uniform, and long contact times are possible [62]. The fixed-bed reactors
871 can be designed adiabatic or isothermal, as abovementioned.

872 Adiabatic fixed-bed reactors typically operate between 250 °C and 700 °C [46], and
873 in a pressure range of between 20 and 100 bar [45]. Mass transfer limitations between
874 the gases and the solid catalyst are a disadvantage of the fixed bed types, whilst the
875 mechanical stress to the catalyst is comparably low, and hence one of the main benefits
876 [64]. For adiabatic fixed bed-reactors, the usual approach relies on a series of adiabatic
877 reactors with intermediate cooling steps for a better temperature control inside the
878 reactors and consequently, obtaining high CO₂ conversion [62,63]. The temperature
879 control is sometimes realised through the partial product gas recirculation. Hence, the
880 processes using adiabatic fixed-bed reactors can be distinguished between processes
881 with the partial product gas recycle and processes without the partial product gas
882 recycle.

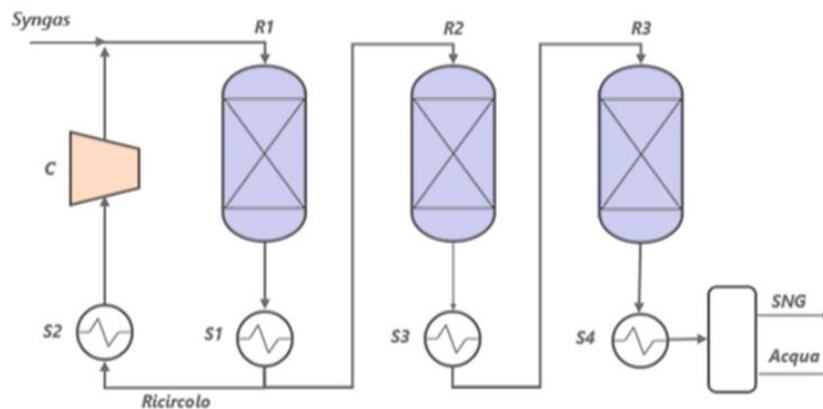
2.2.2.2.1.1 Process with recycle***TREMP process***

884 The TREMP (Topsøe's Recycle Energy efficient Methanation Process) process was
885 developed by Danish company Haldor Topsøe in 1970, through the experimental
886 activity carried out within the context of EVA I and ADAM I projects [66].
887

888 The TREMP technology addresses the essential question of minimum recycle cost
889 and heat recovery most efficiently by recovering the heat as high pressure superheated
890 steam (10 MPa, 540 °C) [67]. The process consisted of three or more adiabatic fixed
891 bed methanation reactors connected in series with intermediate gas cooling and partial
892 product gas recycle (see Figure 2.5). Namely, the exit gas from the first reactor after
893 being cooled enters the subsequent methanation reactors operating at decreasing
894 temperature levels. The number of methanation reactors depend on the operating
895 condition, such as pressure, as well as the SNG product specification. The temperature
896 range of the reactors is from 250 to 700°C and the pressure up to 30 bar [66].

897 Beside the technology, Haldor Topsøe also provided the catalyst (MCR-2X,
 898 MCR4) for the high temperature methanation. The catalyst is based on a ceramic
 899 support with a stabilized micropore system that effectively prevents sintering of nickel
 900 crystals. The resulting high nickel surface area results in the desired high methanation
 901 activity even at relatively low temperature [67].

902 The experimental parameters obtained from the tests on high-temperature
 903 methanation carried out in the bench-scale unit ADAM I are summarized in Table 2.2.



904

905

Figure 2.5 – TREMP process scheme [66].

906

Table 2.2 – Operation parameters and gas composition of TREMP process [66].

	Feed	R1 Inlet	R1 Outlet	R2 Outlet	R3 Outlet	SNG
Temperature [°C]	-	300	604	451	303	23
Pressure [bar]	27.3	27.2	27.1	27.05	27	27
Flow rate [Nm ³ /h]	535	1416	1255	348	334	191
Gas composition [vol. %]						
H ₂	65.45	36.88	20.96	8.10	1.77	3.11
CO	9.84	4.28	1.17	0.00	0.00	0.00
CO ₂	8.96	6.13	4.46	2.07	0.95	1.67
CH ₄	11.30	28.12	37.44	44.36	47.28	82.95
H ₂ O	-	19.19	29.82	38.84	43.06	0.10
N ₂	4.40	5.41	6.15	6.64	6.93	12.16

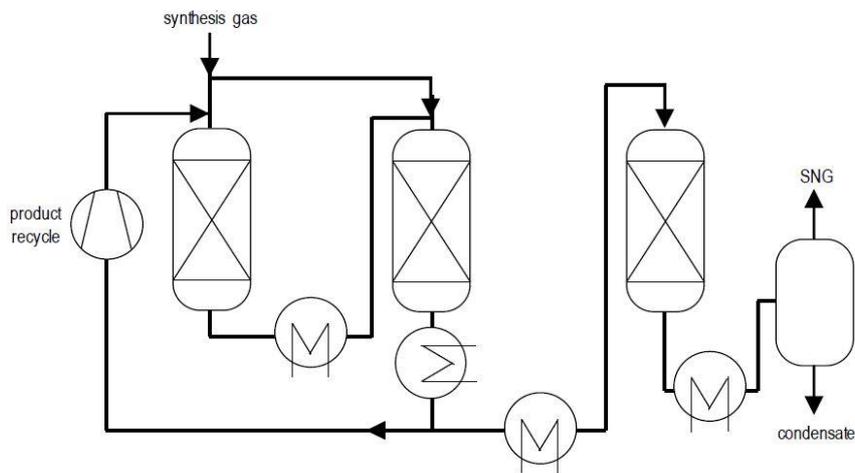
907 **Lurgi process**

908

909

The Lurgi process, developed in the 1930 in Germany, was the first commercially viable technology applicable for the production of SNG in pipeline quality. The first

Temperature [°C]	270	300	450	260	315
Flow rate [Nm ³ /h]	18.2	96.0	89.6	8.2	7.9
Dry gas composition [vol. %]					
H ₂	60.1	21.3	7.7	7.7	0.7
CO	15.5	4.3	0.4	0.4	0.05
CO ₂	13.0	19.3	21.5	21.5	21.3
CH ₄	10.3	53.3	68.4	68.4	75.9
C ₂₊	0.2	0.1	0.05	0.05	0.05
N ₂	0.9	1.7	2.0	2.0	2.0



934

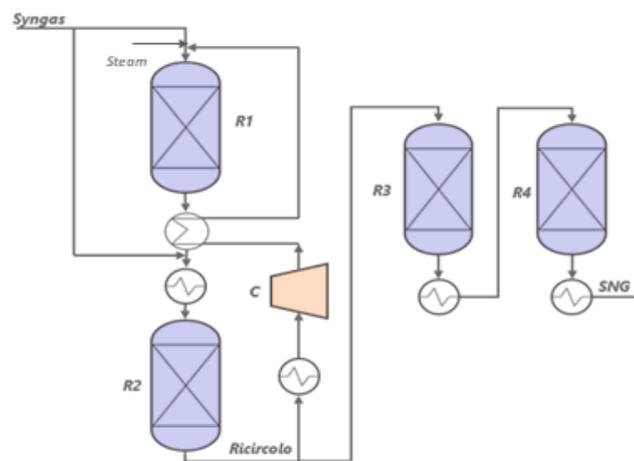
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Figure 2.7 – Lurgi process as installed in Great Plains Synfuels Plant [68].

936 **HICOM process**

937 The HICOM (High Combined Shift Methanation) process was developed by
 938 Conoco and British Gas Corporation in Scotland in 1972 [66], and is distributed
 939 nowadays by Johnson Matthey as DAVYTM technology [68]. The HICOM process
 940 differs from the aforementioned ones as it contains no separate water-gas-shift reactor.
 941 It rather integrates the shift reaction in the first methanation stage, achieving a thermal
 942 efficiency of about 70% [66]. This process aimed at low H₂/CO ratios. In order to
 943 increase the hydrogen amount, hot water was mixed in counter-current in a packed-
 944 bed with the purified syngas flow for gas heating and water saturation. CO₂ removal
 945 downstream of the methanation process lowers the temperature increase in the
 946 methanation reactors but also requires a larger reaction volume due to the increased
 947 overall gas flux. Similar to other large-scale processes, a product gas recycle and
 948 intermediate gas cooling serve for heat control in a series of three adiabatic fixed-bed
 949 reactors [68] (see Figure 2.8). A bench-scale reactor was erected for screening of

950 catalysts and process conditions. The pilot plant consisted of 37 mm diameter reactors
 951 in which long term tests under near-commercial conditions were studied. Several tests
 952 run of up to 2000 hours duration were carried out with catalyst pellets of 3.2 mm and
 953 5.4 mm [66]. Typical gas compositions and operating conditions of the pilot test are
 954 summarised in Table 2.4. A semi-commercial scale plant was built at the Westfield
 955 Development Center (Scotland). The plant was operated with an inlet temperature
 956 between 230 to 320°C and outlet temperatures in the range from 460 to 640 °C at a
 957 syngas flow rate of 5300 m³/h for a total test time of 15 000 hours [68].



958

959

Figure 2.8 – HICOM process scheme [66].

960

Table 2.4 – Operation parameters and gas composition of the HICOM process [66].

	Feed gas	Product gas
Inlet Temperature [°C]	230 – 320	
Flow rate [Nm ³ /h]	25 – 70	
Maximum Temperature [°C]	460 – 640	
Dry gas composition [vol. %]		
H ₂	11.7	5.5
CO	12.6	1.1
CO ₂	43.0	53.1
CH ₄	31.7	39.3
N ₂	1.0	1.1

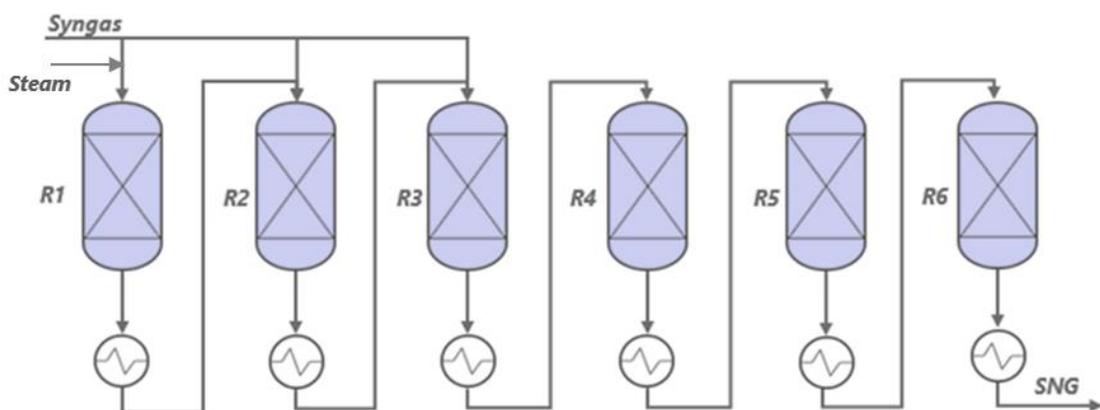
961 2.2.2.2.1.2 Process without recycle

962 RMP process

963 Ralph M. Parsons Company proposed a high temperature methanation concept
 964 without gas recycle and no separate shift conversion unit (RMP process). The
 965 methanation process consists of 4 to 6 adiabatic fixed bed methanation reactors in
 966 series with intermediate gas cooling (see Figure 2.9). Similar to the HICOM process,
 967 the RMP process involves the steam addition before the first stage in combination with
 968 a staged feed gas injection in the first three reactors in order to control the temperature
 969 increase in each single adiabatic methanation reactor. Such configuration is so-called
 970 ‘Steam Quenching Methane Synthesis’ [68].

971 The system pressure was varied between 4.5 and 77 bar, the inlet temperatures of
 972 the reactors were varied between 315 and 538°C, and the H₂/CO ration was varied
 973 between 1 and 3 [66].

974 The following Table 2.5 shows the dry gas composition and the temperatures of an
 975 experimental run at 27 bar. 40% of the total cleaned syngas with H₂/CO ratio of one
 976 entered the first reactor together with steam with an inlet temperature of 482°C. The
 977 product gas was cooled, then mixed with 30% of the syngas, and then fed to the second
 978 reactor. The last 30% of the syngas were added in the third reactor together with the
 979 product gas from reactor two. The inlet of the fourth, fifth, and sixth reactors were
 980 temperature controlled as shown in Table 2.5. In the first reactor, CO was mainly
 981 converted to CO₂ by the water gas shift and minor the CH₄. H₂O and CO₂ are removed
 982 from the product gas leaving the sixth reactor and fed to final dry methanation stage
 983 to reduce the H₂ and CO content below 3 and 0.1 vol%, respectively. After 1977 no
 984 more information about this project are available [66].



985

986

Figure 2.9 – RMP process scheme [66].

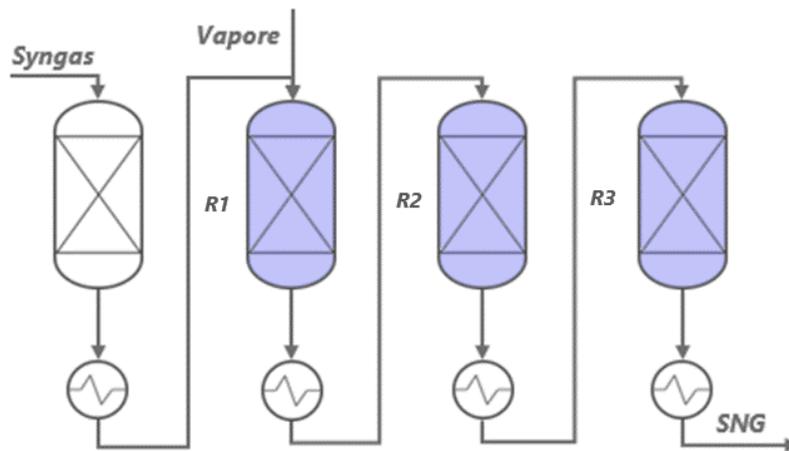
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Table 2.5 – Operation parameters and gas composition of the RMP process [66].

	R1 Inlet	R1 Outlet	R2 Outlet	R3 Outlet	R4 Outlet	R5 Outlet	R6 Outlet
Inlet Temperature [°C]	-	482	538	538	538	316	260
Outlet Temperature [°C]	482	773	779	773	717	604	471
Pressure [bar]	27.3	26.7	25.6	24.6	23.6	22.6	21.5
Inlet Syngas [vol. %]	40	-	30	30	-	-	-
Steam/gas ratio	1.20	0.88	0.56	0.43	0.50	0.65	0.83
Dry gas composition [vol. %]							
H ₂	49.8	53.5	48.1	43.1	36.9	22.9	9.3
CO	49.8	14.0	18.5	20.6	15.3	5.6	0.88
CO ₂	0.1	25.8	24.0	23.6	29.2	39.9	46.8
CH ₄	0.3	5.7	9.4	12.6	18.6	31.6	43.0

988 **ICI process**

989 Similar to the RMP process, the Imperial Chemical Industries ICI (Great Britain)
 990 developed a catalyst and a high temperature once-through methanation process, using
 991 a catalyst with a high nickel content (up to 60%). This process consists of three
 992 adiabatic fixed bed reactors in series with intermediate gas cooling (see Figure 2.10).
 993 The steam added to the second reactor in order to maintain the temperature below
 994 750°C. On the basis of this scheme no large scale plant have been built [66].



995

996

Figure 2.10 – ICI process scheme [66].

997

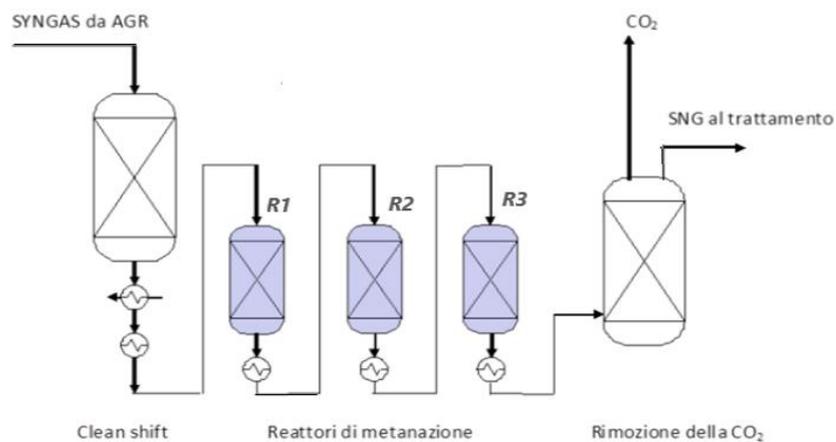
Table 2.6 – Operation parameters and gas composition of the ICI process [66].

	R1 Inlet	R1 Outlet	R2 Inlet	R2 Outlet	R3 Inlet	R3 Outlet
Temperature [°C]	398	729	325	590	300	428
Dry gas composition [vol. %]						

H ₂	42.9	35.5	35.5	20.3	20.3	5.8
CO	31.1	14.5	14.5	4.3	4.3	0.3
CO ₂	24.7	40.2	40.2	53.9	53.9	62.7
CH ₄	0.1	8.5	8.5	19.8	19.8	29.1
N ₂	1.2	1.4	1.4	1.7	1.7	2

998 **VESTA process**

999 The VESTA process was developed by Amec Foster Wheeler and Clariant, on the
 1000 basis of the technology from ICI developed in the 1970s [69]. This process comprises
 1001 a high temperature shift reactor followed by three adiabatic methanation reactors (see
 1002 Figure 2.11). Temperature control is achieved through steam control in the synthesis
 1003 gas. Furthermore, CO₂ removal takes place downstream of methanation and, hence,
 1004 brings additional thermal ballast into the system. These measures keep the adiabatic
 1005 synthesis temperature below the tolerable catalyst limit (i.e., 550°C or 650°C). A first
 1006 pilot plant with 100 Nm³/hr capacity was erected in Nanjing, China, and started-up in
 1007 July 2014 [68].



1008

1009

Figure 2.11 – VESTA process scheme.

1010 **2.2.2.2 Isothermal fixed-bed reactors**

1011 The isothermal conditions in fixed-bed methanation reactors can be accomplished:
 1012 (i) by inserting cooling coils or tubes into the random bed of catalyst particles, (ii)
 1013 keeping the catalyst inside tubes and the cooling medium on the shell side, or (iii) by
 1014 coating the catalyst to surfaces which are cooled from the other side. By far the most
 1015 common type of cooled fixed bed reactors are multi-tubular reactors, where the
 1016 catalyst is filled into the tubes while the cooling medium is on the shell side [70].

1017 In comparison to adiabatic reactors, the isothermal reactors operate in lower
1018 pressure (i.e., 1 – 20 bar) and temperature (i.e., 280 – 300 °C) ranges. Moreover, they
1019 are characterized by scalability, large heat transfer areas, good mass exchange process
1020 and relatively short transport distances within the bed to the next heat transfer. On the
1021 other hand, some of its disadvantages include the construction and maintenance
1022 complexity [70,71].

1023 Recently, PtG raised again interest in cooled reactors. Hence, the first commercial
1024 PtG plant in Werlte, Germany, comprises a cooled DWE® tube bundle reactor
1025 manufactured by MAN engineering, which is in operation since 2013 [68].

1026 *Linde process*

1027 In the 1970s, Linde AG (Germany) developed an isothermal fixed bed reactor with
1028 indirect heat exchange. In this reactor, the cooling tube bundles are embedded in the
1029 catalyst bed, as illustrated in Figure 2.12.

1030 The process flow diagram of the Linde process with one isothermal and one
1031 adiabatic fixed bed reactor is shown in Figure 2.13. The temperature control of the
1032 isothermal reactor is carried out by the production of steam from the heat of the
1033 exothermic methanation reaction. A part of the steam should be added to the gaseous
1034 mixture outgoing from the isothermal reactor to minimize the risk of carbon deposition
1035 on the catalytic bed of the adiabatic reactor. The product gases of both reactors are
1036 finally mixed, cooled, and reaction water is condensed [66].

1037 The process has never been commercialized to produce SNG. Today, the Linde
1038 isothermal reactors are used in other processes of the chemical industry such as
1039 methanol synthesis, hydrogenation, ethylene oxide synthesis etc.

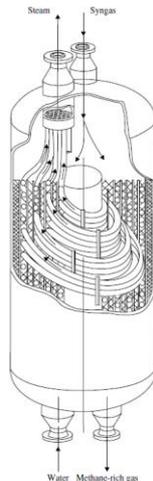


Figure 2.12 – Isothermal reactor scheme [70].

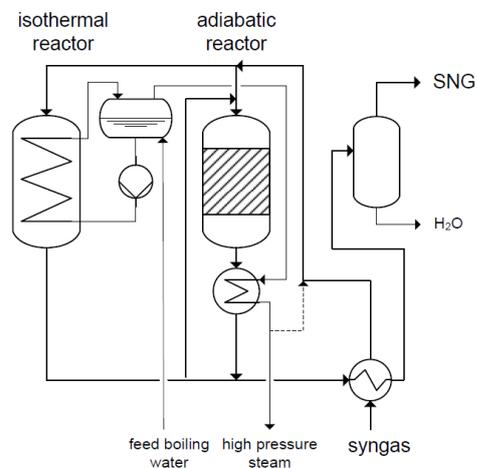


Figure 2.13 – Linde process scheme [66].

1040 2.2.2.2.3 Fluidized-bed reactors

1041 Fluidized-bed reactors are known to be suitable for large-scale operations of
 1042 heterogeneous catalysed reaction with high exothermicity [66]. The mixing of
 1043 fluidized solids leads to almost isothermal conditions in the reactor, facilitating the
 1044 control of the operation [46]. The main advantages of this reactor concept are a good
 1045 heat release and a high specific surface area of the catalyst combined with reduced
 1046 mass transfer limitations. Consequently, reactor cascades are avoided, and so a
 1047 simplified set-up is realized compared with fixed bed systems [64]. Further advantage
 1048 is the possibility to easily remove, add, and recycle catalyst and/or bed material
 1049 continuously during the process [66]. However, due to high mechanical load resulting
 1050 from fluidization, attrition processes take place in relation to the catalyst as well as the
 1051 wall of the reactor. As a consequence, the catalyst is eventually deactivated. A further
 1052 disadvantage can be the incomplete CO₂ conversion caused by bubbling. In addition,
 1053 a fluidized-bed reactor is limited by superficial gas velocity within the reactor: it
 1054 cannot be too low in order to assure minimum fluidization conditions and cannot be
 1055 too high in order to avoid catalyst elutriation [46].

1056 In 1952, the former Bureau of Mines (US Department of the Interior) developed
 1057 two multiple feed fluidized-bed methanation reactors (see Figure 2.14) to produce
 1058 pipeline quality synthetic natural gas from coal via gasification and methanation. Both
 1059 systems were operated in a temperature and pressure range of 200 to 400 °C and up to
 1060 20.7 bar, respectively, and in recycle mode with a ratio of up to four. The H₂/CO ratio

1061 of the feed gas varied between one and three. The results have shown a good
 1062 temperature control in the multiple-feed fluidized bed and that nickel was superior to
 1063 iron as a methanation catalyst. The iron catalysts were not active enough to produce
 1064 SNG in pipeline quality; the conversion of H₂ and CO was less than 80%, and in
 1065 addition, higher hydrocarbon C₂₊ and C₃₊ were formed. Nickel catalyst was found to
 1066 be very active and suitable for methanation of syngas, but extremely vulnerable to
 1067 sulphur poisoning [66].

1068 In 1963, Bituminous Coal Research Inc. (BCR, United States) initiated the Bi-Gas
 1069 project in which was developed a gas-solid fluidized-bed reactor including a second
 1070 feed inlet and two in-tube heat exchanger bundles, as depicted in Figure 2.15. The
 1071 published results of the experiments showed that the conversion of CO was in between
 1072 70 and 95%. The relatively low conversion means the product gas needed to be
 1073 converted in a final fixed bed methanation reactor [66].

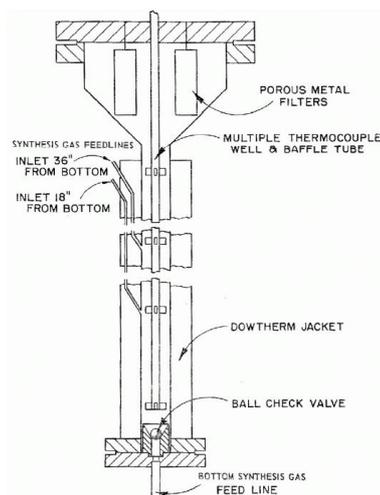


Figure 2.14 – Multiple feed fluidized-bed reactor scheme [66].

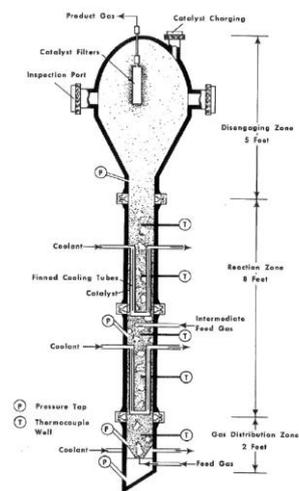


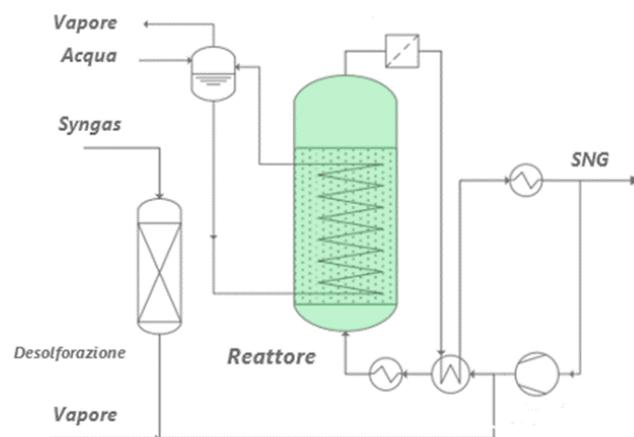
Figure 2.15 – Fluidized bed methanation reactor by Bituminous Coal Research Inc. [66].

1074 **Comflux process**

1075 The biggest fluidized bed methanation reactors so far have been erected and
 1076 operated within the Comflux project in Germany, between 1975 and 1986. plant with
 1077 a diameter of 0.4 m was erected by Didier Engineering GmbH (Germany) and operated
 1078 for several hundred hours, from 1977 to 1981 [66]. The scheme of the so-called
 1079 Comflux process and the experimental conditions are illustrated in Figure 2.16 and
 1080 Table 2.7, respectively.

1081 The Comflux process belongs to the class of isothermal reactors using heat
 1082 exchangers integrated into the fluidized bed. The single stage, nearly isothermal
 1083 fluidized-bed omits the need for a separate shift unit since the water gas shift reaction
 1084 and CO methanation take place simultaneously. The particle movement in the fluidized
 1085 bed allows for a superior heat distribution in the bed and facilitates mass transfer. The
 1086 nearly isothermal conditions also favour the reaction control. Methanation of a
 1087 synthesis gas with a H_2/CO ratio smaller than three requires an additional CO_2 removal
 1088 downstream [68].

1089 Later 1981, the Comflux process was demonstrated on the industrial scale (2000
 1090 $m^3_{SNG} \cdot h^{-1}$; up to 20 MW_{SNG}) in Oberhausen (Germany). The pilot plant consisted of a
 1091 tube with 1.0 m diameter at a fluidized bed temperature varying from 450 °C to 550
 1092 °C. The plant was operated slightly under-stoichiometric with a low recycle-gas
 1093 volume ratio from 0 to 0.3 [66,68]. In specific pilot scale experiments, it was shown
 1094 that the isothermal operation allows the methanation of synthesis gas with a H_2/CO
 1095 ratio of 1.5 by addition of steam, combining the water gas shift reaction and the
 1096 methanation in one apparatus. In commercial plant, the omission of the shift unit and
 1097 the product gas recycle compression as well as the possibility to raise all steam at the
 1098 same high-pressure level should lead to significantly reduced investment and operation
 1099 costs. With that project, the developers aimed at SNG cost 10% lower than from a
 1100 fixed bed process. The technology development was discontinued when the price of
 1101 oil dropped in the mid of the 1980 [66].



1102

1103

Figure 2.16 – Comflux process scheme [66].

1104

Table 2.7 – Experimental conditions for the Thyssengas pilot plant [66].

Parameter	Value
-----------	-------

Temperature [°C]	300 – 500
Pressure [bar]	20 – 60
H ₂ /CO ratio	up to 4
Recycle/Feed ratio	up to 2
Gas velocity [m·s ⁻¹]	0.05 – 0.2
Bed diameter [m]	0.4
Bed height [m]	2 – 4
Particle size [μm]	50 – 250
Catalyst mass [kg]	200

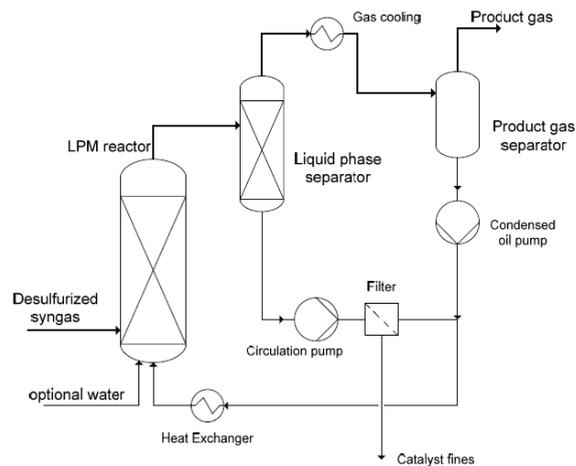
1105 2.2.2.2.4 Three-phase reactors

1106 In 1970, a three-phase fluidized-bed methanation reactor was developed by Chem
 1107 System Inc. (United States) [64,66], as depicted in Figure 2.17. Syngas produced in a
 1108 coal gasifier was introduced into the catalytic liquid phase methanation reactor along
 1109 with a circulating process liquid (i.e., mineral oil) that absorbs the heat of reaction. The
 1110 product gas was separated in a liquid phase separator and in a product gas separator.
 1111 The process liquid was pumped through a filter to remove any catalyst fines and
 1112 recirculated back to the methanation reactor. The product gas, mainly methane and
 1113 carbon dioxide with some unconverted hydrogen and carbon monoxide was analysed
 1114 and sent to a flare. No recycle gas was needed. Experiments with different nickel
 1115 catalysts (Harshaw, Engelhard, CCI and Calscat), process liquids and operational
 1116 conditions (260-360°C, 20.7-69 bar, H₂/CO ratio 1 – 10, adding water) in three
 1117 different reactors were carried out, see Table 2.8. The bench-scale unit (BSU) and
 1118 process development unit (PDU) were operated for several runs of 40 to 80 hours each
 1119 [66].

1120 By introducing a liquid phase, the heat release of the exothermic reactions is
 1121 promoted and thus an isothermal temperature profile in the reactor is achieved. The
 1122 heat carrier fluid for cooling circulates in heat exchangers, which are placed in the
 1123 liquid phase of the methanation reactor. The high heat storage capacity of the fluid is
 1124 favourable with respect to load changes and the isothermal conditions make a once-
 1125 through process without product recycle possible as no thermodynamic limitation
 1126 exists. The fluidization and mixing of the solid-liquid mixture are accomplished by
 1127 pumping the fluid or by the gas bubbles when the catalyst particles are small.
 1128 Furthermore, catalyst abrasion is reduced compared to the fluidized bed. The hydraulic
 1129 operation of a three-phase bubble column is quite sophisticated. Due to the
 1130 introduction of the liquid phase, an additional mass transfer resistance between the

1131 gaseous educts and the solid catalyst incurs which may negatively influence the
 1132 kinetics of the total process. Specifically, the right choice of the fluid is of major
 1133 importance as a low vapor pressure and high thermal stability are mandatorily since
 1134 operating temperature is commonly in the range of 260 – 360°C [64,68].

1135 Since 1972, the American company Chem Systems Inc. has been working on
 1136 commercialization of the concept. In 1977, the company erected a first pilot plant of a
 1137 liquid-phase methanation system with a total capacity of 36 000 Nm³/day at a pressure
 1138 of 34 – 52 bar. The plant was located at the Institute of Gas Technology (IGT) in
 1139 Chicago, Illinois. The project has been terminated 1981. The approach of Chem.
 1140 Systems has foreseen the circulation of the liquid phase with an external heat
 1141 exchanger to temper the reaction medium. However, this approach required a solid-
 1142 liquid separation step and pumps [68].



1143

1144

Figure 2.17 – Three-phase methanation concept [66].

1145

Table 2.8 – Three-phase methanation reactors and operational conditions [66].

Parameter	BSU	PDU	Pilot plant
Reactor diameter [cm]	2.0	9.2	61.0
Reactor height [m]	1.2	2.1	4.5
Gas flow [m ³ N·h ⁻¹]	0.85	42.5	425 – 1534
Catalyst bed height [m]	0.3 – 0.9	0.68 – 1.8	-
Catalyst mass [kg]	-	-	390 – 1000
Pressure [bar]	20.7 – 69		34 – 52
Temperature [°C]	260 – 380		315 – 360
H ₂ /CO ratio	1 – 10		2.2 – 9.5
Catalyst size [mm]	0.79 – 4.76		

1163 maximum storage value (45 bar). In the methanation unit working at 20 bar and
1164 consisting of four intercooled fixed bed reactors, the carbon dioxide and the hydrogen
1165 are converted to SNG.

1166 At the exit of the reactor MR4, the CO₂ conversion achieves the 99.2% but the
1167 methane content is only about 32.1% due to the large amount of the water produced
1168 during the methanation process. The raw SNG (14M) is, therefore, cooled down to 45
1169 °C in the heat exchanger MHE5 and sent to the flash unit where most of the water is
1170 removed. Nevertheless, in order to comply with the water dew point requirements of
1171 the pipelines, the SNG have to be further dehydrated. The technology used for the
1172 SNG drying is based on the membrane technology (a commercial Pebax®-based
1173 membrane [72]), so that a compression step is required for bringing the feed gas to the
1174 operating pressure of the dehydration membrane. The quality of the dry SNG exiting
1175 the upgrading section satisfies the pipelines specifications and can be directly injected
1176 to the low-medium pressure NG transmission grid.

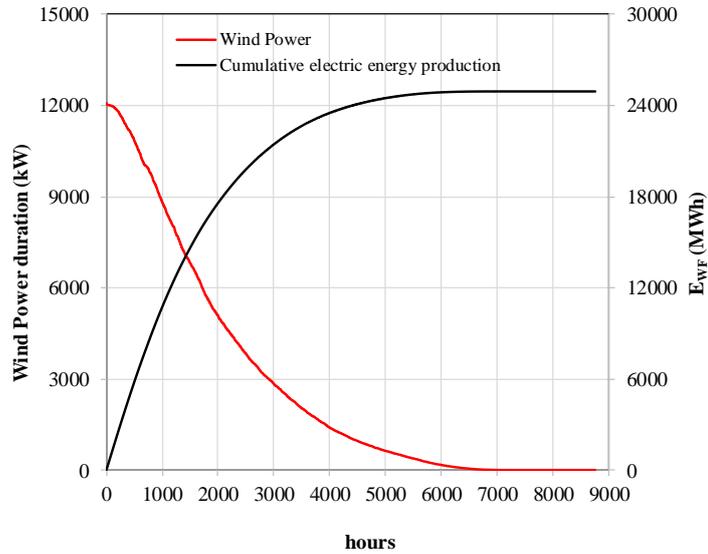
1177 It is assumed that the CO₂ is always available for the PtSNG plant (supplied by
1178 sequestration from other power plant as a by-product or industrial processes) at storage
1179 pressure of 200 bar.

1180 The PtSNG is operated in input-oriented mode [73]. This means that the SNG
1181 production depends on the availability of electrical energy and reactants (H₂ and CO₂).

1182 Finally, all the high-temperature equipment is thermally insulated in order to
1183 minimize heat dissipation, and it could be maintained in hot standby conditions using
1184 electrical trace heating to compensate the estimated heat losses [74,75].

1185 **2.3.1.2 Wind power source**

1186 The power source data (a collection of one-year hourly measurement data) used in
1187 this study refer to a 12 MW (overall capacity) real wind farm installed in Central Italy
1188 close to the NG transmission network. In Figure 2.19, the power duration curve and
1189 the cumulative curve of the electric energy generated during a year of operation are
1190 reported.



1191

1192

Figure 2.19 – Power duration and cumulative electric energy production.

1193

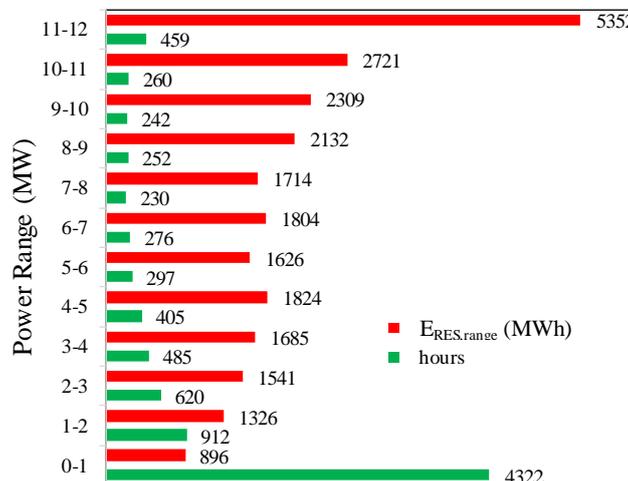
The total electric energy generated is equal to 24,930 MWh and is mainly produced in about 6000 h.

1194

1195

Figure 2.20 shows the distribution of the electric energy production and the duration time for different power ranges.

1196



1197

1198

Figure 2.20 – Distribution of the electric energy production and the duration time for different power ranges.

1199

1200

It is worth noting that the wind farm works mainly in the 0 – 1 MW power range (4322 h) producing only the 3.4% of the electric energy generated in the year, whereas the highest production percentage (21.4%) is reached in the range 11 – 12 MW with a

1201

1202

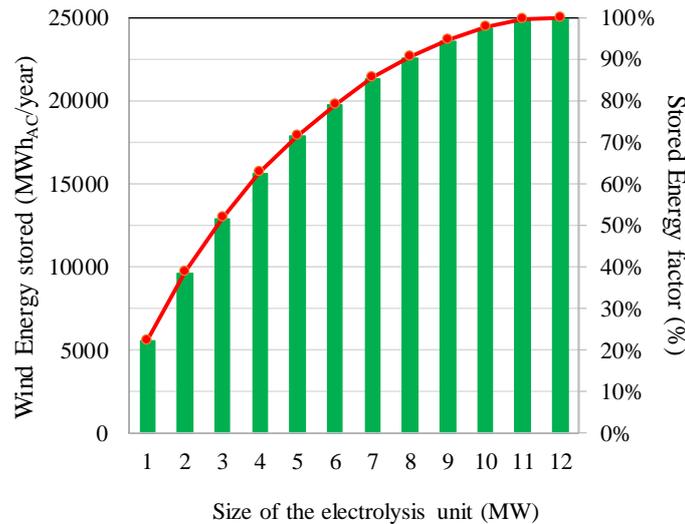
1203 duration time of 459 h. This means that, in order to optimize the PtSNG plant from an
 1204 energetic and economic point of views, the size of the electrolysis unit has to be a
 1205 compromise between the operating time and the installed power.

1206 A size parameter for the PtSNG plant is the stored energy factor, defined as the
 1207 annual electric energy supplied to the electrolysis unit of the PtSNG plant, $E_{PtSNG,EU}$,
 1208 and the annual energy generated by the wind farm E_{Wind} :

$$f_{SE} = \frac{E_{PtSNG,EU}}{E_{Wind}} \quad (2.11)$$

1209 In this parameter, the electric energy requirements of the other plant sections
 1210 (hydrogen compressor or SNG compressor) are not accounted.

1211 Using the wind farm energy data reported in Figure 2.20, the electric energy
 1212 consumption (MWh_{AC}/year) and the stored energy factor for different sizes of the
 1213 electrolysis unit have been calculated and reported in Figure 2.21 .



1214

1215 *Figure 2.21 – Electric energy consumption (MWh_{AC}/year) and stored energy factor.*

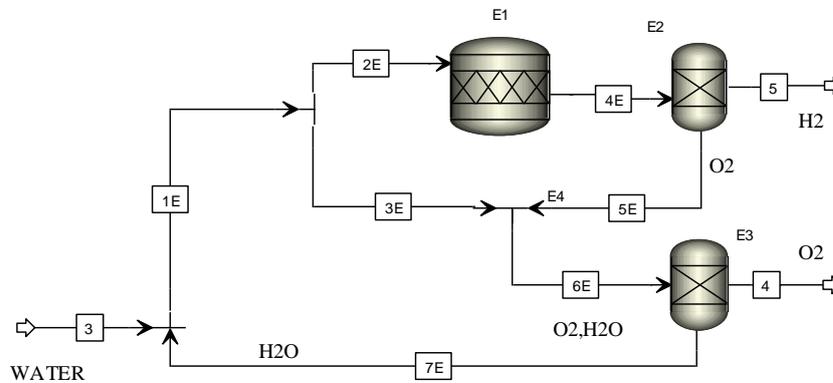
1216 The stored energy factor shows an exponential trend as the capacity of the
 1217 electrolysis unit increases, so that 79% of the wind energy production can already be
 1218 stored using a 6 MW electrolyser.

1219 Thus, in order to evaluate the energy storage potential and the technical feasibility
 1220 of the PtSNG concept to store intermittent renewable sources, the PEM electrolysis
 1221 units considered as power input sizes for the annual performance assessment of the
 1222 plant are 1, 3, and 6 MW_{DC}, corresponding to a stored energy factor (in AC) of 22%,

1242 METUNIT, or the electrolysis unit, ELUNIT) and is modelled in a sub-flowsheet by
 1243 combining operation blocks and calculator blocks for components and processes
 1244 simulation. The operation blocks are defined in the software library, therefore, once
 1245 the specific block is selected, the solving equations are defined. The calculator blocks
 1246 are developed by the user by implementing the model equations in Fortran language.
 1247 The Peng-Robinson cubic equation of state, which can accurately describe both the
 1248 liquid and vapor phase for systems containing hydrocarbons and related compounds
 1249 and for high H₂ content systems in a wide range of pressure and temperature, has been
 1250 used for evaluating properties and characteristics of each material stream.

1251 2.3.1.3.1 Electrolysis Unit (ELUNIT Hierarchy)

1252 The PEM electrolysis unit is built with a modular architecture. The module is
 1253 formed by stacks each of which consists of cells. Therefore, the element of the system
 1254 is the single cell. In Figure 2.23, the flowsheet of the single cell of the PEM electrolysis
 1255 unit is depicted. It is modelled by means of a RStoich operation block, where the water
 1256 decomposition takes place, and a separator operation block. Mixers and splitters are
 1257 also used for accounting of the water utilization factor in the anode side.



1258

1259

Figure 2.23 – Flowsheet of PEM electrolysis unit model.

1260 The electrochemical behaviour of the electrolysis module is simulated by means of
 1261 a Fortran calculator block in which the average cell polarization curve of each stack is
 1262 calculated through the following equation that takes into account the polarization
 1263 losses (i.e., activation overpotential, ohmic overpotential, and concentration
 1264 overpotential):

$$V_{cell} = V_0 + b \cdot \ln(J_{cell}) + R \cdot J_{cell} + m \cdot \exp(-n \cdot J_{cell}) \quad (2.12)$$

1265 In the above equation, V_0 (V), b (V/dec), R ($W \cdot cm^2$), m (V), and n (cm/A) are the
 1266 fitting parameters calculated by applying a regression technique on the experimental
 1267 data of the 3-cells stack reported in [76] and operating at 20 bar and 54, whereas J_{cell}
 1268 is the current density (A/cm^2).

1269 By considering the cells stack number (n_{cells}) and the cell area (A_{cell} , cm^2), the
 1270 electric DC stack power (kW) required at different current is:

$$P_{el,Stack} = \frac{n_{cells} \cdot V_{cell} \cdot J_{cell} \cdot A_{cell}}{1000} \quad (2.13)$$

1271 The stack hydrogen production (mol/s) is calculated by applying the Fick's law:

$$n_{H2,Stack} = \frac{n_{cell} \cdot J_{cell} \cdot A_{cell}}{2F} \quad (2.14)$$

1272 where F is the Faraday's constant.

1273 The electric power consumption of the electrolysis unit is:

$$P_{el,EU} = N_{Modules} \cdot N_{Stack} \cdot P_{el,Stack} + P_{aux} \quad (2.15)$$

1274 where $N_{Modules}$ is the modules number, N_{Stack} is the stack number per module, and
 1275 P_{aux} (kW) is the power adsorbed by the auxiliaries calculated as percentage of the
 1276 nominal power consumption.

1277 The specific electric consumption (kWh/kg) for hydrogen generation, that is a usual
 1278 performance index of an electrolyser, is calculated as:

$$E_{SC} = \frac{P_{el,EU}}{\dot{m}_{H2,EU} \cdot 3600} \quad (2.16)$$

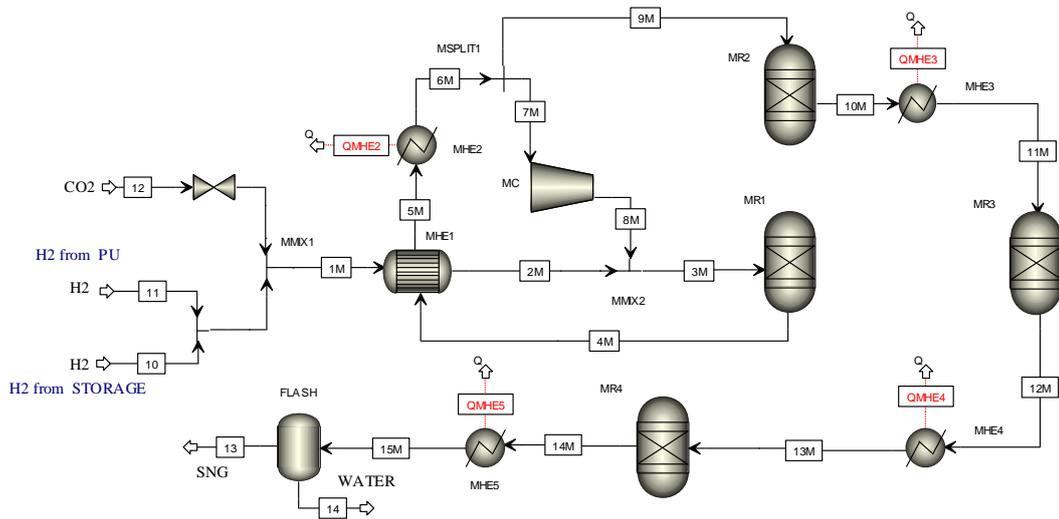
1279 where $\dot{m}_{H2,EU}$ is the hydrogen mass flow generated, expressed in kilogram per second.

1280 **2.3.1.3.2 Methanation Unit (METUNIT Hierarchy)**

1281 The parameters that control the CO₂ methanation are the temperature, the pressure,
 1282 and the feed gases ratio. As previously discussed, the methanation is favoured at
 1283 temperatures in the range of 200 – 550 °C and at high pressure. In order to obtain a
 1284 gas mixture with a high methane content, the ratio of the feed gases H₂/CO₂ must be
 1285 equal to the stoichiometric value according to the reaction R1. Moreover, experimental
 1286 data available in scientific literature [63,77] show how full chemical equilibrium is
 1287 obtained when operating the CO₂ methanation reaction above the stoichiometric ratio
 1288 of 4. As catalyst, nickel is considered to be the optimum choice due to its relatively

1289 high activity, good CH₄ selectivity, and low raw material price [46], but require a high
 1290 purity of the feeding gas.

1291 The flowsheet of the developed model of the methanation unit, similar to the
 1292 TREMP process, is shown in Figure 2.24.



1293

1294

Figure 2.24 – Flowsheet of the methanation unit model.

1295 The methanation is modelled by assuming the chemical equilibrium and the
 1296 adiabatic conditions in each chemical reactor. Thus, according to the plant layout
 1297 reported in Figure 2.18, the methanation unit mainly consists of four adiabatic reactors
 1298 (MR1, MR2, MR3, and MR4) connected in series with intermediate gas cooling (the
 1299 temperature of the streams exiting the heat exchangers MHE2, MHE3 and MHE4 is
 1300 set to 250 °C), a heat exchanger (MHE1) (for preheating the feed gases up to the MR1
 1301 inlet temperature of 250 °C), and a flash drum for water removal from the wet SNG.
 1302 Moreover, in order to control the temperature in the MR1, the product gas is partially
 1303 recirculated to it by means of a splitter operation block (the recirculation factor is 0.70).
 1304 The wet SNG leaving the MR4 reactor is cooled down to 229 °C (MHE5) before the
 1305 flash drum operation block where the 99.7% of the product water is removed.

1306 The chemical reactors are simulated using the RGibbs operation block, where the
 1307 chemical equilibrium of a given set of species is solved through the minimization of
 1308 the Gibbs free energy. This nonstoichiometric approach allows to find the equilibrium
 1309 composition when the reactions system is unknown or very complex. Moreover, the
 1310 hypothesis is that by reaching the chemical equilibrium the gas system can be formed
 1311 by H₂, CO, CO₂, H₂O, CH₄, and C_(s).

1312 Thermal power (QMHE2, QMHE3, QMHE4, and QMHE5) is also available by
 1313 recovering the cooling heat from the heat exchangers MHE2, MHE3, MHE4, and
 1314 MHE5.

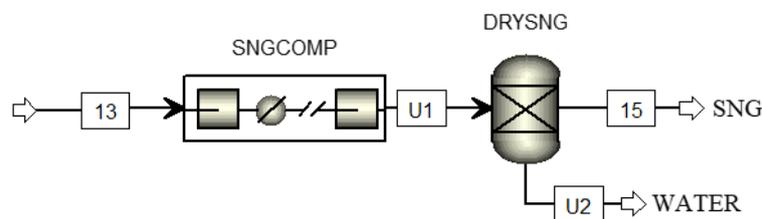
1315 The proposed model has been validated by means of the experimental data reported
 1316 in [66]. In this case, the feeding gas is a syngas coming from a biomass gasification
 1317 unit and only three methanation reactors are considered. The results of the model
 1318 validation of the methanation unit are reported in Appendix A.

1319 2.3.1.3.3 Storage Unit (STORAGE Hierarchy)

1320 The storage unit is modelled by means of mixer, splitter, and calculator block to
 1321 simulate the storage conditions in steady state. The stream STOR0 (see Figure 2.22)
 1322 is a virtual stream needed to close the mass balance that is equal to the hydrogen stored
 1323 in the tank. In a calculator block, the mass balance of the storage tank is implemented.

1324 2.3.1.3.4 SNG Upgrading Unit (SNGUP Hierarchy)

1325 Figure 2.25 shows the flowsheet of the SNG upgrading model. The dehydration
 1326 membrane is modelled by means of a separator operation block with an assigned
 1327 separation efficiency. The SNG compressor is a two-stage intercooled compressor.



1328

1329

Figure 2.25 – Flowsheet of SNG upgrading unit model.

1330 2.3.1.4 Storage control strategy

1331 The storage unit allows to decouple the dynamics of the PEM electrolysis unit from
 1332 that of the methanation unit. The parameter that controls the start and the operation
 1333 modes of the methanation unit is the hydrogen storage pressure. As management
 1334 strategy, three operation modes are allowed: (i) full load in which the hydrogen flow
 1335 rate is equal to the nominal value, (ii) partial load in which the hydrogen flow rate
 1336 varies linearly from 40% to 100% of the nominal value according to the specified
 1337 minimum and maximum storage pressures, and (iii) hot standby mode in which no

1338 hydrogen feeds the unit; in this operation mode, the heat losses of all the items of
1339 equipment have to be compensated by electrical heat tracing system [75,78].

1340 Thus, it is assumed that at the beginning of the operation, the storage tank is empty.
1341 According to the wind energy supply, the product hydrogen is sent to the storage unit
1342 until the pressure in the tank reaches the maximum value (maximum storage pressure,
1343 $p_{storage,max}$) and the tank is completely full. In this condition, the methanation unit starts
1344 at full load and the corresponding hydrogen flow rate is directly delivered from the
1345 electrolysis unit or from the storage unit or from both, depending on the electrical
1346 supply (i.e., if the product hydrogen is less than the required flow rate, the remaining
1347 is delivered from the storage). When the pressure in the storage tank decreases, the
1348 hydrogen flow rate sent to the methanation unit decreases too, and the operation mode
1349 is switched to partial load. This operation mode goes on until the storage pressure
1350 reaches a threshold value (shutdown pressure, $p_{storage,SD}$) beyond which the
1351 methanation unit is kept in hot standby.

1352 When the methanation unit is in hot standby, the hydrogen generated from the
1353 electrolysis is sent to the storage unit until the pressure rises a threshold value (restart
1354 pressure, $p_{storage,RS}$) beyond which the methanation unit restarts under partial load
1355 conditions. During the shutdown and restarting phases, the reactors are flushed with
1356 hydrogen (1/12 of the nominal flow rate for 1 h, as reported in ref. [74]).

1357 The pressure in the storage tank is calculated by applying the EoS for ideal gases:

$$p_{storage}^t = p_{storage}^{t-1} + \frac{(m_{H_2,EU}^t - m_{H_2,MU}^t) \cdot R_{H_2} \cdot T_{storage}}{V_{storage}} \quad (2.17)$$

1358 where $p_{storage}^t$ (Pa) and $p_{storage}^{t-1}$ (Pa) are the storage pressure at the time t (h) and t-
1359 1 (h), $m_{H_2,EU}^t$ (kg) is the hydrogen mass produced in one hour at the time t, $m_{H_2,MU}^t$
1360 (kg) is the hydrogen mass sent to the methanation unit in one hour at the time t, R_{H_2} is
1361 the hydrogen gas constant (J/kg·K), $T_{storage}$ (K) and $V_{storage}$ (m³) are the storage
1362 temperature and tank volume, respectively.

1363 The input parameters of the model are the storage capacity, expressed in kg of
1364 stored hydrogen, the storage temperature, the maximum and minimum storage
1365 pressures, the shutdown and restart pressure, and the hydrogen flow rate feeding the
1366 methanation unit at full load. In this study the storage capacity is chosen for assuring
1367 6.5 hours of full load operation of the methanation unit.

1368 The control strategy proposed has been implemented by using the MATLAB
1369 package software. In the Appendix A, the flow chart of the storage model is reported.

1370 2.3.2 Results and discussion

1371 2.3.2.1 Performance parameters

1372 The nominal efficiencies of the electrolysis unit and the methanation unit as well as
 1373 the overall nominal efficiency of the PtSNG plant are defined as the ratio between the
 1374 useful output powers (i.e., the chemical power of the hydrogen or the SNG) and the
 1375 input powers (i.e., the wind electric power or the chemical power of hydrogen).
 1376 Therefore, the nominal efficiency of the PEM electrolysis unit is:

$$\eta_{EU} = \frac{\Phi_{H_2,EU}}{P_{el,EU}} \quad (2.18)$$

1377 where $\Phi_{H_2,EU}$ is the chemical power of the produced hydrogen at nominal
 1378 conditions calculated on LHV basis, and $P_{el,EU}$ is the electric power consumed by the
 1379 electrolysis unit in DC. It is equal to the input renewable power multiplied by the
 1380 rectifier efficiency.

1381 The nominal efficiency of the methanation unit is:

$$\eta_{MU} = \frac{\Phi_{SNG,MU}}{\Phi_{H_2,MU} + P_{el,MU}} \quad (2.19)$$

1382 where $\Phi_{SNG,MU}$ and $\Phi_{H_2,MU}$ are the chemical powers of the hydrogen feeding and
 1383 the SNG exiting the methanation unit at nominal conditions and $P_{el,MU}$ is the electric
 1384 power consumption of recirculation blower.

1385 The thermal power available from the heat exchangers is not accounted in this
 1386 calculation.

1387 The nominal overall efficiency of the PtSNG plant results:

$$\eta_{PtSNG} = \frac{\Phi_{SNG}}{P_{el,RES} + P_{el,H_2} + P_{el,MU} + P_{el,SNG} - \Phi_{H_2,storage}} \quad (2.20)$$

1388 In the above equation, Φ_{SNG} is the SNG chemical power generated, $P_{el,RES}$ is the
 1389 wind power supply to the electrolysis unit in AC, P_{el,H_2} is the power consumption of
 1390 the hydrogen compressor, $P_{el,SNG}$ is the power consumption of the SNG compressor,
 1391 and $\Phi_{H_2,storage}$ is the hydrogen chemical power sent to the storage unit.

1392 Because of the fluctuating behaviour of the input energy source, the nominal
 1393 efficiencies are not sufficient to evaluate the effectiveness of the plant, but it is
 1394 necessary to evaluate the annual performance [79]. This is defined as the ratio of usable
 1395 system energy output to overall system energy input (electricity and possibly heat)

1396 [79,80]. Therefore, the annual efficiencies of the energy conversion units and the
 1397 annual overall efficiency are calculated as the nominal efficiencies, by replacing the
 1398 power with the energy (electric or chemical) consumed or generated in one year of
 1399 operation. However, the thermal energy required to maintain the plant equipment in
 1400 hot stand-by during the no production hours should be added as energy input:

$$\varepsilon_{EU} = \frac{E_{H_2,EU}}{E_{el,EU} + E_{th,EU,HS}} \quad (2.21)$$

$$\varepsilon_{MU} = \frac{E_{SNG,MU}}{E_{H_2,MU} + E_{el,MU} + E_{th,MU,HS}} \quad (2.22)$$

$$\varepsilon_{PtSNG} = \frac{E_{SNG}}{E_{el,RES} + E_{el,H_2} + E_{el,MU} + E_{el,SNG} - E_{H_2,storage} + E_{th,MU,HS} + E_{th,EU,HS}} \quad (2.23)$$

1401 where $E_{th,EU,HS}$ and $E_{th,MU,HS}$ are the thermal energy required to maintain system in
 1402 hot standby during the no production hours, and $E_{H_2,storage}$ is the chemical energy of
 1403 the remaining hydrogen in the storage tank.

1404 A useful performance parameter for evaluating the correct sizing of the plant is the
 1405 plant load factor. It compares the annual chemical energy production of the produced
 1406 SNG with the maximum production in one year:

$$PLF_{PtSNG} = \frac{E_{SNG}}{8760 \cdot \Phi_{SNG}} \quad (2.24)$$

1407 where Φ_{SNG} is the plant production at rated power.

1408 **2.3.2.2 Plants sizing and performance assessment at** 1409 **nominal conditions**

1410 The integrated model described in the previous section has been applied for
 1411 evaluating the mass and energy balance of the PtSNG plants at nominal conditions
 1412 (full load mode). Because of the same operating conditions, the plants differ only for
 1413 the input and output mass and energy fluxes.

1414 Table 2.9 summarizes the common operating data and parameters.

1415 The PEM electrolysis unit consists of modules of 1 MWDC that is the base module
 1416 for all plant sizes considered. The chosen nominal operating conditions, in terms of
 1417 current density and average cell voltage for each stack, are 2.99 A/cm² and 2.17 V,
 1418 respectively, and are derived from the polarization curve calculated by means of Eq.
 1419 (2.12). By fixing the cell area (1000 cm²) and the number of stacks per module (3), the

1420 number of cells per stack results equal to 55. The power consumption of the module
 1421 auxiliaries is set to 2.8% of the rated power. At nominal conditions, the hydrogen
 1422 generated by the module results equal to 16.8 kg/h with a specific energy consumption
 1423 of 59.6 kWh/kg of the produced hydrogen.

1424 The operating pressure of the methanation unit is set to 20 bar since higher
 1425 methanation pressures do not have a significant impact on the methane content in the
 1426 product gas mixture, as discussed in the paragraph 2.3.1.3.2. In order to contain the
 1427 hydrogen storage tank capacity, the methanation unit is sized for processing the
 1428 maximum hydrogen flow produced by the electrolysis unit. This choice makes the
 1429 decoupling between the dynamics of the electrolysis unit and that of the methanation
 1430 unit less effective but reduces the plant capital costs (based on the storage pressure of
 1431 50 bar, the cost is estimated 375 €/kg [74]).

1432

Table 2.9 – Common operating data of PtSNG plants.

Plant Sections and Components	
PEM electrolysis module	
AC/DC rectifier efficiency [%]	95
Stacks number x module	3
Cells number x stack	50
Active cell area [cm ²]	1000
Cell Temperature [°C]	55
Water utilization factor, UF	0.45
Average cell voltage at nominal power [V]	2.17
Current density at nominal power [A/cm ²]	2.99
Auxiliaries [% of rated power]	2.8
Inverter AC/DC efficiency [%]	95
Methanation Unit	
Pressure [bar]	20
H ₂ /CO ₂ [mol/mol]	4
Recycle ratio to the reactor MR1	0.70
MR1 exit temperature [°C]	561
MR2 exit temperature [°C]	412
MR3 exit temperature [°C]	301
MR4 exit temperature [°C]	260
Recycle compressor efficiency [%]	80
Hydrogen Compressor	

Pressure ratio	2.25
Isentropic efficiency [%]	70
Storage Unit	
Maximum Storage pressure [bar]	45
Minimum Storage pressure [bar]	22
Storage temperature [°C]	30
SNG Upgrading unit	
Dehydration membrane efficiency [%]	90
SNG compressor pressure ratio	1.75
Isentropic efficiency [%]	75

1433 Moreover, the hydrogen storage unit is designed for assuring the full load operation
 1434 of the methanation unit for about 6.5 h. The maximum storage pressure and the storage
 1435 temperature are assumed equal to 45 bar and 30 °C, while the minimum pressure in
 1436 the storage tank is set to 22 bar in order to allow the feeding of the methanation unit
 1437 without a compression system.

1438 In Table 2.10, the mass and energy balances of the considered PtSNG plant sizes
 1439 are reported. The electrolysis unit consists of 1, 3, and 6 modules and generates 16.8,
 1440 50.5, and 101.0 kg/s of hydrogen, respectively. The composition of the produced SNG
 1441 is 95.9% mol CH₄, 3.3% mol H₂, 0.8% mol CO₂, and traces of H₂O, and its low heating
 1442 value (LHV) is 49.2 MJ/kg, whereas the calculated Wobbe Index is 49.7 MJ/Nm³.
 1443 These values fill with the quality foreseen by the ongoing work on European
 1444 standardization of PtH₂ applications, for which most of the European natural gas
 1445 infrastructure can withstand a volume concentration 10% of hydrogen. Depending on
 1446 the electrolysis unit capacity (1, 3, or 6 MW), the SNG production is equal to 34.1,
 1447 102.4, and 204.9 kg/s, respectively, while the maximum amount of the stored hydrogen
 1448 is 110, 325, and 650 kg, respectively. By considering the storage conditions (45 bar
 1449 and 30 °C), the tank capacities are 30, 90, and 180 m³, respectively.

1450

Table 2.10 – Mass and energy balances under nominal conditions.

PtSNG plant capacity (electrical power input DC)	1 MW	3 MW	6 MW
Wind Energy supply [kW _{AC}]	1051	3155	6309
Plant Sections and Components			
PEM Electrolysis Unit (ELUNIT)			
Number of power modules	1	3	6
Water consumption [kg/h]	150.3	451.0	902.0

Hydrogen production [kg/h]	16.8	50.5	101.0
Thermal power consumption [kW]	5.8	17.4	34.8
Electric power consumption [kW _{DC}]	998.9	2996.8	5993.6
Methanation Unit (METUNIT)			
Carbon dioxide mass flow [kg/h]	91.9	275.7	551.3
Hydrogen mass flow [kg/h]	16.8	50.5	101.0
Thermal power from MHE2 [kW]	63.6	190.1	381.7
Thermal power from MHE3 [kW]	12.3	36.7	73.6
Thermal power from MHE4 [kW]	3.8	11.3	22.6
Thermal power from MHE5 [kW]	2.2	6.6	13.3
Recycle blower consumption [kW]	0.5	1.4	2.8
Raw SNG composition [mol %]	95.5 CH ₄ , 3.2 H ₂ , 0.8 CO ₂ , 0.5 H ₂ O		
Storage Unit (STORAGE)			
Storage capacity [kg of H ₂]	110	325	650
Storage volume at 30 °C [m ³]	30	90	180
SNG upgrading (SNGUP)			
Electric power consumption [kW]	1.1	3.4	6.8
SNG production [kg/h]	34.1	102.4	204.9
Plant Performances			
SNG composition [mol %]	95.9 CH ₄ , 3.3 H ₂ , 0.8 CO ₂ , traces H ₂ O		
SNG LHV/HHV [MJ/kg]	49.2/54.7		
Wobbe index [MJ/Nm ³]	49.7		

1451 2.3.2.3 Dynamic operation and annual mass and energy 1452 balance

1453 The time-domain simulation of the PtSNG system has been performed using
1454 MATLAB computing software. The annual energy balance of the PtSNG plant has
1455 been carried out by assuming that each electrolysis module can operate between the
1456 20% (200 kWh_{DC}) and 100% of its nominal load [78], while, as previously discussed,
1457 the minimum load of the methanation unit for assuring the quality of the produced
1458 SNG is 40%. The isentropic efficiencies of the BoP components and devices such as
1459 the hydrogen compressor and the SNG compressor have been assumed constant
1460 because of their very small impact on the total electric consumptions (about 0.1% of
1461 the electric consumption of the electrolysis unit).

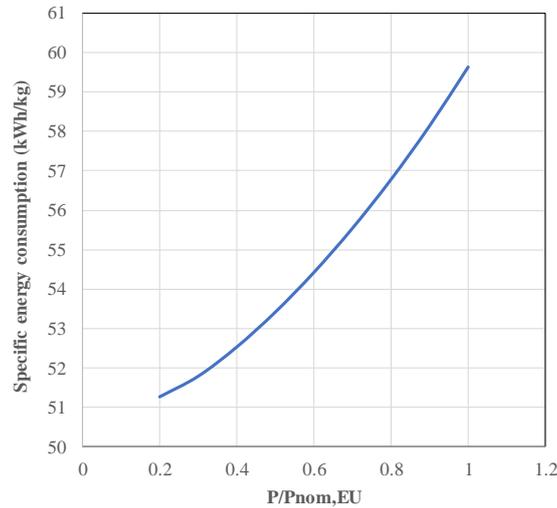
1462 The efficiency of the electrolysis unit varies with the power supplied as well as the
1463 hydrogen production. Using the specific energy consumption defined in Eq. (2.16) and

1464 reported in Figure 2.26, the hydrogen production (kg) as function of the supplied wind
 1465 energy (kWh) is obtained by the following equation:

$$M_{H_2} = \frac{E_{Wind,supplied} \cdot \eta_{AC/DC}}{E_{SC}} \quad (2.25)$$

1466 where $E_{Wind,supplied}$ (kWhAC) is the hourly electricity supplied to the electrolysis
 1467 unit (it depends on the stored energy factor) and $\eta_{AC/DC}$ is the rectifier efficiency.

1468 Therefore, by starting from the hourly electric energy generated by the wind farm
 1469 and by applying the integrated PtSNG model, the hourly hydrogen production as well
 1470 as the hourly mass and energy balances are calculated for each plant size. In Table
 1471 2.11, the annual operation time of the electrolysis and methanation units are reported
 1472 for the three plant sizes.



1473

1474 *Figure 2.26 – Specific energy consumption of the electrolysis module in the load range.*

1475

Table 2.11 – Annual operation time.

	1 MW		3 MW		6 MW	
Stored Energy Factor [%]	22%		52%		79%	
Plant section	EU	MU	EU	MU	EU	MU
Operation time [h]	5868	6047	5868	5004	5868	4170
Full load [h]	4372	2500	2813	1256	1646	486
Partial load [h]	1496	3547	3055	3748	4222	3684
Hot standby [h]	2892	2713	2892	3756	2892	4590
Shutdown	477	212	477	225	477	233
ON/OFF cycle	117	0	117	0	117	0

1476 The annual operation time of the EU unit results equal to 5868 h with 477 shutdown
 1477 and 117 ON/OFF cycles. While the annual operation time of the EU unit only depends
 1478 on the wind energy availability, the full load and partial load hours strictly depends on
 1479 the chosen size of the EU. Therefore, as expected (see Figure 2.19), the full load time
 1480 decreases from 4372 to 2813 h (-35.7%) and 1646 h (-63.2%). With respect to the
 1481 annual operation time of the methanation unit, the greatest value (6047 h) is obtained
 1482 for the PtSNG plant sized for 1 MW of energy input, whereas in the cases of 3 and 6
 1483 MW, it results smaller than the annual operation time of the electrolysis unit. In order
 1484 to increase the annual operation time of the methanation unit, a smaller hydrogen flow
 1485 rate at full load mode (i.e., the nominal hydrogen flow rate) has to be chosen,
 1486 involving, however, the need of higher capacities for the storage. The calculated
 1487 shutdowns are more than halved (212, 225, and 233), and the ON/OFF cycles are
 1488 suppressed.

1489 The flushed hydrogen flows are 641, 2142, and 4520 kg/year for 1, 3, and 6 MW,
 1490 respectively.

1491 Table 2.12 summarizes the annual mass and energy balance and the annual
 1492 efficiencies of the PtSNG plants.

1493 The annual overall efficiencies are decreasing with the plant size (from 43.7% to
 1494 41.6%), due to the growing impact of the energy required to balance the heat losses in
 1495 the hot standby mode (the standby period increases from 2713 to 4590 h). This may
 1496 suggest that, in the case in which the standby period is significant (this estimation
 1497 should be made in terms of forecasting), it could be better to keep the system in cold
 1498 standby, even if the start-up procedure use more energy depending on the duration of
 1499 the plant downtime.

1500 Finally, the plant load factor shows the greatest value (60.0%) in the case of 1 MW
 1501 plant capacity and the smallest (35.4%) for the 6 MW plant capacity.

1502 *Table 2.12 – Annual mass and energy balance.*

PtSNG plant capacity (electrical power input DC)	1 MW	3 MW	6 MW
Volumetric flow rates [10³ Nm³/year]			
Annual H ₂ O consumption	0.79	1.85	2.83
Annual H ₂ production	1000	2338	3573
Annual CO ₂ consumption	248	585	894
Annual H ₂ consumption	999	2334	3567
Annual SNG production	179	417	635
Energy [MWh/year]			

Annual electric energy consumption of the electrolysis unit (AC/DC)	5483/5209	12849/12207	19661/18768
Annual electric energy consumption for compressors and auxiliaries	15	37	62
Annual electric energy consumption for hot-standby*	106	458	1143
Annual SNG chemical energy (LHV basis)	2448	5705	8683
Chemical energy of the remaining hydrogen** (LHV basis)	4	10	17
Plant Performances			
Stored energy factor [%]	22.0	51.5	78.8
Electrolysis unit annual efficiency (LHV basis)	56.6	56.6	56.6
Methanation unit annual efficiency (LHV basis)	80.0	77.4	74.2
SNG Plant annual efficiency (LHV basis) [%]	43.7	42.8	41.6
Plant load factor [%]	60.0	46.5	35.4
*The thermal energy required to maintain the equipment in hot standby is calculated by using the data reported in [75] that considers electrical trace heating.			
**The remaining hydrogen in the storage tank is the difference between the annual hydrogen production and the annual hydrogen consumption			

1503 **2.4 Case study #3: Heat recovery from a PtSNG plant** 1504 ***coupled with wind energy***

1505 **2.4.1 Material and methods**

1506 **2.4.1.1 PtSNG Plant description**

1507 The PtSNG plant, shown in Figure 2.27, is directly coupled with a 12 MW wind
1508 farm and is sized for 1 MW of electric power input. It consists of four sections: i) the
1509 hydrogen generation unit based on the proton exchange membrane (PEM) electrolyser,
1510 ii) the methanation unit based on the fixed-bed reactors technology, iii) the H₂ storage
1511 unit, in which gaseous hydrogen is stored into pressurised vessels and iv) the SNG
1512 upgrading unit based on the membrane technologies.

1513 The demineralised water (1) is pumped and heated to the operative pressure and
1514 temperature of the PEM electrolysis unit where O₂ and H₂ are produced. On the basis
1515 of a specific control strategy, H₂ generated via electrolysis (5) can be partially or totally
1516 sent to the methanation unit (6) and/or to the intercooled compressor C1 (8) to increase
1517 the pressure up to the maximum storage pressure. The compressed H₂ (9) is stored in
1518 high-pressure tanks. The H₂ streams from electrolysis unit and/or storage unit are
1519 mixed with the CO₂ stream (13) from the CO₂ storage unit. The CO₂ is assumed to be
1520 always available for the PtSNG plant (i.e., supplied by sequestration from other power
1521 plants or industrial processes) at the storage pressure of 200 bar.

1522 The methanation unit consists of four adiabatic fixed-bed reactors (MR1, MR2,
1523 MR3, MR4) connected in series with intermediate gas cooling (heat exchangers
1524 MHE1, MHE2, MHE3) for assuring the fixed inlet reactors temperatures. The H₂ and
1525 the CO₂ entering the methanation unit are mixed with the hydrogen (19) from the SNG
1526 upgrading unit and then preheated up to the inlet temperature of the methanation
1527 reactor MR1 (300 °C). This preheating is performed by mixing the reactant flow (14)
1528 with hot streams from the MR1 and MHE1 defined by specified recirculation ratios
1529 (stream 3M at the splitter SPT1 and stream 8M at the splitter SPT2). Thus, the syngas
1530 (9M) sent to the MR2 reactor is a fraction of the product gases leaving the first
1531 methanation reactor. In the following reactors (MR3 and MR4) the CO₂ hydrogenation
1532 is led at decreasing temperature levels, reaching the conversion of 94% at the exit of
1533 the last methanation reactor. During the methanation process a large amount of water
1534 is produced, so the raw SNG (14M) is cooled in the heat exchanger MHE4 and sent to
1535 the FLASH unit for water removal. In order to meet the required grid specifications
1536 [81] the SNG (15) is further upgraded by means of a dehydration membrane (a
1537 commercial Pebax®-based membrane [72]) to eliminate the residual moisture to
1538 comply with the H₂O dew point requirements of the pipelines and of a H₂ separation
1539 membrane (polysulfone-based membrane).

1540 The quality of the SNG exiting from the upgrading section (20) satisfies the current
1541 pipelines specifications effective in various European countries, as reported in [81]
1542 and therefore, can be directly injected into the low-medium pressure NG transmission
1543 network. If the country specific pipeline requirements are more stringent, additional
1544 CO₂ or H₂ removal systems should be considered. The hydrogen from the upgrading
1545 unit is recycled to the first methanation reactor, as mentioned above.

1546 Finally, the waste heat available from the methanation unit and the flash unit can
1547 be usefully recovered in the waste heat recovery system according to concepts that will
1548 be described below.

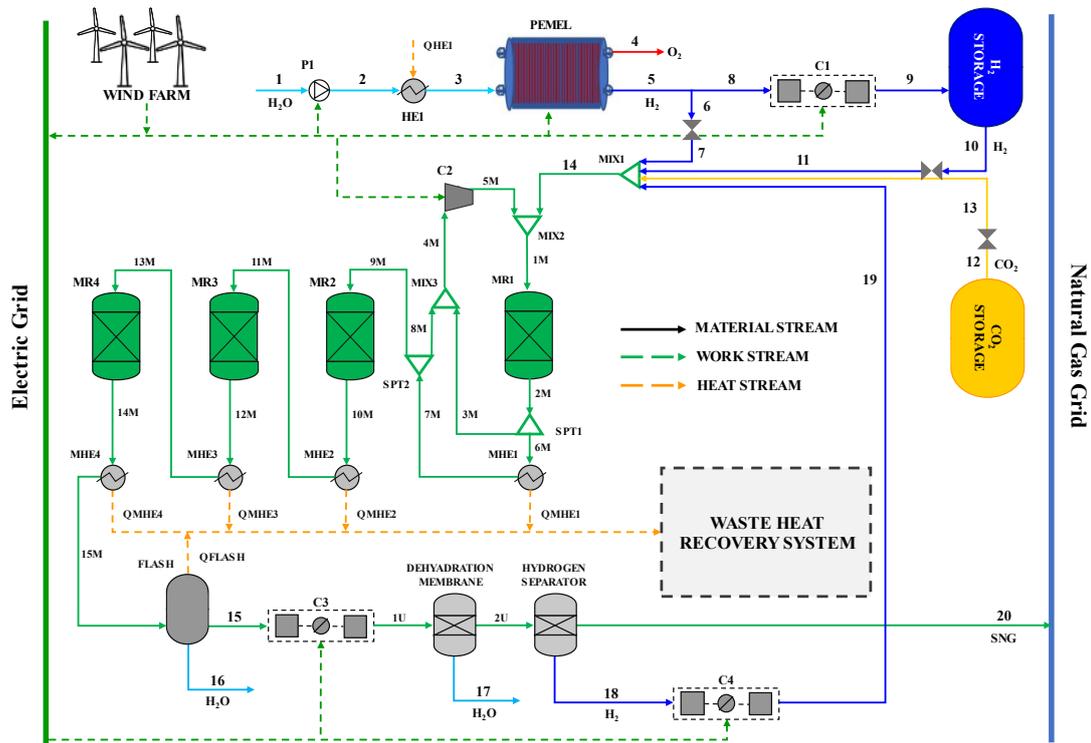


Figure 2.27 – PtSNG plant layout.

The PtSNG plant is operated in input-oriented mode. In order to maximise the operating hours of the plant, a hydrogen storage unit is employed to decouple the H₂ generation and the SNG production.

As management strategy, four operation modes are allowed for the methanation unit: i) full load, in which the hydrogen flow rate entering the methanation unit from the storage and/or the electrolyser is equal to the nominal value; ii) partial load, in which the hydrogen flow rate sent to methanation reactors varies from 40% to 100% of the nominal value, according to the specified minimum and maximum hydrogen storage pressures; iii) hot-standby mode, in which the reactors are kept at the minimum operating temperature of the catalysts to allow a quick restart; iv) cold-standby mode, in which no carbon dioxide and hydrogen feed the methanation unit. Even if the restart from cold-standby requires more energy than that required to maintain the hot-standby, this operation mode can be convenient when the number of hours from the shut-down to the restart is high. The parameter that controls the start of the plant and the operation modes of the methanation unit is the hydrogen storage pressure that follows the control strategy widely described in [82].

The mass and energy balances of the PtSNG plant in steady state conditions as well as the annual performance of the plant are performed by applying the thermochemical

1569 and electrochemical model (in Aspen Plus environment) and the dynamic model (in
1570 MATLAB environment) developed and described in [82].

1571 The thermochemical and electrochemical model consists of sub-models for each
1572 plant section; in this study the sub-model of the methanation unit, modified by
1573 considering the kinetics of the process, is described in Appendix B.

1574 The dynamic model based on the annual hourly energy input (the wind energy
1575 supplied to the plant) and on the storage control strategy allows forecasting the annual
1576 operation time (full and partial load hours, standby hours) and the annual performances
1577 (SNG production, energy consumption, efficiency) of the PtSNG plant.

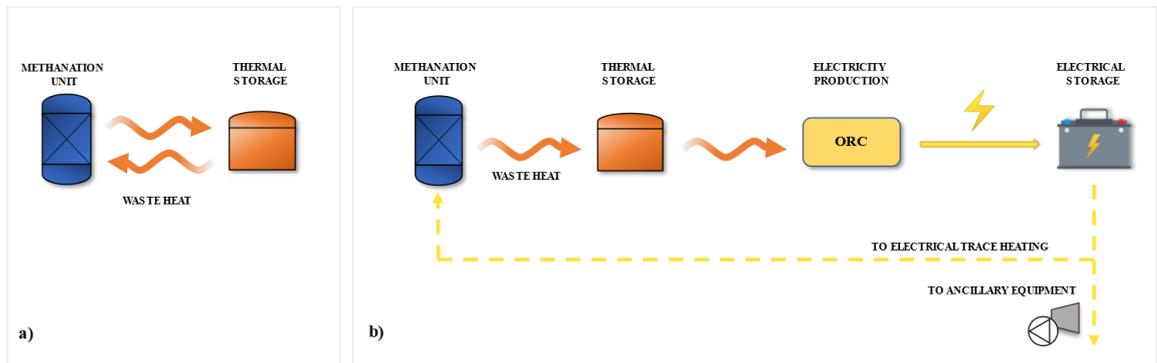
1578 **2.4.1.2 Heat recovery management**

1579 The heat recovery management is based on the valorisation of the waste heat,
1580 generated during the plant operation in the methanation unit, to self-sustain, from an
1581 energetic point of view, the standby conditions. To this end, two heat recovery
1582 configurations based on different heat recovery concepts are proposed and
1583 investigated. In both configurations, the recovery of the waste heat coming from the
1584 methanation unit is performed by means of a heat transfer fluid (i.e., diathermic oil)
1585 stored in insulated tanks for thermal energy storage.

1586 Figure 2.28 shows the conceptual schemes of the two heat recovery systems.

1587 In the first one (Thermal storage configuration, Figure 2.28.a), the stored thermal
1588 energy is used to satisfy the thermal energy demand of the reactors during the standby
1589 periods by means of a thermal carrier.

1590 In the second one (Electrical storage configuration, Figure 2.28.b), the stored
1591 thermal energy is used to produce electricity by means of an organic Rankine cycle
1592 (ORC). This electrical energy is stored in batteries, and it is used to meet the energy
1593 requirements of the electrical trace heating system of the methanation unit during the
1594 standby periods. In this case the stored electric energy is also used to satisfy the
1595 ancillary equipment electricity demand.



1596

1597 *Figure 2.28 – Conceptual schemes and energy flows of the two proposed waste heat recovery systems:*
 1598 *a) Thermal storage; b) Electrical storage.*

1599 In the first one (thermal storage configuration, Figure 2.28.a, the stored thermal
 1600 energy is used to satisfy the thermal energy demand of the reactors during the standby
 1601 periods by means of a thermal carrier.

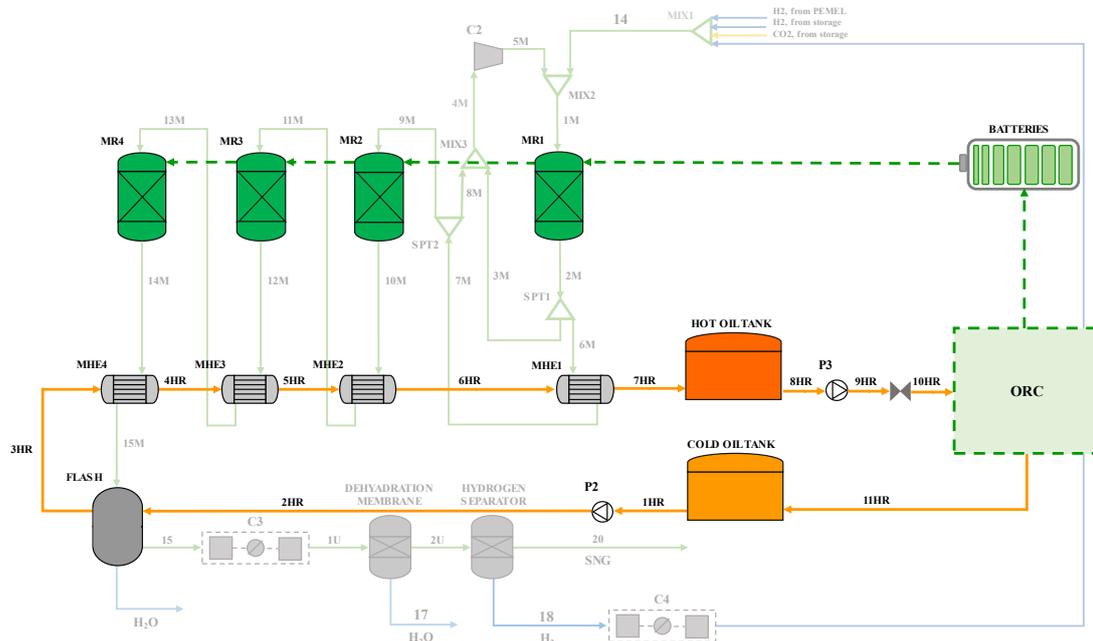
1602 In the second one (electrical storage configuration, Figure 2.28.b, the stored thermal
 1603 energy is used to produce electricity by means of an Organic Rankine Cycle (ORC).
 1604 This electrical energy is stored in batteries, and it is used to meet the energy
 1605 requirements of the electrical trace heating system of the methanation unit during the
 1606 standby periods. In this case, the stored electric energy is also used to satisfy the
 1607 ancillary equipment electricity demand.

1608 2.4.1.2.1 Thermal storage

1609 In the thermal storage system, the waste heat is recovered and stored by means of a
 1610 two-tanks diathermic oil circuit; the stored thermal energy is supplied, when required,
 1611 to the methanation reactors by means of a thermal carrier. Hydrogen is used as thermal
 1612 carrier because it removes the residual carbon dioxide preventing catalyst deactivation.
 1613 Figure 2.29 shows the layout of the thermal storage configuration.

1614 During the plant operation, the diathermic oil (1HR) from the cold oil tank is pushed
 1615 (2HR) by the pump P2 and sent to the methanation unit for heat recovering (heat
 1616 exchangers MHE1, MHE2, MHE3, MHE4 and the FLASH unit) in such a way to
 1617 minimise the temperature gradients and to optimise the heat transfer efficiencies; at
 1618 the exit of the MHE1 the hot diathermic oil (7HR) is stored into the hot oil tank.

1619 When the reactors need to be heated, the diathermic oil from the hot oil tank is sent
 1620 to the heat exchanger HRHE1 for heating the hydrogen used as thermal carrier. The
 1621 diathermic oil leaving the HRHE1 is cooled down in the HRHE2 to the storage



1640

1641

Figure 2.30 – Schematic diagram of electrical storage system.

1642

The diathermic oil circuit works as described in the thermal storage configuration.

1643

When the hot oil tank is full, the diathermic oil (8HR) is sent to the ORC unit for

1644

supplying the thermal power required by the ORC cycle. The returning cold oil (11HR)

1645

is then stored in the cold oil tank.

1646

With referring to the ORC power unit, by taking into account the heat sources

1647

temperatures as well as additional features (low toxicity, low global warming potential

1648

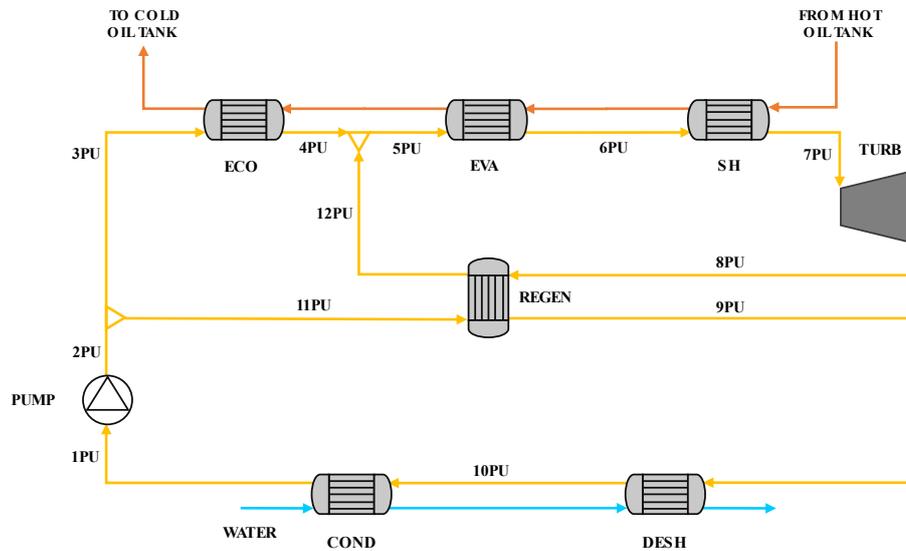
and ozone depletion potential, good compatibility and chemical stability in operation

1649

with other materials, etc.), the iso-butane (R600a) is selected as working fluid. In

1650

Figure 2.31, the functional scheme of the ORC power unit is shown.



1651

1652

Figure 2.31 – Functional scheme of the ORC.

1653 The organic fluid (1PU) leaving the condenser (COND) is pressurised up to the
 1654 maximum pressure of the cycle (2PU). In order to optimise the use of the heat from
 1655 the thermal storage and the heat internally available, it is split in two streams 3PU and
 1656 11PU. The first one (3PU) is sent to the economiser (ECO) where it is pre-heated up
 1657 to the saturation condition (4PU) by the diathermic oil, while the second one is pre-
 1658 heated (12PU) in the regenerator (REGEN) by the stream 8PU that exits the scroll
 1659 expander, that is still in superheated condition (the iso-butane is a dry fluid). The
 1660 streams 4PU and 12PU are mixed and the resulting is vaporised and superheated up to
 1661 the turbine inlet temperature by the diathermic oil. The superheated vapour (7PU)
 1662 expands in the scroll expander down to the condenser pressure. The desuperheating of
 1663 the iso-butane (9PU) is completed in the heat exchanger DESH where it is cooled down
 1664 to its dew-point temperature (10PU) and finally condensed in the condenser (COND).

1665 2.4.1.3 Heat recovery system sizing criteria

1666 Both proposed heat recovery configurations are sized for assuring the thermal
 1667 energy needed to maintain the hot-standby conditions of the methanation unit and the
 1668 thermal energy needed for the cold restart.

1669 The sizing procedure of the heat recovery systems is performed through the
 1670 evaluation of (i) the thermal energy that can be recovered from the methanation unit
 1671 (QMHE1, QMHE2, QMHE3, QMHE4) and the flash unit (QMHE5) and the
 1672 temperatures at which they are available; (ii) the thermal energy required by the
 1673 methanation unit in the hot-standby status and that required for the restart after a cold-

1674 standby; (iii) the optimal combination between the number of hot-standby hours and
1675 that of cold-standby.

1676 The thermal energy recoverable from the methanation unit and the flash unit as well
1677 as the temperatures at which it is available is assessed by the mass and energy balance
1678 of the plant.

1679 The thermal energy needed to maintain the hot-standby status (i.e., to maintain the
1680 reactors at the catalysts activation temperature) as well as that needed to warmup
1681 the reactors after a cold-standby (the thermal energy consumption to warm-up the
1682 reactors from ambient temperature to the catalysts activation temperature) has been
1683 calculated by considering convective and conductive heat transfer mechanisms.

1684 The standby hours are calculated by applying the dynamic model. The number of
1685 hours of hot-standby and, as a consequence, that of cold-standby, is evaluated by
1686 assuming the minimum capacity of the storage devices (the diathermic oil tanks for
1687 the thermal energy storage and the batteries for the electric energy storage) able to
1688 satisfy the energy requirements and, at the same time, to make the plant economically
1689 feasible. Thus, in order to identify the optimum combination between hot and cold-
1690 standby hours, statistical analysis of the plant's standby hours distribution is carried
1691 out in terms of the average duration and frequency. It is worth noting that this analysis
1692 is site-specific and closely linked to the typical annual wind energy profile of the
1693 considered plant, therefore tailor-made considerations would be required for other
1694 plants.

1695 **2.4.1.4 Performance parameters**

1696 The PtSNG plant nominal efficiency is defined as the ratio between the useful
1697 output powers (i.e., the chemical power of the SNG) and the input powers. The input
1698 powers for the PtSNG plant are represented by the electric power consumption for the
1699 electrolyser and the ancillary equipment, namely the hydrogen compressor for
1700 hydrogen storage, the blowers for the methanation unit and the raw SNG compressor
1701 and the hydrogen compressor for the upgrading unit. The efficiencies are calculated
1702 taking into account both the lower heating value (LHV) and the higher heating value
1703 (HHV) for the calculation of the chemical powers.

1704 The nominal overall efficiency of the PtSNG plant results in the following equation:

$$\eta_{PtSNG} = \frac{\Phi_{SNG}}{P_{el,RES} + P_{el,H2} + P_{el,MU} + P_{el,UU}} \quad (2.26)$$

1705 In the above equation, Φ_{SNG} is the chemical power of the generated SNG, $P_{el,RES}$ is
 1706 the wind power supply to the electrolysis unit in AC, $P_{el,H2}$ is the power consumption
 1707 of the hydrogen compressor for the H₂ storage, $P_{el,MU}$ is the electric power
 1708 consumption of recirculation blower in the methanation unit, $P_{el,UU}$ is the power
 1709 consumption of the upgrading unit, that includes the power consumption of the SNG
 1710 compressor and that of the hydrogen compressor downstream the hydrogen membrane
 1711 separation. The thermal power available from the methanation unit heat exchangers is
 1712 not accounted in this calculation.

1713 Because of the fluctuating behaviour of the input energy source, the nominal
 1714 efficiencies are not sufficient to evaluate the effectiveness of the plant, but it is
 1715 necessary to evaluate the annual performance [78]. This is defined as the ratio of the
 1716 usable system energy output to the overall system energy input (electricity and
 1717 possibly heat) [75,78]. Therefore, the overall annual efficiency of the PtSNG plant is
 1718 calculated as the nominal efficiency, by replacing the power with the energy (electric
 1719 or chemical) consumed or generated in one year of operation. However, in this case,
 1720 the thermal energy required to maintain the plant equipment in hot-standby during the
 1721 no production hours, as well as that required to restart the reactors after a cold-standby,
 1722 should be added as energy input:

$$\varepsilon_{PtSNG} = \frac{E_{SNG} + E_{H2,storage}}{E_{el,RES} + E_{el,H2} + E_{el,MU} + E_{el,UU} + E_{th,MU,HS} + E_{th,MU,CS}} \quad (2.27)$$

1723 where $E_{th,MU,HS}$ is the thermal energy required to maintain the methanation reactors in
 1724 hot-standby during the no production hours, $E_{th,MU,CS}$ is the thermal energy required
 1725 to warm up the reactors after a cold standby and $E_{H2,storage}$ is the chemical energy of
 1726 the remaining hydrogen in the storage tank. For the purposes of calculating the
 1727 efficiencies for the proposed configurations, having carried out the heat recovery
 1728 necessary to make the plant self-sufficient from the point of view of the energy
 1729 demands of the standby periods, the terms relating to the standby thermal requirements
 1730 ($E_{th,MU,HS}$ and $E_{th,MU,CS}$) are considered null. Indeed, these energy requirements are
 1731 satisfied by re-using internal energy.

1732 **2.4.2 Results and discussion**

1733 **2.4.2.1 Plant mass and energy balance results**

1734 Table 2.13 summarises the main operating parameters of the PtSNG plant. The 1
 1735 MWDC PEM electrolysis unit consists of three stacks with nominal operating
 1736 conditions, in terms of current density and average cell voltage for each stack, equal
 1737 to 2.99 A/cm² and 2.17 V, respectively. The power consumption of the unit auxiliaries
 1738 is set to 2.8% of the rated power. At nominal conditions, the hydrogen generated is
 1739 equal to 16.8 kg/h with a specific energy consumption of 59.6 kWh/kg of the produced
 1740 hydrogen. To contain the hydrogen storage tank capacity, the methanation unit is sized
 1741 for processing the maximum hydrogen flow generated by the electrolysis unit. With
 1742 this assumption, the decoupling between the dynamics of the electrolysis unit and that
 1743 of the methanation unit is less effective, but the plant capital costs (based on the storage
 1744 pressure of 50 bar, the cost is estimated at 375 €/kg [82]) are reduced. Furthermore,
 1745 the hydrogen storage unit is sized to ensure the full-load operation of the methanation
 1746 unit for about 6.5 h.

1747

Table 2.13 – Operational data of PtSNG plant.

Plant Sections and Components	
PEM Electrolysis Unit	
Nominal power [MW]	1
AC/DC rectifier efficiency [%]	95
Stacks number/Cells number per stack	3/50
Active cell area [cm ²]	1000
Cell pressure/temperature [bar/°C]	20/55
Water utilization factor, UF	0.45
Current density [A/cm ²] @Average cell voltage [V] at rated power	2.17
Auxiliaries [% of rated power]	2.8
H₂ Compressor (C1)	
Pressure ratio	2.25
Isentropic efficiency [%]	70
H₂ Storage Unit	
Maximum Storage pressure [bar]	45
Minimum Storage pressure [bar]	22
Storage temperature [°C]	30
Methanation Unit	
Pressure [bar]	10
H ₂ /CO ₂ [mol/mol]	4
Recycle ratio to MRI	0.70

Recycle ratio from SPT1 to MIX3	0.30
Recycle ratio from SPT2 to MIX3	0.57
MR1 in/out temperature [°C]	300/585
MR2 in/out temperature [°C]	300/468
MR3 in/out temperature [°C]	350/409
MR4 in/out temperature [°C]	350/365
Recycle blower (C2) isentropic efficiency [%]	85
SNG Upgrading Unit	
Dehydration membrane efficiency [%]	90
H ₂ separation membrane efficiency [%]	90
SNG compressor (C3) pressure ratio	1.75
H ₂ recycle compressor (C4) pressure ratio	10
Isentropic efficiency [%]	85

1748 In Table 2.14, the mass and energy balances under nominal conditions are reported.
 1749 The composition of the produced SNG is 92.5% mol CH₄, 2.2% mol H₂, 5.3% mol
 1750 CO₂ and traces of H₂O, and its lower heating value (LHV) is 43.4 MJ/kg, whereas the
 1751 calculated Wobbe Index is 43.9 MJ/Nm³. This value fits with the quality foreseen by
 1752 the ongoing work on European standardisation of PtH₂ applications, for which most
 1753 of the European natural gas infrastructure can withstand a volume concentration 10%
 1754 of hydrogen. The SNG production is equal to 38.52 kg/h while the maximum amount
 1755 of the stored hydrogen is 110 kg. By considering the storage conditions (45 bar and 30
 1756 °C), the tank capacity is 30 m³. The PtSNG plant efficiency on an LHV basis is 43.9%.

1757

Table 2.14 – Mass and energy balances under nominal conditions.

PtSNG Plant Capacity (Electrical Power Input) [MW_{DC}]	1 MW
Wind Energy supply [kW _{AC}]	1051
Plant Sections and Components	
Water Pump (P1)	
Water pump power consumption [kW]	0.11
PEM Electrolysis Unit	
Number of power modules	1
H ₂ O consumption [kg/h]	150.3
H ₂ production [kg/h]	16.8
Thermal power consumption [kW]	5.96
Electric power consumption [kW _{DC}]	998.9
H₂ Compressor (C1)	

Compressor power consumption [kW]	7.74
H₂ Storage Unit	
Storage capacity [kg of H ₂]	110
Storage volume [m ³] at 30 °C	30
Methanation Unit	
Carbon dioxide mass flow [kg/h]	96.8
H ₂ mass flow [kg/h]	17.74
H ₂ recycle mass flow [kg/h]	0.912
Recycle blower (C2) power consumption [kW]	0.34
Raw SNG composition [mol %]	76.7 CH ₄ , 18.5 H ₂ , 4.4 CO ₂ , 0.3 H ₂ O
SNG upgrading	
SNG compressor (C3) power consumption [kW]	3.05
H ₂ recycle compressor (C4) power consumption [kW]	1.08
Plant Performances	
SNG production [kg/h]	38.52
SNG composition [mol %]	92.5 CH ₄ , 2.2 H ₂ , 5.3 CO ₂ , traces H ₂ O
SNG LHV/HHV basis [MJ/kg]	43.4/48.2
SNG LHV/HHV [MJ/Nm ³]	30.5/33.9
SNG Plant nominal efficiency, LHV/HHV basis [%]	43.9/48.8

1758 Table 2.15 reports the thermal powers recoverable from the methanation unit and
 1759 the flash unit as well as the range of temperature at which they are available. It is worth
 1760 nothing that, due to the water condensation, a large amount of heat is available in the
 1761 flash unit, albeit at a low temperature. In heat exchanger MHE4, the thermal energy is
 1762 available at relatively low temperature levels compared to the other ones (the
 1763 temperature level increases towards MHE1).

1764 *Table 2.15 – Thermal powers available from the methanation unit and SNG inlet and outlet*
 1765 *temperatures in the heat exchangers.*

	MHE1	MHE2	MHE3	MHE4	FLASH
Q (kW)	58.1	9.9	4.8	15.8	59.1
T_{in} (°C)	585	468	409	365	158
T_{out} (°C)	300	350	350	158	25

1766 Thus, considering these temperature levels, the commercial Therminol-VP1, a
 1767 diathermic oil capable of operating at high temperatures (up to 400 °C), has been
 1768 chosen as heat transfer fluid for both the proposed heat recovery systems. The oil is

1769 maintained at a pressure of 10 bar to allow its operation at high temperatures while
1770 preventing the formation of vapours.

1771 Finally, the plant operation time is listed in Table 2.16.

1772 *Table 2.16 – Annual operation time of 1MW PtSNG plant.*

Operation time [h]	6047
Full load [h]	2500
Partial load [h]	3547
Standby [h]	2255
Transient [h]	458

1773 2.4.2.2 Thermal energy requirement

1774 The reactors are multitube type and are insulated by means of a microporous
1775 material with reinforcing filaments in pyrogenic silica (Steelflex-1100[®]) with a
1776 thermal conductivity at 300 °C of about 0.035 W/m·K and a thickness of 2 cm.

1777 Table 2.17 reports the thermal energy requirement (kWh/h) during the hot-standby
1778 and the thermal energy (kWh) required to restart the reactors after the cold-standby.
1779 The thermal energy required during the hot-standby has been calculated by assuming
1780 a minimum working temperature of 250 °C for each reactor (the minimum temperature
1781 for catalysts activation), while the thermal energy required to warm up the reactors
1782 after cold-standby, up to the catalysts' activation temperature, has been calculated by
1783 assuming 20 °C as the reference ambient temperature.

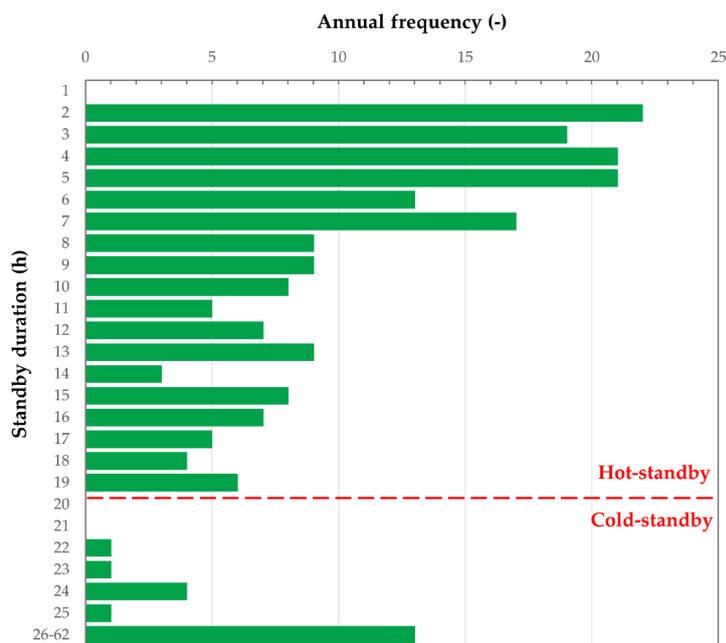
1784 *Table 2.17 – Thermal energy requirements of the methanation unit.*

	MHE1	MHE2	MHE3	MHE4	Total
Hot-standby [kWh/h]	1.01	0.74	0.74	0.56	3.06
Cold-standby [kWh]	7.78	4.60	3.18	1.94	17.79

1785 By comparing the total energy requirement of the two standby options due to
1786 reactors' characteristics, it can be noted that if no heat recovery system is coupled to
1787 the plant, hot-standby is cheaper than cold-standby if its duration is less than six
1788 consecutive hours. In this case, the energy consumption would be covered by energy
1789 coming from outside the plant, representing a cost. Furthermore, hot-standby is
1790 preferable to avoid thermal stress and for reasons of flexibility, i.e., for restarting SNG
1791 production as soon as conditions return favourable [83]. Moreover, it is clear that the
1792 amount of energy that has to be stored in the storage devices (the oil tanks in the case
1793 of the thermal storage configuration and the batteries in the case of electrical storage

1794 configuration) depends on the number of hot-standby hours, with the thermal energy
 1795 (17.79 kWh) required to restart the methanation unit from cold-standby being fixed.

1796 Thus, in order to identify the optimum combination between hot- and cold-standby
 1797 hours when the plant is coupled with energy recovery systems, statistical analysis on
 1798 the plant standby hours distribution is carried out in terms of the average duration and
 1799 frequency. It is worth noting that this analysis is site-specific and closely linked to the
 1800 typical annual wind energy profile of the considered RES facility, therefore tailor-
 1801 made considerations would be required for other plants. Figure 2.32 shows the
 1802 distribution of the standby periods in terms of the duration and frequency in the year.



1803

1804 *Figure 2.32 – Results of the statistical analysis and distribution of the standby periods.*

1805 It can be noted that most of the standby periods have a duration less than 20 h and
 1806 few occurrences have longer standby durations (the maximum is 62 h). By designing
 1807 the energy storage in an appropriate way, it is possible to extend the hot-standby
 1808 duration without requiring any additional external energy. Although extending the hot
 1809 standby for a greater number of hours results in higher energy consumption, in this
 1810 case, the latter is covered with recovered waste energy, therefore it does not represent
 1811 an operational cost. Thus, the duration of the standby periods and their frequency allow
 1812 one to identify a threshold value for the maximum desirable duration of the hot-
 1813 standby also useful for storage sizing. Beyond this threshold value, the control strategy
 1814 switches the operation mode of the methanation unit to cold-standby. In this case,

1815 when the plant is coupled with heat recovery systems, the selected threshold value is
1816 19 h, as reported in Table 6 that summarises the results of the statistical analysis.

1817 Finally, according to the data reported in Table 2.18, the total energy that has to be
1818 stored in the storage devices is equal to 75.9 kWh (58.1 kWh to cover the energy
1819 consumption of the 19 h hot-standby and 17.8 kWh to cover the energy consumption
1820 related to the restart from the cold-standby).

1821 *Table 2.18 – Results of the statistical analysis and distribution of the standby periods.*

Total number of methanation unit standby hours (h)	2255
Longest standby (h)	62
Total number of standby periods	213
Average annual standby duration (h)	10.6
Threshold value (h)	19
Number of hot-standby	193
Total hours of hot-standby (h)	1521
Number of cold-standby	20
Total hours of cold-standby (h)	734

1822 **2.4.2.2.1 Thermal storage results**

1823 Results of the thermal storage model show that the hydrogen flow rate needed to
1824 maintain the temperature of the methanation reactors at least at the minimum catalysts
1825 activation temperature (250 °C) is equal to 6.5 kg/h and its temperature at the inlet of
1826 the first reactor (MR1) is equal to 367 °C. The temperature of the hydrogen exiting the
1827 last reactor (MR4) is 250 °C (stream 15HR in Figure 2.29). Then, in order to ensure
1828 the cyclic operation, the hydrogen is re-heated by the diathermic oil from the hot oil
1829 tank (in the heat exchanger HRHE1) up to 367 °C.

1830 With reference to the two-tank oil circuit, the hot oil flow (8HR) is equal to 31.5
1831 kg/h and it is stored at 387 °C. Because the diathermic oil leaves the HRHE1 heat
1832 exchanger at 281 °C, it has to be cooled down to the storage temperature of the cold
1833 oil tank (30 °C). This cooling heat (from the HRHE2) can be further recovered for
1834 external or internal use.

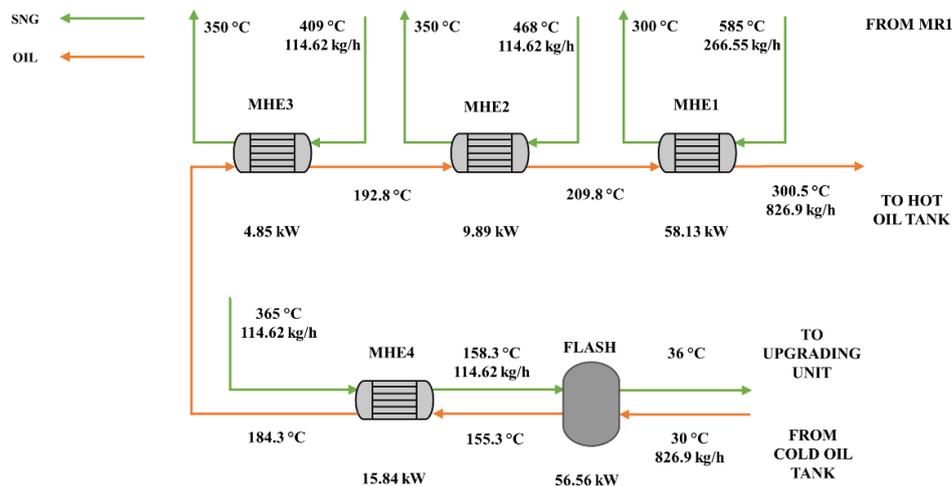
1835 The mass of diathermic oil necessary to store the required thermal energy (75.9
1836 kWh) is equal to 1067.3 kg (the specific heat is 2.416 kJ/kg·K) and the hot oil tank
1837 volume results in 1.50 m³ (the density at 387 °C is 713.2 kg/m³).

1838 **2.4.2.2.2 Electrical storage results**

1839 The ORC operating parameters have been chosen in order to maximise the waste
 1840 heat recovery and the ORC efficiency. These constraints are conflicting. In fact, to
 1841 maximise the waste heat recovery, a high flow rate of diathermic oil is required due to
 1842 a large amount of heat available from the flash unit, while a high turbine inlet
 1843 temperature, necessary for a high ORC efficiency, cannot be reached due to the
 1844 temperature of the hot oil, which decreases as its flow rate increases.

1845 Therefore, a compromise has been found by assuming an oil flow rate equal to 827
 1846 kg/h. The temperature of the hot oil exiting the last heat exchanger is about 300 °C.

1847 Figure 2.33 shows the energy balance of the oil circuit. The total amount of waste
 1848 heat recovered and supplied to the ORC is therefore equal to 145.2 kW, out of a total
 1849 of 148 kW that was recoverable from the methanation unit.



1850

1851 *Figure 2.33 – Diathermic oil circuit for waste heat recovery from the methanation unit, sized for*
 1852 *coupling with ORC power unit.*

1853 With respect to the operating parameters of the ORC cycle, the minimum and
 1854 maximum pressures and the inlet turbine temperature are set to 3 bar, 30 bar and 183
 1855 °C, respectively.

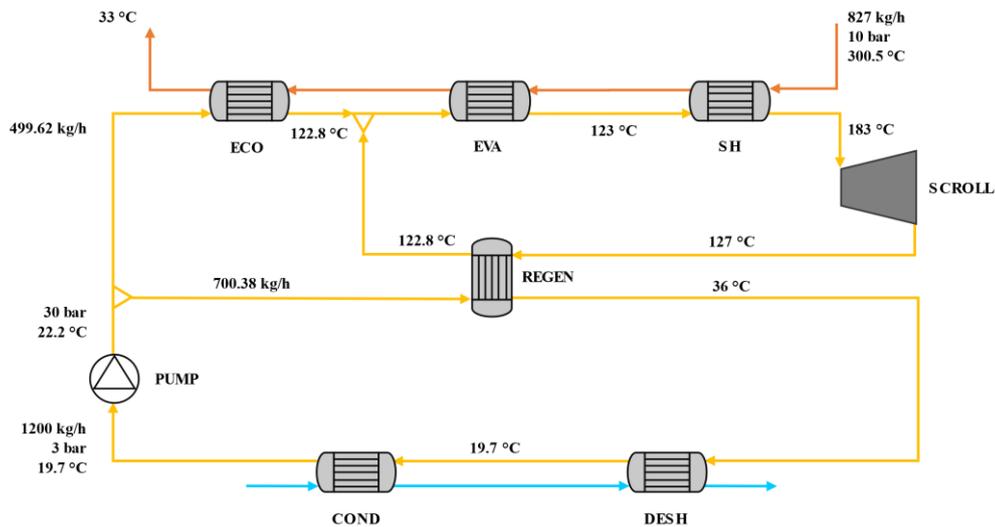
1856 Table 2.19 summarises the main operating conditions and performance of the ORC
 1857 unit. Detailed results concerning streams' temperatures, pressures and flow rates are
 1858 also given in Figure 2.34.

1859

Table 2.19 – Operating conditions and performance of ORC unit.

ORC Operating Parameter	
Isobutane flow rate [kg/h]	1200

Condenser pressure [bar]	3
Pressure ratio	10
Turbine inlet temperature [°C]	183
Split to ECO [kg/h]	499.6
Pump efficiency	0.70
Scroll expander efficiency	0.65
ORC Performance	
Compression power [kW]	2.47
Expansion power [kW]	26.57
Net mechanical power [kW]	24.1
Provided heat power [kW]	145.2
Overall ORC efficiency [%]	16.6



1860

1861

Figure 2.34 – Results for the ORC power unit

1862 Regarding the size of the oil tanks necessary to decouple the heat recovery from the
 1863 electricity generation allowing more continuous operation of the ORC, the thermal
 1864 energy stored is 435.6 kWh (3 h of operation in full load). Therefore, 2481 kg of
 1865 diathermic oil is stored at 300 °C. Since the diathermic oil density at 300 °C is 817
 1866 kg/m³, the volume of the hot oil tank results equal approximately 3 m³.

1867 Referring to the battery sizing, by assuming an electrical efficiency equal to 0.95,
 1868 the battery is designed for a capacity of 80 kWh. When the methanation unit is in
 1869 operation and the ORC produces electrical energy, part of this is used for plant
 1870 auxiliaries (12.3 kWh at full load operation for 1 h) so that the electrical energy useful

1871 to charge the batteries is equal to 11.8 kWh, resulting in a battery charge time of 6.8
1872 h.

1873 2.4.2.3 Annual performance results

1874 Table 2.20 reports the annual energy balance as well as the annual performances of
1875 the PtSNG plant. The annual water, hydrogen and carbon dioxide consumptions are
1876 equal to 790 Nm³, 999,000 Nm³ and 261,286 Nm³, respectively. The SNG production
1877 is equal to 265,287 Nm³.

1878 *Table 2.20 – Annual mass and energy balance and annual performances.*

PtSNG Plant Capacity (Electrical Power Input DC)		1 MW	
Energy [MWh/year]			
Annual electric energy consumption of the electrolysis unit (AC/DC)		5483/5209	
Annual electric energy consumption for compressors and auxiliaries		29.8	
Annual energy consumption for hot-standby		4.7	
Annual energy consumption for cold-standby		0.36	
Annual SNG chemical energy (LHV basis)		2424	
Chemical energy of the remaining hydrogen * (LHV basis)		3.6	
Plant Performances			
<i>Waste Heat recovery configuration</i>		<i>Thermal storage</i>	<i>Electrical storage</i>
PtSNG Plant annual efficiency, LHV basis [%]		44.0	44.3
PtSNG Plant annual efficiency, HHV basis [%]		48.9	49.2

*The remaining hydrogen in the storage tank is the difference between the annual hydrogen production and the annual hydrogen consumption.

1879 From the data in Table 2.20, it can be noted that the energy consumption for hot
1880 and cold standby states (5.06 MWh/year in total) is only a small percentage (0.09%)
1881 of the total energy required to operate the plant (electric energy in AC for the
1882 electrolyser and that for ancillary equipment).

1883 As regards the annual efficiencies, the PtSNG plant in the base case, without any
1884 heat recovery systems, shows efficiencies of 43.9% on an LHV basis and 48.8% on an
1885 HHV basis. The results for the thermal storage and electrical storage configurations
1886 show that the energy recovery intended to satisfy internal consumption only is
1887 beneficial for a PtSNG plant, albeit slightly. Compared to the base case (in which no
1888 energy recovery is carried out), the increase in the annual efficiency was proportional
1889 to the annual energy savings achieved by the two proposed systems, equal to 5.06
1890 MWh/year and 34.86 MWh/year for the thermal storage and the electrical storage,

1891 respectively. In the base case, it would be reasonable to expect a decrease in the annual
1892 efficiency with respect to the nominal efficiency, due to shutdowns and the fluctuating
1893 plant behaviour. However, the annual efficiency remains close to the nominal
1894 efficiency values in the case in which waste energy is recovered for internal uses.
1895 Moreover, the electrical storage configuration shows a slightly higher annual
1896 efficiency with respect to the thermal storage configuration because in the first case,
1897 the ancillary equipment power consumption is also satisfied by the waste energy
1898 recovered. If the surplus electricity produced by the ORC was also valued or sold
1899 externally, slightly higher annual efficiencies could be achieved.

1900 A rough discussion on costs is also given. In principle, the proposed energy
1901 recovery systems should have very low operating expenditures, linked mainly to their
1902 maintenance as they are operated with waste energy, while the capital costs should be
1903 carefully evaluated. This will be the subject of future investigation in greater detail.
1904 Capital expenditures are expected to be low in the case of thermal storage (purchase
1905 of diathermic oil tanks and heat transfer fluid), and higher in the case of electrical
1906 storage (higher volumes of diathermic oil, ORC and batteries). However, electrical
1907 storage could lead to greater annual savings, avoiding the expense of electricity
1908 supplied to ancillary equipment, and providing a faster return on investment. In reality,
1909 considering the heat recovery only to satisfy the internal demands of the plant is
1910 probably not convenient, but it can become interesting if one considers the possibility
1911 of being able to sell all the recoverable waste energy, whether it is heat (for example
1912 for district heating) or electricity that can be sold to the grid. Meeting the internal
1913 energy demand of the plant with recovered waste energy could also lead to
1914 environmental benefits, where previously the additional external energy was provided
1915 by the electric grid (affected by the energy mix) or by natural gas.

1916 Finally, it is emphasised that in this work, the heat recovery systems have been
1917 designed only to ensure the self-sustenance of the plant regarding the standby periods
1918 and ancillary equipment energy requirements. However, the plant internal energy
1919 consumption is significantly lower than the waste energy available from the
1920 methanation process. A remarkable further improvement of the PtSNG plant
1921 performance can be reached if this surplus is employed for external thermal or
1922 electrical utilities.

1923

1924 **CHAPTER 3. ANALYSIS OF IMPACT OF**
1925 **HYDROGEN INJECTION NATURAL GAS**
1926 **NETWORKS**

1927 **3.1 Challenges to injecting hydrogen into natural gas**
1928 **networks**

1929 In 2020, the European Commission (EC) published the European Union (EU)
1930 hydrogen strategy on the roadmap to develop an integrated energy system based on
1931 the large-scale hydrogen supply chain [84]. The EU hydrogen strategy has the twofold
1932 objective of expanding hydrogen use to replace fossil fuels and to decarbonize its
1933 production. In detail, the path set by the EC is divided into three phases. The first phase
1934 entails the installation of at least 6 GW of renewable hydrogen electrolyzers in the EU
1935 by 2024 and producing up to one million tonnes of renewable hydrogen. The second
1936 one relies on the installation of at least another 40 GW of renewable hydrogen
1937 electrolyzers by 2030 and the production of up to 10 million tons of renewable
1938 hydrogen in the EU. The third and last phase aims to deploy large scale renewable
1939 hydrogen technologies to reach all hard-to-decarbonise sectors by 2050.

1940 To achieve such goals, it will be effective to consider the reuse of the existing
1941 natural gas (NG) infrastructure. The latter will play a crucial role in the development
1942 of a decarbonized energy system based on a large usage of hydrogen as energy carrier
1943 due to its widespread presence and its capacity to provide a cost-effective option for
1944 transporting and storing large amounts of energy for long-term periods, exploiting the
1945 NG transportation and distribution networks, as well as the storage complexes of the
1946 existing NG infrastructure.

1947 On the other hand, as discussed in the previous chapter, the increasing weight of
1948 the non-programmable RESs is currently leading to relevant issues on the management
1949 of the electricity grid, owing to their intermittent and random nature. The storage
1950 capacities required to guarantee the stability and flexibility of the electrical energy

1951 system can be ensured by the application of PtG technologies [2]. In fact, the hydrogen
1952 produced from non-programmable RESs during overgeneration periods can be
1953 injected into NG pipelines, thus overcoming balancing and regulation issues related to
1954 the electricity grid management, as well as decarbonizing end-use sectors (e.g.,
1955 residential, urban and public transport, industrial). Thus, the blending of hydrogen
1956 produced from RESs into NG mixtures nowadays represents an effective tool for
1957 developing a climate neutral integrated energy system.

1958 Nevertheless, several factors must be considered to assess the possibility of using
1959 mixtures of hydrogen and natural gas (H₂NG) into existing network components (e.g.,
1960 pipelines, valves, compressor, measuring systems). Indeed, increasing content of H₂
1961 lead to significant changes in physical and thermal properties of the flowing gaseous
1962 mixture. Methane has greater density, viscosity, calorific value by volume and
1963 solubility in water compared to hydrogen; on the contrary, hydrogen has a higher heat
1964 capacity, diffusivity, calorific value by mass, maximum flame temperature,
1965 autoignition temperature and wider explosive and fire danger. Moreover, hydrogen has
1966 a lower density and viscosity than methane and air at normal temperature and pressure
1967 [85].

1968 The changes in thermodynamic properties, in turn, result in changes in combustion
1969 and calorific properties. Therefore, the addition of hydrogen could originate minor or
1970 major issues depending on the volumetric percentage of hydrogen added. Some of the
1971 key issues related to blending hydrogen into natural gas pipeline are the: i) durability
1972 of the natural gas network components and the integrity of pipelines, ii) impact on the
1973 energy capacity of the gas supply system and gaseous leaks, iii) implication on safety
1974 of the transmission, distribution and use of the gaseous mixture, iv) impact on end-use
1975 appliance and v) effect on measuring systems. In general, only minor criticalities occur
1976 with blends of less than 5% – 15% hydrogen by volume, depending on site specific
1977 conditions and particular natural gas mixture compositions. Therefore, it is agreed that
1978 extensive studies and tests of pipelines, compressor, valves, measuring devices and
1979 appliances are needed [86].

1980 In the following paragraphs, an overview on the effect of hydrogen injection in
1981 natural gas measurements is presented. After the literature review on analysis of the
1982 effects of hydrogen injection on natural gas networks measurement, the effect of
1983 hydrogen injection on gas measurements, both from a theoretical and an experimental
1984 point of view is analysed as case study. In detail, a comprehensive analysis of the
1985 changes of thermo-physical properties when gases with different H₂ content are used
1986 has been carried out, as well as of the theoretical effects on the accuracy of volume

1987 conversion at standard conditions. The impact of the hydrogen injection on the
1988 accuracy of static ultrasonic domestic gas meters has also been experimentally
1989 investigated and the related results have been presented and discussed.

1990 **3.2 Literature review on analysis of the effects of** 1991 **injection on natural networks**

1992 Several studies have been devoted to the analysis of the effects of H₂NG mixtures
1993 on the reliability of existing measurement systems. In [87] the changes in the properties
1994 of different gas types of the "H" group of EN 437:2019 were investigated,
1995 demonstrating all gases remained in group "H" when hydrogen is injected into natural
1996 gas with a content ranging from 2% to 23%. In particular, the thermo-physical
1997 properties of the mixtures vary as follows: i) the speed of sound increases in the range
1998 1 – 13.5%; ii) the density (relative density) decreases in the range 1.7 – 20.5%; iii) the
1999 higher calorific value decreases in the range 1.4 – 16%; iv) the Wobbe index decreases
2000 in the range 0.49 – 5.7%. In conclusion, it is affirmed that the injection of more than
2001 10% hydrogen to natural gas allows to manage existing gas transportation and gas
2002 consumption systems without the need for reconstructions, modifications, changes in
2003 the algorithms for calculating physical parameters of the gas mixture and volumetric
2004 flow (volume); on the other hand, further studies and the development of new
2005 algorithms are needed for higher hydrogen contents. In [88] no problems were
2006 recorded relating to the materials used in the construction of ultrasonic, membrane and
2007 turbine gas meters for hydrogen up to 30%. In particular, diaphragm, turbine and
2008 ultrasonic gas meters should be insensitive to this effect. However, they encourage
2009 field tests to determine the measurement accuracy of gas meters with H₂NG mixtures,
2010 as well as long-term metrological stability. Iskov [89] analyse the medium-term effects
2011 (about one year) of 100% hydrogen on the metrological properties of traditional
2012 turbine, rotary piston and diaphragm gas meters at a pressure of 20 mbar, finding that
2013 hydrogen had no effect on the metrological characteristics of the gas meters tested,
2014 despite some issues with the turbine meter seal. Dehaeseleer [90] demonstrated that
2015 there are no operational issues with traditional gas meters (turbine, rotary piston and
2016 diaphragm) up to 10% of the hydrogen content in natural gas, however, above this
2017 value a new calibration could be necessary, together with the evaluation of changes to
2018 the design of the meters. Jaworski [91] experimentally evaluated the durability of
2019 diaphragm gas meters with hydrogen, demonstrating they are not significantly affected
2020 by injection of hydrogen up to 15%. The long-term drift has also been related to the

2021 wear of the internal parts of the meter, rather than to the hydrogen content in the gas
2022 mixture. In addition, no damage was found that could represent a threat to operational
2023 safety as all gas meters tested maintained unaltered tightness during and after the test.
2024 Jaworski and Dudek [92] found capillary thermal mass gas meters comply with
2025 accuracy limits and only occasional values exceed the maximum permissible errors
2026 (MPE) for H2NG with 2, 4 and 5% of hydrogen; on the other hand, with hydrogen
2027 content of 10 and 15% they significantly exceeded the MPEs with error curves
2028 systematically shifted towards negative errors (e.g., up to -3.4% and -8.4% of the
2029 Weighted Mean Error (WME), at 10% and 15% of H₂, respectively).

2030 No experimental evidences are available in the literature against the injection of H₂
2031 up to 10% for ultrasonic meters, whereas for higher levels further experimental
2032 research is required [47]. Finally, as regards the effects on the volume conversion,
2033 Łach [93] investigates the effects of H2NG on the calculation of the compressibility
2034 factor. To this end, tests were carried out in the laboratory of the Polish National
2035 Research Oil and Gas Institute to compare SGERG-88 and AGA8-DC92 methods to
2036 determine the compressibility coefficient of gas with an increasing hydrogen content.
2037 In fact, both methods have application limitations in terms of hydrogen content in
2038 natural gas up to 10% (mol/mol). Research has shown that the AGA8-DC92 method
2039 provides satisfactory results when the hydrogen content is less than 40% (mol/mol).

2040 ***3.3 Case study #4: Impact of hydrogen injection on*** 2041 ***natural gas networks measurement***

2042 **3.3.1 Theory and methods**

2043 **3.3.1.1 Thermodynamic properties**

2044 The density of the real gaseous mixture at metering reference conditions is obtained
2045 using the ISO 6976:2017 [94] formula:

$$D(p, t) = \frac{D^0}{Z(p, t)} \quad (3.1)$$

2046 where : i) $Z(p, t)$ is the compressibility factor of the gaseous mixture calculated
2047 conforming to the ISO 6976:2017 standard [94]; ii) D^0 is the density of the ideal
2048 gaseous mixture at the metering reference conditions evaluated through the following
2049 equation:

$$D^0(p, t) = \frac{M}{V^0} \quad (3.2)$$

2050 where: i) V^0 is the ideal molar volume of the gaseous mixture evaluated according to
2051 the ideal gas law; ii) M is the molar mass of the gas mixture calculated from:

$$M = \sum_{j=1}^N x_j M_j \quad (3.3)$$

2052 where: i) x_j is the mole fraction of the j -th component in the gas mixture; ii) M_j is the
2053 molar mass of the j -th component.

2054 The relative density is defined as the ratio of the density of the gaseous mixture to
2055 the density of dry air of reference composition at the same specified conditions of
2056 pressure and temperature. It has been calculated in accordance with the ISO 6976:2017
2057 [94] formula for the real gaseous mixture:

$$G(p, t) = \frac{G^0 \cdot Z_{air}(p, t)}{Z(p, t)} \quad (3.4)$$

2058 where: i) $Z_{air}(p, t)$ is the compressibility factor of dry air of reference composition;
2059 ii) $Z(p, t)$ is the compressibility factor of the gaseous mixture calculated in accordance
2060 with the ISO 6976:2017 [94]; iii) G^0 is the relative density of the ideal gas, which is
2061 evaluated through the formula below:

$$G^0 = \sum_{j=1}^N x_j \frac{M_j}{M_{air}} \quad (3.5)$$

2062 where: i) x_j is the mole fraction of the j -th component in the gas mixture; ii) M_j is the
2063 molar mass of the j -th component; iii) M_{air} is the molar mass of dry air of reference
2064 composition.

2065 The specific heat capacity has been estimated employing the correlations depending
2066 on the temperature and gaseous mixture composition available in literature. Namely,
2067 the specific heat capacity at constant pressure has been evaluated as the weighted
2068 average of the specific heat capacities at constant pressure of the components
2069 constituting the gaseous mixture:

$$c_p(t_1) = \sum_{j=1}^N x_j \cdot c_{p,j} \quad (3.6)$$

2070 where: i) x_j is the mass fraction of the j -th component in the gas mixture; ii) $c_{p,j}$ is the
 2071 specific heat capacity at constant pressure of the j -th component calculated according
 2072 to the equation in [95]:

$$c_{p,j}(T) = A_j + B_j T + C_j T^2 + D_j T^3 + E_j T^4 \quad (3.7)$$

2073 where i) T is the absolute temperature; ii) A_j, B_j, C_j, D_j, E_j are the regression coefficients
 2074 for the j -th component of the gaseous mixture, reported in Table 3.1.

2075 *Table 3.1 – Regression coefficients for chemical compound [96].*

Component, j	Temperature range [K]	A	B	C	D	E
Methane, CH ₄	50-1500	34,942	-3,9957E-02	1,9184E-04	-1,5303E-07	3,9321E-11
Ethane, C ₂ H ₆	100-1500	28,146	4,3447E-02	1,8946E-04	-1,9082E-07	5,3349E-11
Propane, C ₃ H ₈	200-1000	31,986	4,27E-02	5,00E-04	-6,56E-07	2,56E-10
n-Butane, n-C ₄ H ₁₀	200-1500	20,056	2,8153E-01	-1,3143E-05	-9,4571E-08	3,4149E-11
iso-Butane, iso-C ₄ H ₁₀	200-1500	6,772	3,1447E-01	-1,0271E-04	-3,6849E-08	2,0429E-11
n-Pentane, n-C ₅ H ₁₂	200-1500	26,671	3,2324E-01	4,2820E-05	-1,6639E-07	5,6036E-11
iso-Pentane, iso-C ₅ H ₁₂	200-1000	16,288	3,1754E-01	2,0237E-04	-4,3027E-07	1,8001E-10
n-Hexane, C ₆ H ₁₄	200-1500	25,924	4,1927E-01	-1,2491E-05	-1,5592E-07	5,8784E-11
Nitrogen, N ₂	50-1500	29,342	-3,5395E-03	1,0076E-05	-4,3116E-09	2,5935E-13
Carbon dioxide, CO ₂	50-5000	27,437	4,2315E-02	-1,9555E-05	3,9968E-09	-2,9872E-13
Hydrogen, H ₂	250-1500	25,399	2,0178E-02	-3,8549E-05	3,1880E-08	-8,7585E-12

2076

2077 The calculation of the higher calorific values $(Hv)_G^0$ on volume basis, at a
 2078 temperature t , has been carried out basing on the ISO 6976:2017 standard [94] through
 2079 the following equations:

$$(Hv)_G^0(p, t) = \frac{(Hc)_G^0(t)}{V^0} \quad (3.8)$$

2080 where: i) $(Hc)_G^0(t)$ is the ideal higher molar-basis calorific value; ii) V^0 is the ideal
 2081 molar volume of the mixture calculated according to the ideal gas law.

2082 The Wobbe Index is defined as the ratio of the higher calorific value, at specified
 2083 reference condition, to the square root of the relative density. Conforming to the ISO
 2084 6976:2017 [94], it has been calculated from the following formula:

$$W_G^0(t, p) = \frac{(Hv)_G^0(t, p)}{\sqrt{G^0}} \quad (3.9)$$

2085 where: i) $(Hv)_G^0$ is the higher calorific value on volume basis of the ideal mixture;
 2086 ii) G^0 is the relative density the ideal mixture.

2087 The speed of sound in a gaseous flow can be estimated using the AGA 10 formula
 2088 [97]:

$$u = 18.591 \left(\frac{T \cdot Z \cdot k}{\rho_s} \right)^{0.5} \quad (3.10)$$

2089 where: i) T is the absolute temperature; ii) Z is the compressibility factor of the gaseous
 2090 mixture calculated employing the ISO 6976:2017 standard [94]; iii) k is the adiabatic
 2091 coefficient of the gaseous mixture (e.g. the ratio between heat capacities at constant
 2092 pressure and volume, respectively); iv) ρ_s is the density of the gas mixture at reference
 2093 conditions.

2094 3.3.1.2 Volume conversion factor

2095 The accurate knowledge of NG volumetric flow rate is crucial in commercial
 2096 transactions and balancing issues of the NG infrastructures. To this aim, the NG
 2097 mixture volume measured at operative conditions is commonly converted to the
 2098 standard reference conditions (i.e., $p_S=101.325$ kPa, $T_S=15$ °C). Such volume
 2099 conversion is obtained through the following equation:

$$V_S = V \cdot KTvo \quad (3.11)$$

2100 where: i) V_S and V are the NG mixture volume at reference and metering conditions,
 2101 respectively; ii) $KTvo$ is the volumetric correction factor calculated using the formula
 2102 below:

$$KTvo = \frac{p}{p_S} \cdot \frac{T_S}{T} \cdot \frac{Z_S}{Z} \quad (3.12)$$

2103 where: i) p_S and p are the absolute pressure at reference and metering conditions,
 2104 respectively; ii) T_S and T are the temperature at reference and metering conditions,
 2105 respectively; iii) Z_S and Z are the compressibility factor calculated at reference and
 2106 metering conditions, respectively.

2107 The compressibility factor at reference conditions, Z_S , has been calculated
 2108 according to the ISO 6976:2017 [94] equation:

$$Z_S = 1 - \left(\frac{p}{p_0} \right) \cdot \left[\sum_{j=1}^n x_j \cdot s_j(t, p) \right]^2 \quad (3.13)$$

2109 where: i) $p_0 = 101325$ kPa is the reference pressure and p_2 is the pressure at
 2110 metering conditions; ii) x_j is the mole fraction of the j -th component in the gas mixture;
 2111 iii) s_j is the summation factor of the j -th component in the gas mixture, whose values
 2112 are reported in ISO 6976:2017 [94] at the metering reference temperatures.

2113 The compressibility factor at metering conditions, Z , has been calculated in
 2114 accordance with the ISO 12213–2:2010 standard [98]. The standard employs the
 2115 AGA8-DC92 detailed characterization equation, developed by Starling e Savidge in
 2116 1992 [99]. This is an extended virial-type formulation, only capable to calculate the
 2117 properties in the gaseous phase. The input data required to use the AGA8-DC92
 2118 equation are the absolute pressure, absolute temperature and the detailed molar
 2119 composition of gaseous mixture, in which all constituents present in amounts
 2120 exceeding a mole fraction of 0.00005 should be represented. It may be written as:

$$Z = 1 + B\rho_m - \rho_r \sum_{n=13}^{18} C_n^* + \sum_{n=13}^{58} C_n^*(b_n - c_n k_n \rho_r^{k_n}) \rho_r^{k_n} \exp(-c_n \rho_r^{k_n}) \quad (3.14)$$

2121 where: i) B is the second virial coefficient; ii) ρ_m is the molar density; iii) ρ_r is the
 2122 reduced density; iv) b_n, c_n, k_n are constants; v) C_n^* are the coefficient which depend on
 2123 the temperature and NG mixture composition. The reduced density ρ_r is correlated to
 2124 the molar density ρ_m through the below relation:

$$\rho_r = K^3 \rho_m \quad (3.15)$$

2125 where: i) K is a mixture size coefficient; ii) ρ_m is the molar density calculated using
 2126 the following equation:

$$\rho_m = \frac{p}{ZRT} \quad (3.16)$$

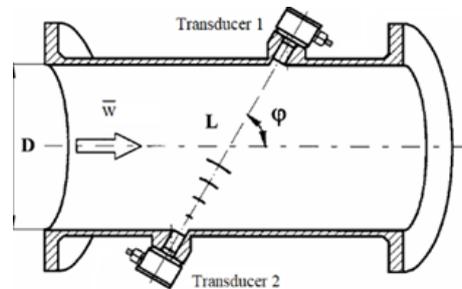
2127 where: i) p is the absolute pressure; ii) R is the universal gas constant; iii) T is the
 2128 absolute temperature.

2129 Observing the equation (12), it can be pointed out the NG volume at base condition
 2130 is highly influenced by the compressibility factor. Therefore, the calculation method
 2131 of compressibility factor is critical in NG industries especially for large flow metering
 2132 plants [100]. In order to assess its reliability, the calculation of the compressibility
 2133 factor has been carried out also using the ISO 12213-3, AGA NX 19 and AGA NX 19
 2134 Mod. standards, which are still used in NG industries.

2135 3.3.1.3 Ultrasonic static gas meter measuring principles

2136 The ultrasonic gas meters for domestic use determine the volume of the gas flowing
 2137 through the gas meter measuring the transit times (e.g., Time of flight, TOF) of the
 2138 ultrasonic waves generated by a pair of piezoelectric transducers operating as
 2139 transmitter/receiver (see Figure 3.1). The measurement of the transit times (e.g., t_1 and
 2140 t_2) and the knowledge of geometric characteristics of the measurement tube (e.g.,
 2141 diameter, D , propagation angle, φ and distance between the two transducers, L) allow
 2142 to calculate the average velocity (\bar{w}) of the gas flow and therefore, the volumetric flow
 2143 rate (V), employing the following equation:

$$\bar{w} = \frac{L}{2 \cos \varphi} \left(\frac{1}{t_1} - \frac{1}{t_2} \right) = \frac{L^2}{2 D} \cdot \frac{\Delta t}{t_1 t_2} = \frac{L^2}{2 D} \cdot \Delta f \quad (3.17)$$



2144

2145 *Figure 3.1 – Measurement principle of the ultrasonic gas meters.*

2146 Since the speed of sound in the gas flow (c_0) is unknown, depending on the
 2147 thermodynamic state (e.g., pressure and temperature) and composition of the fluid, the
 2148 measurement of \bar{w} is performed through the measurement of transit times of two
 2149 different ultrasonic signals. The transit time (t_1) of an ultrasonic signal traveling
 2150 cocurrent with the gas flow from Transducer 1 to Transducer 2 is measured. After the
 2151 measurement is completed, the transit time (t_2) of an ultrasonic signal traveling counter
 2152 current the gas flow from Transducer 2 to Transducer 1 is measured. Specifically, the
 2153 transit time of the signal cocurrent with the gas flow will be lower than that of the
 2154 signal counter current with the gas flow owing to the velocity of the gas within the
 2155 meter. This involves different frequencies of the two train waves.

2156 From the analysis of equation (3.17), it can be deduced that the average velocity
 2157 (\bar{w}), and consequently the volumetric flow rate (V), is not theoretically influenced by
 2158 the thermodynamic conditions and composition of the gas flow. However, if the speed
 2159 of sound is analysed it can be observed that it is significantly influenced by the H_2
 2160 content in the NG mixture.

2161 The effects of gas mixtures of variable compositions (mixtures of natural gas,
 2162 natural gas with hydrogen additives, and mixtures of synthetic biogas) on the accuracy

2163 of domestic gas meters (e.g., ultrasonic, thermal-mass and turbine) have been
 2164 investigated at laboratories of the Oil and Gas Institute – National Research Institute.
 2165 To this aim, a test bench capable of testing the metrological performance of gas meters
 2166 with natural gas mixtures with different H₂ content was designed and built, enabling
 2167 measurement in the range of 0.016 to 25 m³/h at low gas pressure (approx. 20 mbar).
 2168 A wet drum gas meter (for flow-rate range up to 0.4 m³/h) and a rotary gas meter (in
 2169 the range from 0.3 to 25 m³/h) were used as reference gas meters. The measurement
 2170 uncertainty on the test bench is within 0.3% in the range from 0.6 to 6 m³/h and within
 2171 0.45% below 0.6 m³/h. The research is underway and will be completed at the end of
 2172 2021. Therefore, a detailed description of the measurement stand, test methodology
 2173 and the obtained test results with discussion will be presented in a separate publication.
 2174 Only a general outline of the tests and example results for ultrasonic gas meters are
 2175 presented below. The research program included testing the errors of indications with
 2176 the use of gas mixtures presented in *Table 3.2*.

2177 *Table 3.2 – Characteristics of the gas mixtures used for the metrological tests.*

Sample	Gas mixture
Air	Atmospheric air
2E/H0	2E natural gas mixture, without hydrogen
2E/H5	2E natural gas mixture, with 5% hydrogen content (V/V)
2E/H10	2E natural gas mixture, with 10% hydrogen content (V/V)

2178 Domestic G4 thermal and ultrasonic gas meters have been tested at flow rates Q_{\min} ,
 2179 $3 \cdot Q_{\min}$, $0.1 \cdot Q_{\max}$, $0.2 \cdot Q_{\max}$, $0.4 \cdot Q_{\max}$, $0.7 \cdot Q_{\max}$ and Q_{\max} (three repetitions), whereas an
 2180 industrial turbine gas meter was tested at flow rates of 5, 8, 10, 16, 20, 25 m³/h (three
 2181 repetitions).

2182 **3.3.2 Results and discussion**

2183 **3.3.2.1 Impact of H₂ injection on the thermodynamic** 2184 **properties**

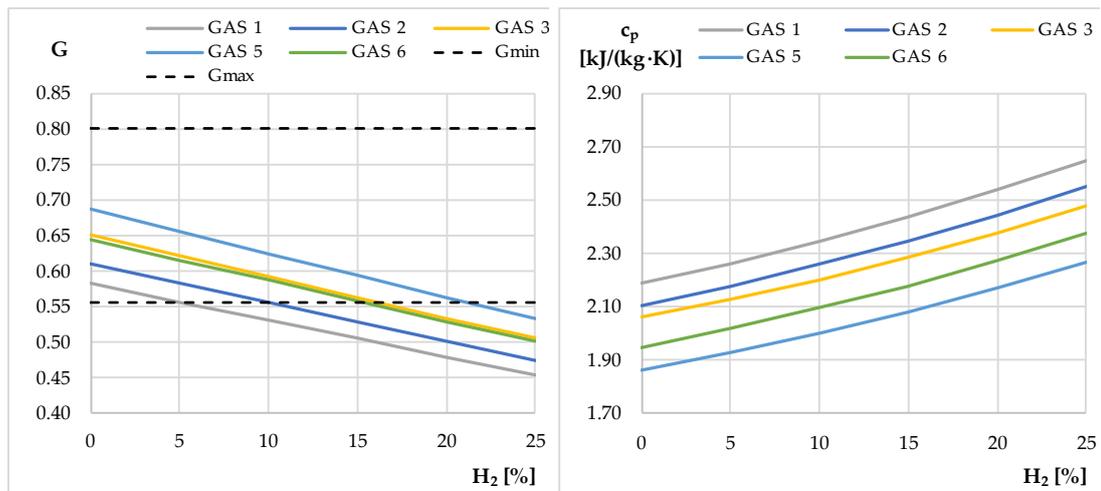
2185 In order to analyse the impacts of hydrogen injection in NG mixtures on the above
 2186 mentioned thermodynamic properties, the 5 NG mixtures (i.e., the ones without
 2187 presence of H₂) used as example calculations in Annex C of ISO 12213–2:2010
 2188 standard [98], as described in Table 3.3, have been investigated.

2189

Table 3.3 – Molar composition of the investigated gas mixtures [98].

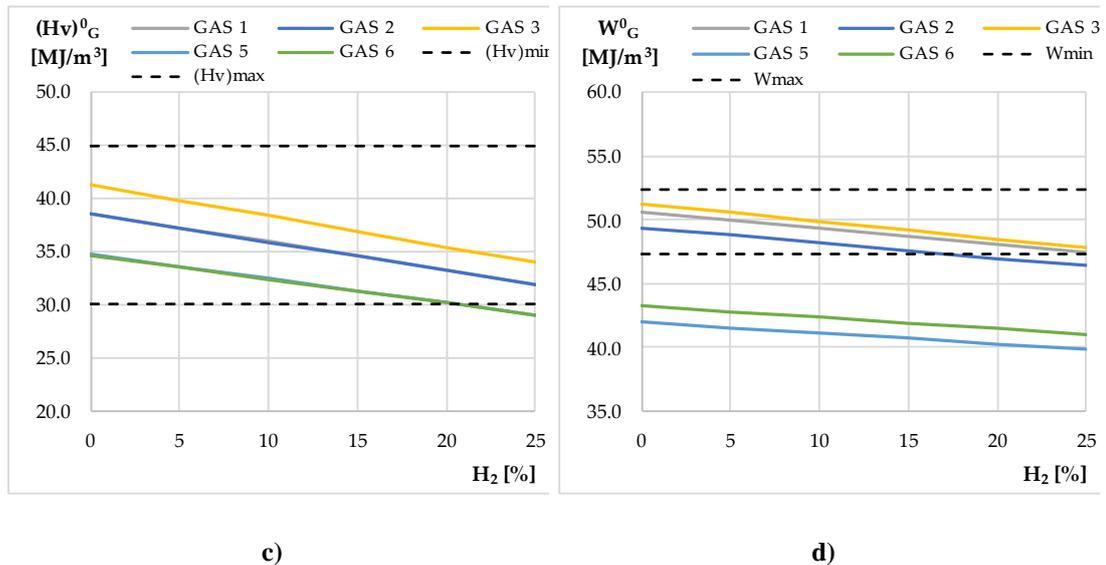
Component, j	GAS 1	GAS 2	GAS 3	GAS 5	GAS 6
Methane, CH ₄	96.50	90.70	85.90	81.20	82.60
Ethane, C ₂ H ₆	1.80	4.50	8.50	4.30	3.50
Propane, C ₃ H ₈	0.45	0.84	2.30	0.90	0.75
n-Butane, n-C ₄ H ₁₀	0.10	0.15	0.35	0.15	0.12
iso-Butane, iso-C ₄ H ₁₀	0.10	0.10	0.35	0.15	0.12
n-Pentane, n-C ₅ H ₁₂	0.03	0.04	0.05	0.00	0.04
iso-Pentane, iso-C ₅ H ₁₂	0.05	0.03	0.05	0.00	0.04
n-Hexane, C ₆ H ₁₄	0.07	0.04	0.00	0.00	0.02
Nitrogen, N ₂	0.30	3.10	1.00	5.70	11.70
Carbon dioxide, CO ₂	0.60	0.50	1.50	7.60	1.10

2190 The obtained results have been graphically represented in Figure 3.2. The dashed
 2191 lines, where present, show the allowable limit in accordance with the main current
 2192 applicable standards and national regulations (e.g., relative density and higher heating
 2193 value [15], Wobbe Index [16]).



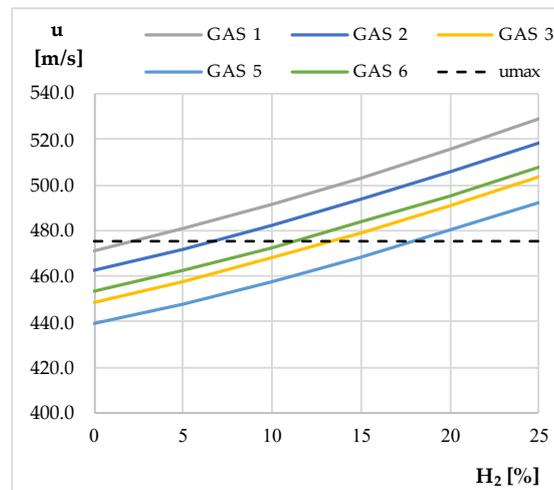
a)

b)



2194 *Figure 3.2 – Trend of the thermophysical properties as a function of the H₂ content in NG mixtures: a)*
 2195 *relative density, b) specific heat capacity, c) higher calorific value, d) Wobbe Index.*

2196 In Figure 3.3, the trend of the sound speed as a function of the H₂ content for the 5
 2197 NG mixture reported in Table 1 has been depicted.



2198
 2199 *Figure 3.3 – Trend of the sound speed as a function of the H₂ content in NG mixtures.*

2200 From Figure 3.3 it can be pointed out the speed of sound increases as H₂
 2201 concentration in the NG mixture increases and it becomes tendentially higher than the
 2202 allowable limit of 475 m·s⁻¹ (dashed line) of ISO 14236:2010 [101] for household
 2203 ultrasonic gas meters even at low H₂ contents.

2204 Hence, even though the measurement of the gas flow velocity is not theoretically
 2205 influenced by the speed of sound, since this is intrinsically compensated by the sing-

2206 around technique and mediated in the cross section (no effect in the transition of the
2207 flow from laminar to turbulent), the change of transit times measured in favor and
2208 against flow can lead to unpredictable deviations and a drift in the accuracy of the gas
2209 meters.

2210 From the analysis of the obtained results, it can be highlighted that:

- 2211 iv. the relative density decreases as the H₂ content increases in the NG mixture
2212 leading to values below the corresponding limit of ISO 12213–2 standard [98]
2213 at about 10% of H₂;
- 2214 v. the specific heat capacity at constant pressure significantly increases as the H₂
2215 content increases (e.g., up to + 21.8% at about 25% of H₂ for Gas 6);
- 2216 vi. the higher calorific value on volume basis notably decreases as the H₂ content
2217 increases (e.g., up to – 17.8% for Gas 3 at x_{H2}=25%) and for investigated Gas
2218 5 and 6, it reduces to below the corresponding limit indicated by the ISO
2219 12213–2 standard [98] at about 25% of H₂;
- 2220 vii. the Wobbe Index slightly decreases as the H₂ content increases (e.g., up to –
2221 6.6% at x_{H2}=25% for Gas 2);
- 2222 viii. the speed of sound increases as the H₂ content increases.

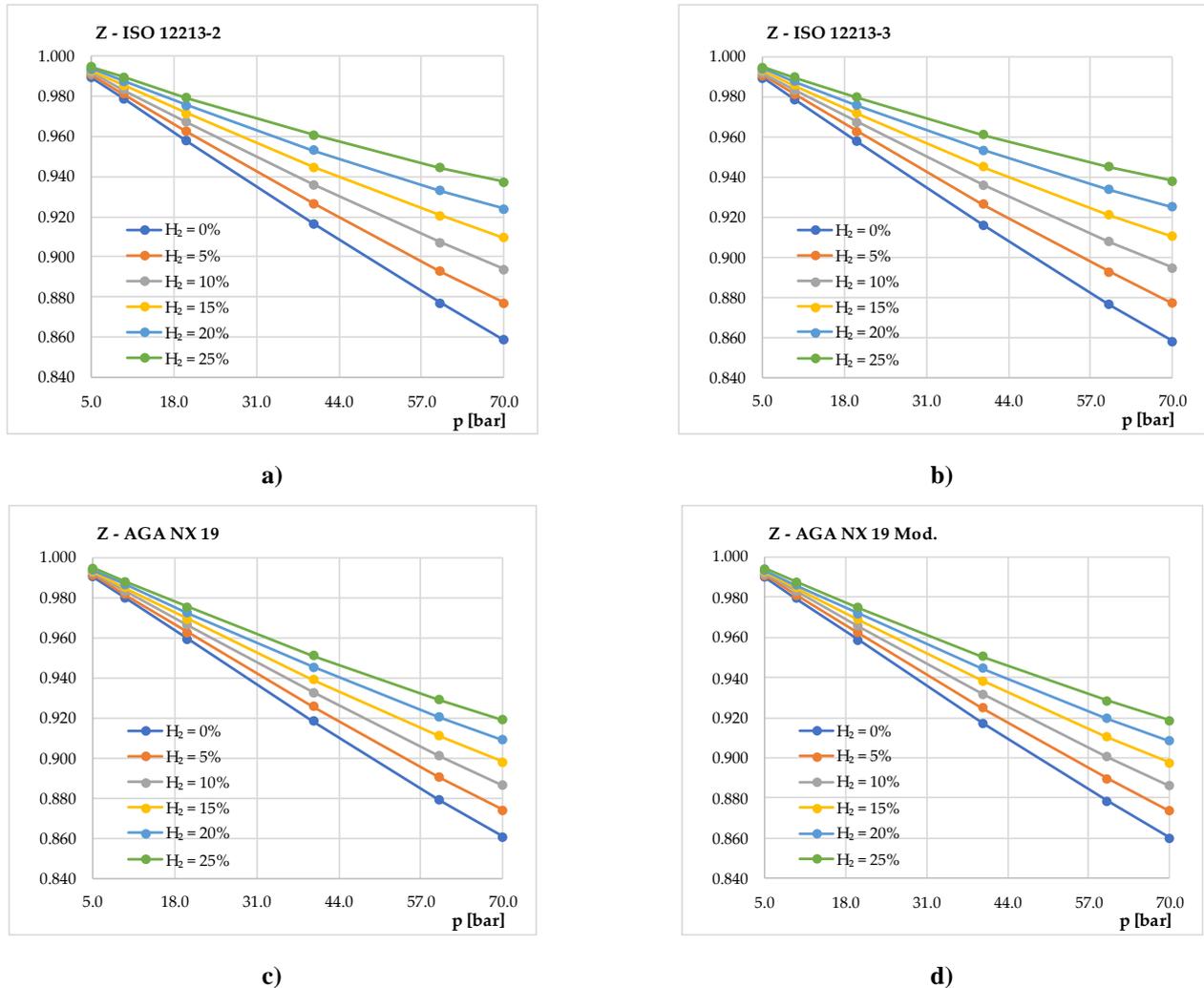
2223 3.3.2.2 Impact of H₂ injection on the volume conversion 2224 factor

2225 In order to evaluate the effects of injection of H₂ in NG mixtures on the accuracy
2226 of the volume conversion at standard conditions, a typical NG mixture distributed in
2227 Italy has been investigated (see Table 3.4).

2228 *Table 3.4 – Molar composition of the investigated gas mixture.*

Component, j	MIX 1
Methane, CH ₄	95.99
Ethane, C ₂ H ₆	2.292
Propane, C ₃ H ₈	0.639
n-Butane, n-C ₄ H ₁₀	0.097
iso-Butane, iso-C ₄ H ₁₀	0.102
n-Pentane, n-C ₅ H ₁₂	0.012
iso-Pentane, iso-C ₅ H ₁₂	0.018
n-Hexane, C ₆ H ₁₄	0.012
Nitrogen, N ₂	0.650
Carbon dioxide, CO ₂	0.188

2229 Figure 3.4 depicts the effects of the increase of H₂ content up to 25% in the
 2230 investigated NG mixture on the compressibility factor calculated employing the
 2231 aforementioned standards. Moreover, to investigate the influence of the metering
 2232 pressure, the compressibility factor calculation has been performed at different gas
 2233 pressures fixing a constant value of the gas temperature (i.e., T=15 °C).



2234 Figure 3.4 – Trend of the compressibility factor as a function of the H₂ content in NG mixtures: a) ISO
 2235 12213–2, b) ISO 12213–3, c) AGA NX 19 and d) AGA NX 19 Mod.

2236 From the obtained results, it can be observed that the compressibility factor
 2237 increases as the H₂ content increases. The compressibility factor is more influenced by
 2238 the H₂ concentration at high pressures than at low pressures: for instance, using the
 2239 ISO 12213–2 standard [98] an increasing in the range 0.10 – 0.44% at 5 bar and 2.1 –
 2240 9.1% at 70 bar has been found. Moreover, the results show that pressure more
 2241 significantly affects the compressibility factor at low H₂ concentrations. Finally, as

2242 expected, the compressibility factor decreases as pressure increases. As for example,
 2243 a decrease of Z has been found in the range 0.25 – 14% at $x_{H_2}=0\%$, whereas a lower
 2244 decrease in the range 0.05 – 0.2% at $x_{H_2}=25\%$ occurs.

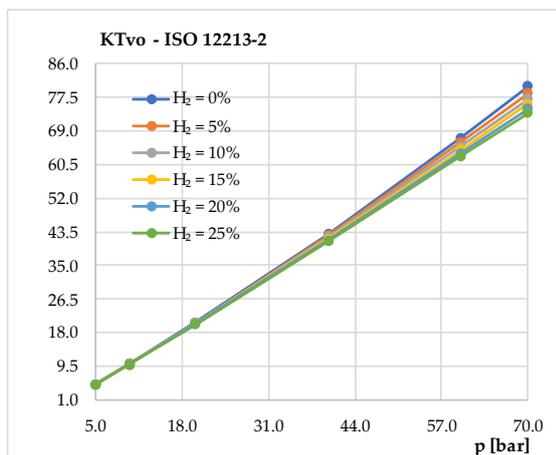
2245 In Table 3.5, the compressibility factor values and the corresponding deviations
 2246 between the ISO 12213–2 standard and the other standards on varying the H_2
 2247 concentration and pressure have been reported. From data in Table 5 it can be observed
 2248 that ISO 12213–2 and ISO 12213–3 methods lead to deviations within 0.09%, whereas
 2249 the use of AGA NX 19 and AGA NX 19 Mod. methods can lead to deviations up to
 2250 about 2% at high pressure and high H_2 content.

2251 *Table 3.5 – Compressibility factor values evaluated employing the ISO 12213-2, ISO 12213-3, AGA*
 2252 *NX 19 and AGA NX 19 Mod.*

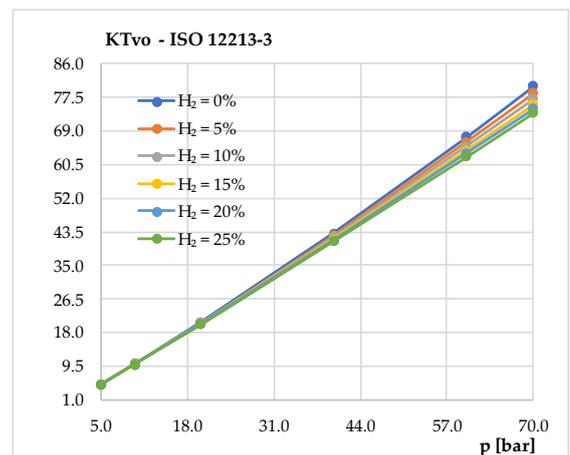
p [bar]	H ₂ [%]	Z _{ISO12213-2}	Z _{ISO12213-3}	Z _{AGANX19}	Z _{AGANX19Mod}	ΔZ _{ISO2-ISO3}	ΔZ _{ISO2-AGA19}	ΔZ _{ISO2-AGA19Mod}
5.00	0.00	0.98940	0.99026	0.99026	0.98997	0.09%	0.09%	0.06%
	0.50	0.99056	0.99118	0.99118	0.99090	0.06%	0.06%	0.03%
	10.00	0.99167	0.99206	0.99206	0.99179	0.04%	0.04%	0.01%
	15.00	0.99273	0.99290	0.99290	0.99262	0.02%	0.02%	0.01%
	20.00	0.99374	0.99369	0.99369	0.99342	0.00%	0.00%	0.03%
	25.00	0.99469	0.99445	0.99445	0.99417	0.02%	0.02%	0.05%
10.00	0.00	0.97881	0.97868	0.97974	0.97943	0.01%	0.09%	0.06%
	0.50	0.98117	0.98108	0.98154	0.98124	0.01%	0.04%	0.01%
	10.00	0.98341	0.98337	0.98326	0.98296	0.00%	0.02%	0.05%
	15.00	0.98555	0.98553	0.98489	0.98460	0.00%	0.07%	0.10%
	20.00	0.98757	0.98756	0.98643	0.98614	0.00%	0.12%	0.14%
	25.00	0.98949	0.98948	0.98789	0.98760	0.00%	0.16%	0.19%
20.00	0.00	0.95774	0.95748	0.95906	0.95871	0.03%	0.14%	0.10%
	0.50	0.96256	0.96243	0.96266	0.96232	0.01%	0.01%	0.03%
	10.00	0.96713	0.96709	0.96607	0.96574	0.00%	0.11%	0.14%
	15.00	0.97146	0.97147	0.96930	0.96897	0.00%	0.22%	0.26%
	20.00	0.97555	0.97558	0.97234	0.97202	0.00%	0.33%	0.36%
	25.00	0.97941	0.97944	0.97522	0.97491	0.00%	0.43%	0.46%
40.00	0.00	0.91635	0.91588	0.91803	0.91734	0.05%	0.18%	0.11%
	0.50	0.92642	0.92629	0.92545	0.92480	0.01%	0.11%	0.17%
	10.00	0.93586	0.93595	0.93242	0.93182	0.01%	0.37%	0.43%
	15.00	0.94471	0.94492	0.93898	0.93841	0.02%	0.61%	0.67%
	20.00	0.95300	0.95324	0.94515	0.94461	0.02%	0.82%	0.88%

	25.00	0.96076	0.96094	0.95094	0.95044	0.02%	1.02%	1.07%
60.00	0.00	0.87694	0.87628	0.87880	0.87825	0.07%	0.21%	0.15%
	0.50	0.89254	0.89255	0.89020	0.88968	0.00%	0.26%	0.32%
	10.00	0.90700	0.90741	0.90085	0.90036	0.04%	0.68%	0.73%
	15.00	0.92043	0.92101	0.91080	0.91033	0.06%	1.05%	1.10%
	20.00	0.93290	0.93349	0.92011	0.91966	0.06%	1.37%	1.42%
	25.00	0.94448	0.94493	0.92881	0.92839	0.05%	1.66%	1.70%
70.00	0.00	0.85843	0.85769	0.86052	0.85992	0.09%	0.24%	0.17%
	0.50	0.87682	0.87695	0.87387	0.87331	0.02%	0.34%	0.40%
	10.00	0.89378	0.89441	0.88631	0.88578	0.07%	0.84%	0.89%
	15.00	0.90945	0.91028	0.89791	0.89742	0.09%	1.27%	1.32%
	20.00	0.92394	0.92475	0.90874	0.90827	0.09%	1.65%	1.70%
	25.00	0.93736	0.93797	0.91883	0.91839	0.07%	1.98%	2.02%

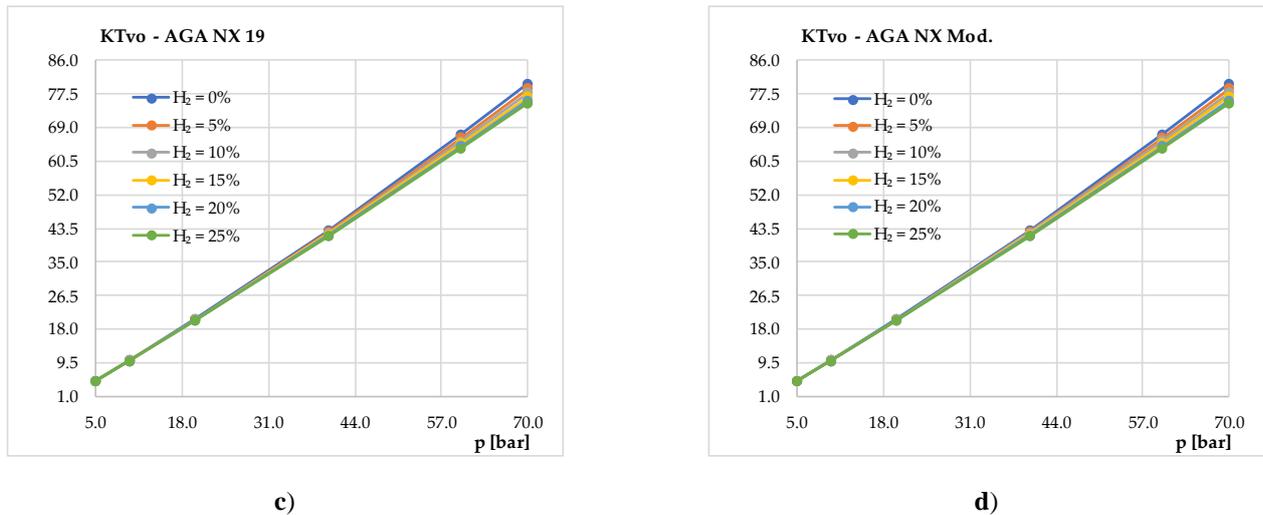
2253 Finally, the author investigated the effects of H₂ injection on the volumetric
 2254 correction factor, $KTvo$. From Figure 5, it can be observed that the volumetric
 2255 correction factor decreases as the H₂ content increases, regardless the calculation
 2256 algorithm used. Similarly to the compressibility factor, the H₂ content affects $KTvo$
 2257 more at high pressures than at low pressures. As for example, the $KTvo$ calculated
 2258 with the ISO 12213–2 standard decreases as H₂ increase in the range 0.10 – 0.43% at
 2259 5 bar, whereas it increases in the range 2.1 – 8.3% at 70 bar.



a)



b)



c)

d)

2260 Figure 3.5 – Trend of the volumetric correction factor as a function of the H₂ content in NG mixtures:
 2261 a) ISO 12213–2, b) ISO 12213–3, c) AGA NX 19 and d) AGA NX 19 Mod.

2262 3.3.2.3 Impact of H₂ injection on the ultrasonic domestic gas 2263 meters

2264 Domestic ultrasonic gas meters manufactured in 2021, with a measuring range of
 2265 0.04 to 6 m³/h, were subjected to metrological tests with the use of various gas
 2266 mixtures. Metrological tests were carried out according to par. 5.3.2 of EN
 2267 14236:2018, at selected flow rates (Q_{\min} , $3 \cdot Q_{\min}$, $0.1 \cdot Q_{\max}$, $0.2 \cdot Q_{\max}$, $0.4 \cdot Q_{\max}$, $0.7 \cdot Q_{\max}$,
 2268 Q_{\max}), each repeated three times. The gas mixtures listed in Table 2 have been used.
 2269 Experimental test results are presented in Table 3.6 and Figure 3.6.

2270 The average errors of indication of ultrasonic gas meters, the differences between
 2271 the errors of indication obtained for reference natural gas and individual gas mixtures,
 2272 and the metrological analysis of the obtained test results are presented below. For the
 2273 metrological assessment, average error values were used, while, when analysing the
 2274 maximum and minimum error values, the values obtained for each gas meter were
 2275 used individually. In order to define the metrological criterion necessary to assess the
 2276 impact of H₂ injection on the gas meter errors of indications (metrologically significant
 2277 or metrologically insignificant), the measurement uncertainty at single flow rates was
 2278 estimated. As the assessment covers the difference in the average errors of indications
 2279 for individual mixtures in relation to the errors of indications obtained for the reference
 2280 mixture of natural gas without hydrogen addition (e.g., between 2E/H0 and 2E/H5),
 2281 the measurement uncertainty is the total uncertainty in determining the errors in gas
 2282 meter readings for the gas reference 2E/H0 and the tested mixture with the addition of

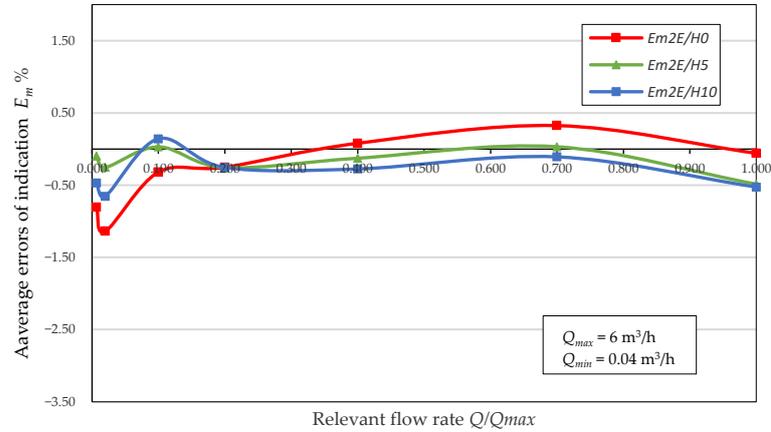
2283 hydrogen or biogas (e.g., 2E/H5, 2E/H10). The uncertainty of the average errors of
 2284 indications, $U(E_m)$, includes the type B uncertainty of the measuring equipment and
 2285 the type A uncertainty (i.e., the standard deviation of the results of the three tested gas
 2286 meters). The maximum allowable difference in errors $U(E_{mH-m2E})$ is obtained by the
 2287 square sum $U(E_m)$ of the two mixtures; thus, the metrological assessment means
 2288 “insignificant” if the difference in average errors is below the permitted errors
 2289 difference.

2290 In addition, the weighted mean errors (WME) was also evaluated according to par.
 2291 3.2.5 OIML R 137 1 & 2:2012 [102], according to which the WME for class 1.5 gas
 2292 meters should be within $\pm 0.6\%$. Table 3.6 shows the average errors of indications
 2293 (E_{mAIR} , $E_{m2E/H0}$, $E_{m2E/H5}$, $E_{m2E/H10}$) obtained for individual gas mixtures, average
 2294 differences ($\Delta E_{2E/H5}$, $\Delta E_{2E/H10}$) between the errors obtained for natural gas ($E_{m2E/H0}$)
 2295 and individual gas mixtures ($E_{m2E/H5}$, $E_{m2E/H10}$), and weighted mean errors WMEs
 2296 together with WMEs drift for G4 ultrasonic gas meters.

2297 *Table 3.6 – Average errors and average errors of indication drifts as well as WMEs and WMEs drifts*
 2298 *of G4 ultrasonic gas meters tested with different gas mixtures.*

	Flow Rate							
	Q_{\min}	$3 \cdot Q_{\min}$	$0.1 \cdot Q_{\max}$	$0.2 \cdot Q_{\max}$	$0.4 \cdot Q_{\max}$	$0.7 \cdot Q_{\max}$	Q_{\max}	
	Average error E_m [%]							WME [%]
E_{mAir}	-0.63	-0.99	-0.66	-0.44	-0.40	-0.11	-0.33	-0.30
$E_{m2E/H0}$	-0.81	-1.14	-0.32	-0.25	0.08	0.32	-0.06	0.07
$E_{m2E/H5}$	-0.10	-0.25	0.03	-0.26	-0.13	0.03	-0.49	-0.15
$E_{m2E/H10}$	-0.47	-0.65	0.14	-0.25	-0.28	-0.11	-0.53	-0.25
	Average error drift ΔE_m [%]							WME drift [%]
$\Delta E_{m2E/H5}$	0.71	0.88	0.35	-0.01	-0.21	-0.29	-0.43	-0.22
$\Delta E_{m2E/H10}$	0.34	0.48	0.46	0.00	-0.36	-0.43	-0.47	-0.31

2299 Figure 3.6 shows the average errors of indications of G4 ultrasonic gas meters
 2300 obtained for the gas mixtures 2E/H0, 2E/H5, 2E/H10.



2301

2302 *Figure 3.6 – Average errors of indications of G4 ultrasonic gas meters tested with different gas*
 2303 *mixtures as a function of relative flow rates Q/Q_{max} .*

2304 The error of indication of the investigated ultrasonic gas meters with 2E/H0 natural
 2305 gas mixture (i.e., without hydrogen) ranges -0.58% to 0.41% above $0.1 \cdot Q_{max}$ and
 2306 -0.65% to -0.19% below $0.1 \cdot Q_{max}$. On the other hand, error of indication with 2E/H5
 2307 mixture ranges -0.68% to 0.26% above $0.1 \cdot Q_{max}$, and 0.07% to 0.70% below $0.1 \cdot Q_{max}$.
 2308 The errors of indications of all gas meters with both the 2E/H0 and 2E/H5 mixtures,
 2309 were within the permissible limits (i.e., $\pm 2\%$ in the range from $0.1 \cdot Q_{max}$ to Q_{max} and
 2310 $\pm 3.5\%$ in the range below $0.1 \cdot Q_{max}$) of EN 14236:2018 [23] for ultrasonic gas meters
 2311 with temperature conversion. The maximum WME of the investigated gas meters was
 2312 found: (i) ranging 0.01% to 0.12% for 2E/H0, (ii) ranging -0.22% to 0.00% for
 2313 2E/H5, and (iii) ranging -0.38% to -0.10% for 2E/H10 mixture. Table 3.7 shows the
 2314 average error of indication drifts ($\Delta E_{2E-2E/H5}$) between 2E/H5 and 2E/H0 mixtures,
 2315 together with the related metrological assessment.

2316 *Table 3.7 – Metrological assessment of average error of indication drift of ultrasonic G4 gas meters*
 2317 *with 2E/H5 and 2E/H0.*

Flow Rate	Average Errors Drift [%]	Uncertainty in Average Errors [%]		Permitted Errors Difference [%]	Metrological Assessment
	$\Delta E_{m(2E-2E/H5)}$	$U(E_m)$		$U(\Delta E_{2E-2E/H5})$	
		2E/H5	2E		
Q_{max}	-0.43	0.17	0.16	± 0.46	Insignificant
$0.7 \cdot Q_{max}$	-0.29	0.15	0.15	± 0.42	Insignificant
$0.4 \cdot Q_{max}$	-0.21	0.17	0.22	± 0.55	Insignificant
$0.2 \cdot Q_{max}$	-0.01	0.24	0.27	± 0.72	Insignificant
$0.1 \cdot Q_{max}$	0.35	0.21	0.14	± 0.50	Insignificant

$3 \cdot Q_{\min}$	0.88	0.20	0.18	± 0.53	Significant
Q_{\min}	0.71	0.28	0.24	± 0.72	Insignificant

2318 In this case, the metrological assessment was found significant only at 3 Q_{\min} flow
 2319 rate, meaning that there is a slight influence on the gas meter accuracy when a 5%
 2320 H2NG mixture is used. On the other hand, when analysing the errors of indications, it
 2321 should be noted that hydrogen injection up to 10% vol does not significantly affect the
 2322 accuracy of gas meters.

2323 3.3.3 Quality measurements of NH2NG mixtures

2324 At present, typical process chromatographs dedicated to natural gas analyses do not
 2325 allow the in-line determination of the hydrogen content. This is due to the fact that
 2326 currently used process chromatographs (PGCs) use helium as a carrier gas, so they are
 2327 not capable to detect and determine hydrogen due to the similar thermal conductivity
 2328 of both these gases (151 and 180 $\text{Wm}^{-1}\text{K}^{-1}$ for helium and hydrogen, respectively).
 2329 Therefore, solutions that can be used to analyse natural gas-hydrogen mixtures should
 2330 be sought in devices dedicated to other types of gaseous fuels or constructed on the
 2331 basis of customer guidelines. Aiming at analysing the composition of natural gas-
 2332 hydrogen mixtures (hydrogen content up to 37% mol/mol), inter alia, a four-channel
 2333 chromatograph equipped with four TCD detectors, can be used [103–105]. This device
 2334 uses two carrier gases (argon and helium) and the declared duration of the analytical
 2335 cycle for a configuration that allows determination of all required components is 3 to
 2336 5 min. A frequent calibration of this chromatograph is recommended ranging from
 2337 daily to every 3 months.

2338 On the other hand, in the refinery sector, a first solution relies on three-channel
 2339 analysers that enable the determination of components in the gas such as hydrogen,
 2340 helium, nitrogen, oxygen, carbon monoxide (II), carbon monoxide (IV) and
 2341 hydrocarbons from the C1–C5 range, including unsaturated hydrocarbons broken
 2342 down into isomers, and the total content of C6+ hydrocarbons [106,107]. These
 2343 analysers are equipped with thermal conductivity detectors (TCD) and a flame
 2344 ionization detector (FID), which is the third measurement channel enabling the
 2345 determination of the content of hydrogen and other basic components of natural gas-
 2346 hydrogen mixtures up to 100%. The typical time of analysis with their use is 15 min;
 2347 however, it requires additional technical gases supplying the FID detector to the
 2348 chromatograph, i.e., hydrogen and synthetic air as well as helium and nitrogen, which
 2349 are reference gases in TCD detectors, which increases the cost of operating the device.

2350 Additionally, it is possible to reduce the analysis time to 7.5 min, if hydrogen is used
2351 as the carrier gas, then the hydrogen content in the sample is determined by using a
2352 TCD detector with nitrogen as the reference gas. However, due to the explosive
2353 properties of hydrogen, its use as a carrier gas in gas chromatography is often rejected
2354 by laboratories. Single (double) channel chromatograph with a single (double) thermal
2355 conductivity detector (TCD) are also available [108,109], enabling in the scope of such
2356 components as pseudo-component C6+, hydrogen, helium, oxygen, carbon monoxide
2357 (IV), methane, ethane, propane, butanes, pentanes (with the double-channel
2358 configuration even nitrogen can be detected). Optionally the chromatograph can be
2359 equipped with an additional third channel with a flame ionization detector (FID),
2360 allowing individual determination of C6–C16 hydrocarbons. A further chromatograph
2361 dedicated to natural gas analyses enables the determination of the content of C1–C9
2362 hydrocarbons, oxygen, nitrogen, carbon monoxide (IV), helium and hydrogen sulphide
2363 in the gas [110]. However, no detailed information on the range of the analysed
2364 concentrations and the duration of the analysis is available. An important feature of
2365 gas chromatographs dedicated to refinery gas analysis is that they allow the
2366 determination of hydrogen content in the entire concentration range by using two TCD
2367 detectors and an FID detector [111]. Depending on the model, the analysis time ranges
2368 from less than 9 min to 17 min.

2369 Chromatographs for both H2NG mixtures and for refinery gas analyses are also
2370 available in the market [112,113]. Although chromatographs dedicated to refinery gas
2371 analyses are characterized by a sufficiently wide analytical range for hydrogen
2372 determination; however, due to the possibility of methane analysis only up to 80%
2373 mol/mol, they cannot be used in the case of analyses of certain mixtures of natural gas
2374 with hydrogen and natural gas from the H group. On the other hand, chromatographs
2375 dedicated to natural gas analyses allow for analysis in the scope of determination of at
2376 least: He, O₂, N₂, CH₄, CO, CO₂, C₂–C₅ hydrocarbons, H₂S and of course H₂ up to
2377 10%. The analysis time depends on the model and ranges from 10 to 40 min. These
2378 chromatographs are equipped with two TCD detectors or two TCD detectors and a
2379 FID detector.

2380

2381 **CHAPTER 4. ANALYSIS OF RELIABILITY**
2382 **OF THERMODYNAMIC EQUILIBRIUM**
2383 **GASIFICATION MODELS**

2384 **4.1 Biomass gasification process overview**

2385 Biomass gasification as a conversion technology is becoming increasingly
2386 important on account of renewability and environmental benefits, as well as
2387 sociopolitical benefits [114]. Biomass gasification is expected to play a crucial role in
2388 supplying the energy demand in many countries and being one of the economically
2389 attractive technology for the production of clean energy.

2390 Gasification is a thermochemical conversion process of carbonaceous combustible
2391 through the controlled delivery of an oxidizing agent (e.g., air, vapor, oxygen, or
2392 carbon dioxide). This process occurs under sub-stoichiometric conditions of the
2393 oxidant, in a high temperature and low oxygen content environment. The oxygen
2394 amount supplied is limited to avoid complete combustion so that the energy is not
2395 converted all into heat, but also into chemical energy in form of a gaseous fuel known
2396 as syngas or producer gas. The latter is mainly composed of CO, H₂, CO₂, N₂, methane
2397 CH₄, tar and ash [115]. The physical and chemical properties of the syngas, such as its
2398 chemical composition, heating value and temperature, vary as a function of the
2399 biomass properties, the gasification agent (e.g., type, flow rate, temperature, etc.), the
2400 operating conditions (e.g., temperature and pressure), the heating rate and the type of
2401 gasifier used [116]. A conventional gasification process follows several successive
2402 steps: i) drying, ii) pyrolysis, iii) partial combustion of gases, vapours and char, and
2403 iv) reduction of the combustion products. The drying zone constitutes the region where
2404 the raw biomass is dried vaporizing its moisture content that contributes to lower
2405 quality syngas. In that zone, the temperature typically reaches around 100 °C and no
2406 thermal decomposition has still started at this stage [115,117,118]. In the pyrolysis
2407 zone, the thermal composition of biomass feedstock takes place in absence of oxygen
2408 or air within a temperature range of 125 °C to 500 °C. At this stage, the feedstock is

2409 decomposed into charcoal, liquid tars and a mixture of gases containing carbon
2410 monoxide, hydrogen, carbon dioxide and hydrocarbons. The oxidation zone concerns
2411 the region where the volatile materials generate mainly CO, CO₂ and H₂O through a
2412 series of exothermic reactions, under temperatures that may vary from 1000 °C to
2413 1500 °C. At the last stage, the reduction of the tar particles occurs at temperatures
2414 around 1000 °C, while endothermic reactions take place generating combustible
2415 products such as CO, H₂, and CH₄. Such stage is crucial for syngas quality since the
2416 presence of tar particles reduce the biomass conversion efficiency, in addition to being
2417 a severe potential of clogging fuel lines, filters and engine.

2418 The major advantage of the gasification process is that it lets convert waste biomass
2419 and other similar biodegradable wastes, which would otherwise have been discarded
2420 and unused, into valuable fuel. The synthesis gas produced through biomass
2421 gasification is suitable for multiple applications as it can be i) burned at higher
2422 temperatures, (ii) employed in fuel cells, (iii) employed to produce methanol and
2423 hydrogen and (iv) converted via the Fischer-Tropsch process into a range of synthesis
2424 liquid fuels suitable for use gasoline engines or diesel engines [119]. The by-products
2425 (e.g., biochar, tar) derived from the gasification process can also be used in multiple
2426 ways. Such as biochar can be employed as fertilizer in agriculture, filter absorber in
2427 industry or energy carrier [120] and recently, some researchers have investigated the
2428 feasibility of using biochar as carbon sequestering additive in cement mortar [121].
2429 Furthermore, the gasification process allows to use as feedstocks a variety of biomass
2430 resources, including wood material, pulp and paper industry residues, agricultural
2431 residues, organic municipal material, sewage, manure and food processing by-
2432 products. A great number of agricultural cultivation residues such as straws, nutshells,
2433 fruit shells, fruit seeds, plant stalks and stover, green leaves and molasses are globally
2434 produced each year [122]. For example, 89.5 million tonnes of rapeseed have been
2435 cultivated worldwide in 2017 [123]. Rapeseed plantations entail the production of a
2436 large quantity of straw that needs to be disposed of and since the disposal by open field
2437 burning causes air pollution, the use of a gasification technology can have a scarce
2438 environmental impact and smaller consequences on the environment than landfill
2439 operations or combustion of the waste. For these reasons, in the last decades,
2440 gasification has been reputed a valuable solution also for disposal of refuse-derived
2441 fuel from municipal solid waste, especially in the developing countries [124].

2442 The thermochemical conversion process of gasification is a mature, reliable and
2443 flexible technology, well known and it has been described extensively in literature
2444 [115,125–127]. Even though biomass gasification is a mature technology, some

2445 technological barriers limit its large-scale commercial development. In particular, it
2446 has been observed that the main constraints in the implementation of biomass
2447 gasification are represented by: i) the variability of biomass feedstock physical
2448 properties, which requires a high flexibility of the gasifier, ii) the high moisture content
2449 of biomass since it decreases the energetic efficiency of the process, iii) the production
2450 of ash during the process as it can cause air pollution, iv) the presence of tar in the end
2451 products due to its tendency of causing fouling and plugging of the plant pipes and v)
2452 the handling and management of biomass feedstock [118].

2453 In order to overcome some of the aforementioned technological barriers, a deep
2454 understanding of the different phenomena and operational parameters involved in the
2455 gasification process is required. To this aim, a mathematical modelling is a useful tool
2456 to understand the physical and chemical mechanisms that occur inside the gasifier. A
2457 numerical model allows to evaluate process performance on varying the biomass
2458 properties and operating parameters, providing a set of optimum operating conditions
2459 and designating the risks and limits of the system. Besides, a numerical model helps
2460 to reduce considerably the costs of research and development of new and innovative
2461 devices [128].

2462 In following paragraphs, a comprehensive analysis of the thermodynamic
2463 equilibrium gasification models is presented. Especially, the reliability of the
2464 equilibrium models for the simulation of the gasification process as a function of the
2465 biomass properties and the operating conditions is assessed. This study aims to better
2466 understand in which conditions of practical interest such models can be applied with
2467 acceptable reliability, even neglecting tar and char formation to minimise model
2468 complexity. Indeed, in case of the analysis of complex energy systems as
2469 polygenerative plants a simple and flexible model can represent an easily usable tool
2470 for the evaluation of the optimal process parameters. To this aim, a thermodynamic
2471 equilibrium model implemented in a non-commercial script is proposed as case study.
2472 The developed model has been applied to a large number of experimental case studies
2473 available in the scientific literature, assessing thermodynamic equilibrium models
2474 performance as for gasification different biomasses and in different operating
2475 conditions.

2476 **4.2 Biomass gasification modelling**

2477 Mathematical models for the biomass gasification process can be categorized into:
2478 i) thermodynamic equilibrium models, ii) kinetic rate models, iii) computational fluid
2479 dynamics models (CFD) and iv) artificial neural networks (ANNs) models.

2480 Thermodynamic equilibrium models are based on the assumption that the system
2481 of chemical reactions involved in the process reaches the state of thermochemical
2482 equilibrium inside the gasifier reactor [126,129]. Equilibrium models are able to
2483 predict the producer gas composition, the maximum yield and, the optimal conditions
2484 of energy efficiency and syngas heating value for the operation of each specific reactor
2485 according to biomass feedstock properties [115,125–127]. This approach does not
2486 require details of system geometry or estimation of the necessary time to reach the
2487 equilibrium [130].

2488 Kinetic models are based on the mathematical description of the reaction kinetics
2489 of the main reactions and the hydrodynamic phenomena among phases that occur
2490 inside the gasification reactor [126,127]. Such models are able to predict the trend and
2491 composition of the product in different positions along a reactor, as well as the overall
2492 gasifier performance as a function of operating conditions and configuration of the
2493 gasifier [127,131]. Therefore, they constitute a useful approach to study gasification,
2494 especially for the design and optimization of gasifiers [125,126]. Application of kinetic
2495 models involves the knowledge of reaction rate, residence time, reactor
2496 hydrodynamics (e.g., superficial velocity, diffusion rate) and gasifier geometry. The
2497 key issues with kinetic modelling are the associated high cost for computation and the
2498 use of parameters that make them difficult to apply to different plants [126,132].
2499 Indeed, kinetics modelling is quite complicated and requires several assumptions that
2500 have yet to be validated theoretically and experimentally measured.

2501 Computational fluid dynamics (CFD) modelling is one of the most advanced tools
2502 for the analysis of the gasification process. CFD models are based on the solution of a
2503 set of mathematical equations based on conservation of mass, momentum, energy, and
2504 species over a discrete region of the gasifier reactor [115,126,127]. CFD simulations
2505 provide the temperature profiles of the gasifier, solid and gas phases, as well as a
2506 complete chemical and fluid mechanical analysis of the reactor bed. CFD analysis can
2507 be used for the optimization of the gasifier design and to study the characteristics of
2508 the gasification process such as operating parameters, kinetics and physical properties
2509 of the feedstock. However, comprehensive CFD simulations for biomass gasification
2510 are scarce, mainly due to the lack of large computational resources and the anisotropic
2511 nature of biomass [125].

2512 Artificial neural networks (ANNs) models are a relatively new approach to model
2513 the gasification process. ANN technique concerns the ability of the system to learn by
2514 itself from previous experiences, finding patterns in the responses and improving the
2515 subsequent outputs, imitating some human features. ANN models are able to predict
2516 gas yield and syngas composition with sufficient accuracy [117,127]. Such models do
2517 not provide an analytical solution but only numerical results [115,127]. They do not
2518 require the mathematical description of the phenomena occurring in the process, the
2519 only input required is the input data sets. For this reason, they constitute a useful tool
2520 for simulating and scaling the complex biomass gasification process. Nevertheless, the
2521 application of ANN for biomass gasification is rather rare [117].

2522 The choice of the most suitable modelling approach to investigate the biomass
2523 gasification process is determined by the scope of analysis and the available
2524 experimental data. Complex models (e.g., kinetic, CDF and ANNs models) provide
2525 detailed information on mechanisms. However, the computational cost and the number
2526 of input data required are usually high. Although equilibrium models are valid under
2527 chemical equilibrium conditions, they are valuable because they can predict the
2528 thermodynamic limits of a gasification system, leading to the design, evaluation and
2529 improvement of the process [130,131]. Furthermore, the equilibrium models can
2530 constitute a useful design support in evaluating the limiting possible behaviour of a
2531 complex system that is difficult or unsafe to reproduce experimentally or in
2532 commercial operation [131].

2533 **4.2.1 Literature review on thermodynamic equilibrium** 2534 **modelling**

2535 The thermodynamic equilibrium approach is based upon the assumption that the
2536 components react in a fully mixed condition for an infinite period of time. This
2537 assumption can be considered realistic when the residence time of the reactants in the
2538 gasification zone is long compared to the half-life of all the reactants; gasification
2539 temperature can be assumed as constant, and chemical mixing is almost perfect [133].
2540 These conditions are met in some specific types of reactors, mainly in downdraft fixed-
2541 bed reactors, and at high temperatures (e.g. > 1500 K) [134]. However, the equilibrium
2542 state may not be reached in some gasifiers and under specific operative conditions, in
2543 particular for gasifiers in which the operating temperatures are relatively low.
2544 Nonetheless, thermodynamic equilibrium models have been extensively employed to
2545 investigate the biomass gasification process [135].

2546 Equilibrium models can be further categorized as non-stoichiometric or
2547 stoichiometric equilibrium approaches. Stoichiometric models are adapted from the
2548 equilibrium constants of a set of reactions [132], while the nonstoichiometric models
2549 are developed on the Gibbs free energy minimization. This formulation requires only
2550 the elemental composition of biomass as input, therefore it is particularly suitable for
2551 cases in which the system of chemical reactions that take place in the gasifier is not
2552 fully known [115]; while for the application of the stoichiometric equilibrium models,
2553 the knowledge of the main chemical reactions and species involved in the process is
2554 required.

2555 Zainal et al. [136] proposed a stoichiometric equilibrium model for a downdraft
2556 gasifier investigating the effects of initial moisture content in the biomass feedstock
2557 and the temperature in the gasification zone on the producer gas composition and the
2558 calorific value.

2559 Li et al. [137] developed a thermodynamic equilibrium model based upon the
2560 minimization of Gibbs free energy to predict the performance of a circulating fluidized
2561 bed gasifier. The authors proposed a phenomenological model modifying a previous
2562 model through the introduction of an availability function that uses empirical data (e.g.
2563 unconverted carbon and methane) to consider the non-equilibrium factors.

2564 Sharma [138] presented a stoichiometric equilibrium model of the global reduction
2565 reactions for char – gas and gas – gas reactions that occurs in the reduction zone of a
2566 downdraft gasifier. In order to predict accurately the reaction temperature, the model
2567 considers by energy equations the heat exchanges that take place in the reduction zone.
2568 The results have showed a reasonable agreement with data collected from various
2569 sources for downdraft biomass gasifiers.

2570 Jarungthammachote et al. [139] employed a thermodynamic equilibrium model
2571 based upon the equilibrium constants to predict the producer gas composition. In order
2572 to improve the accuracy of simulation results, the authors modified the model
2573 multiplying the equilibrium constant with appropriate coefficients calculated on the
2574 basis of experimental data.

2575 Baratieri et al. [140] presented an equilibrium model for the biomass gasification
2576 process modified according to experimental data. The model has been calibrated using
2577 the concentration of CH_4 and C_2H_4 measured experimentally and the amount of char
2578 residue. The results of the modified model predicted accurately the syngas composition
2579 for the gasification process in the analysed fluidized gasifier.

2580 Barba et al. [141] developed the Gibbs Free Energy Gradient Method Model
2581 (GMM). The gasification process has been modelled in two steps: in the first step, the

2582 raw material is decomposed to produce a carbonaceous residue and a primary gas. In
2583 the second stage, the gas previously produced changes its composition due to the
2584 occurrence of water shift and steam reforming reactions.

2585 In [142], the GMM thermodynamic model was employed to simulate an
2586 experimental campaign carried out at the ENEA Research Centre of Trisaia (Italy),
2587 which showed good accuracy to reproduce the experimental results of steam
2588 gasification of Refuse-derived fuel.

2589 An advanced two-stage fluidized bed gasification was modelled by Materazzi et al.
2590 [143] using a non-stoichiometric approach. The model has a systematic structure
2591 consisting of principles of atomic conservation and equilibrium calculation routines.
2592 The model has been validated employing the experimental data of a pilot plant.

2593 Sreejith et al. [144] presented an equilibrium model for steam gasification of
2594 biomass based upon the minimization of Gibbs free energy to identify the optimum
2595 values for the reactor varying temperature and pressure. The gas phases have been
2596 modelled as real gases in accordance with the Redlich–Kwong equation of state.
2597 Kangas et al. [145] developed a non-stoichiometric equilibrium model with the
2598 constrained Gibbs energy method. The assumed chemical system consists of 14
2599 constituents in the gaseous phase, liquid water phase and two solid phases for char and
2600 ash. Six methods for modelling global or local equilibrium were implemented by
2601 varying the constrained species.

2602 Mendiburu et al. [146] developed four thermodynamic equilibrium models for the
2603 downdraft gasifiers, called M1, M2, M3 e M4. The first model M1 is a general
2604 stoichiometric equilibrium model. Models M2, M3 and M4 have been obtained
2605 modifying the model M1. In particular, in the model M2 two multiplicative
2606 coefficients calculated using experimental data have been introduced in the
2607 equilibrium equations. Model M3 implements the correlations for the CO/CO₂ and
2608 CO/H₂ according to [147]. Model M4 implements the substitution of the formula for
2609 the calculation of equilibrium constants with the Arrhenius type relations.

2610 To improve the accuracy of the prediction of pyrolysis products, Biagini et al. [148]
2611 developed a multizonal model adapted from the non-stoichiometric equilibrium
2612 approach. The proposed model simulates the bypass of the oxidation zone of some
2613 products of pyrolysis phase inside the gasifier introducing a repartition coefficient of
2614 the volatile matter.

2615 Gagliano et al. [116] presented a stoichiometric model to simulate the gasification
2616 process through a downdraft gasifier filled with residual biomasses. The proposed
2617 model can predict the percentage of tar and char produced, as well as the chemical

2618 species in the producer gas. A good agreement between experimental data and
 2619 simulation data has been found.

2620 Gambarotta et al. [149] developed a thermodynamic equilibrium model for the
 2621 biomass gasification process in downdraft gasifiers. The model is able to predict not
 2622 only the concentration of the main component of syngas but also the content of minor
 2623 products, especially the pollutant substances containing nitrogen and sulphur (e.g.,
 2624 ammonia, hydrogen sulphide).

2625 ***4.3 Case study #5: Reliability of thermodynamic*** 2626 ***equilibrium gasification models for selected biomass*** 2627 ***types and compositions***

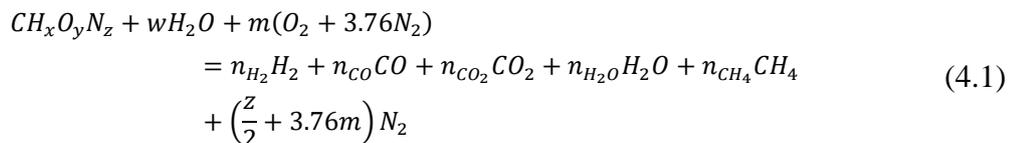
2628 **4.3.1 Material and methods**

2629 In the present chapter, the author presents an analytical model able to predict
 2630 gasification average temperature of the biomass bed, composition and flow rate of the
 2631 synthesis gas produced as well as its energy content. The proposed model describes
 2632 the process of biomass gasification using a concentrated parameter approach
 2633 [150,151].

2634 The hypotheses at the core of the model are: i) steady state; ii) uniform and constant
 2635 properties of biomass inside the gasifier; iii) local thermodynamic equilibrium; iv)
 2636 complete chemical reactions (i.e., the reactions taking place completely
 2637 stoichiometrically); v) uniform and constant temperature of biomass bed. Input data
 2638 for calculation are: i) the mass flow rate of biomass feedstock; ii) temperature and
 2639 volumetric flow rate of combustion air; iii) biomass composition.

2640 For modelling purposes, the chemical formula of biomass is defined as $CH_xO_yN_z$,
 2641 where x, y and z are the number of hydrogen, oxygen and nitrogen atoms per number
 2642 of carbon atoms in the feedstocks, respectively [152]. The latter is calculable knowing
 2643 the elemental composition of biomass based on ultimate and proximate analysis.

2644 Assuming thermodynamic equilibrium, the biomass gasification process is
 2645 modelled according to the following global reaction:



2646 where w is the amount of moisture per mole of carbon and, $m=O/C$ is the amount of
 2647 oxygen per mole of carbon. The number of moles of the different species, n_i , on the
 2648 right-hand side represents the unknown quantities of the problem. The moles produced
 2649 of H_2 , CO , CO_2 , H_2O and CH_4 were determined by solving simultaneous five
 2650 equations. The first three equations were related to mass balance on the involved
 2651 chemical elements, as shown in Eqs. (4.2) – (4.4) [152–154].

2652 Carbon mass balance:

$$n_{CO}CO + n_{CO_2}CO_2 + n_{CH_4}CH_4 - 1 = 0 \quad (4.2)$$

2653 Hydrogen mass balance:

$$2n_{H_2} + 2n_{H_2O} + 4n_{CH_4} - x - 2w = 0 \quad (4.3)$$

2654 Oxygen mass balance:

$$n_{CO} + 2n_{CO_2} + n_{H_2O} - w - 2m = 0 \quad (4.4)$$

2655 The two additional equations were obtained under the assumption of chemical
 2656 equilibrium, minimizing energy Gibbs free energy relative to the methanation reaction:



2657 and to the water – gas shift reaction:



2658 The equilibrium constants for the reactions (4.5) and (4.6) are given by the
 2659 following equation [151,152,155]:

$$\ln(K) = -\frac{\Delta G_T^0}{\bar{R}T} = -\frac{\sum_i v_i \overline{(\Delta g)_{f,T,i}^0}}{\bar{R}T} \quad (4.7)$$

2660 where $\bar{R} = 8.314 \left(\frac{kJ}{kmol \cdot K} \right)$ is the universal gas constant, ΔG_T^0 is the standard Gibbs
 2661 function of reaction, T is the temperature and $\overline{(\Delta g)_{f,T,i}^0}$ is the standard Gibbs function
 2662 of formation of the gaseous species in the producer gas at temperature the T . The latter
 2663 can be evaluated employing the empirical equation below:

$$\overline{(\Delta g)_{f,T}^0} = \bar{h}_f^0 - a'T \ln(T) - b'T^2 - \frac{c'}{2}T^3 - \frac{d'}{3}T^4 + \frac{e'}{2T} + f' + g'T \quad (4.8)$$

2664 The values of coefficients a' , b' , c' , d' , e' , f' and g' are reported in [152] for the i -
 2665 th component involved (i.e., CO , CO_2 , CH_4 , H_2O).

2666 In this model, thermodynamic equilibrium was assumed for all chemical reactions
 2667 in the gasification zone. Moreover, all gases were assumed to be ideal and all reactions

2668 form at pressure 1 atm. Therefore, the equilibrium constants for methanation reaction
 2669 and water gas shift reaction are given by ensuing relations [139,156,157]:

2670 Methanation reaction:

$$K_1 = \prod_i x_i^{v_i} \left(\frac{p}{p_0} \right)^{\sum_i v_i} = \frac{n_{CO_2} \cdot n_{total}}{n_{H_2}^2} \quad (4.9)$$

2671 Water gas shift reaction:

$$K_2 = \prod_i x_i^{v_i} \left(\frac{p}{p_0} \right)^{\sum_i v_i} = \frac{n_{CO_2} \cdot n_{H_2}}{n_{CO} \cdot n_{H_2O}} \quad (4.10)$$

2672 where x_i is the mole fraction of species i in the ideal gas mixture, v is the stoichiometric
 2673 number (positive value for products and negative value for reactants), p_0 is the standard
 2674 pressure and n_{total} is the total mole of producer gas. Using the equations (4.9) – (4.10),
 2675 it is possible to calculate the concentration of products and reactants. Mass balances
 2676 and chemical equilibrium relations equations are solved together with the energy
 2677 conservation equation for the gasification process, allowing the gasification
 2678 temperature prediction. The energy balance for this process can be written as:

$$\begin{aligned} h_{f,CH_xO_yN_z}^0 + w h_{f,H_2O}^0 + \frac{\dot{n}_{O_2}}{\dot{n}_{CH_xO_yN_z}} h_{T_{IN},O_2} + \frac{\dot{n}_{N_2}}{\dot{n}_{CH_xO_yN_z}} h_{T_{IN},N_2} \\ = n_{H_2} h_{T_{OUT},H_2} + n_{CO} h_{T_{OUT},CO} + n_{CO_2} h_{T_{OUT},CO_2} + n_{CH_4} h_{T_{OUT},CH_4} \\ + n_{H_2O} h_{T_{OUT},H_2O} \end{aligned} \quad (4.11)$$

2679 where $h_{f,CH_xO_yN_z}^0$ and h_{f,H_2O}^0 are the formation enthalpy of organic matter and water at
 2680 reference condition (298 K and 101325 Pa) in kJ/kmol, respectively; n_i represents the
 2681 number of moles produced, for the same chemical species, per mole of consumed fuel;
 2682 lastly, $h_{T,i}$ is the enthalpy associated with the i -th chemical species at the temperature
 2683 T , and it is the sum of the formation enthalpy and the difference of enthalpy between
 2684 the state in which the substance is found and the reference status.

2685 The enthalpy of formation for organic matter, $h_{f,CH_xO_yN_z}^0$, in reactant is calculated
 2686 according to the relation suggested in [139]:

$$h_{f,CH_xO_yN_z}^0 = LHV + \sum_{k=prod} [n_k (\bar{h}_f^0)_k] \quad (4.12)$$

2687 where LHV is the lower heating value of the dry fraction of biomass estimated in
 2688 accordance with the formula reported in [139]:

$$LHV = HHV - 9 \cdot H \cdot h_{fg} \quad (4.13)$$

2689

2690 where H is the mass fraction of hydrogen in the biomass, h_{fg} is the enthalpy of
 2691 vaporization of water at reference conditions and HHV is the higher heating value
 2692 calculated on the basis of biomass composition conforming to the following equation
 2693 [156]:

$$HHV = 0.3491 \cdot C + 1.1783 \cdot H + 0.1005 \cdot S - 0.1034 \cdot O - 0.0151 \cdot N +$$

$$-0.0211 \cdot ash \quad (4.14)$$

2694 where C, H, O, N, S and ash are the mass fraction of carbon, hydrogen, oxygen,
 2695 nitrogen, sulphur and ash in the biomass.

2696 The calculation of the temperature, the chemical composition and the volumetric
 2697 flow rate of the syngas produced from biomass gasification process is based on an
 2698 iterative process. Namely, the calculation procedure is based on the following steps: i)
 2699 an average gasification temperature of the first attempt is assumed; ii) the mass
 2700 conservation and chemical equilibrium equations are solved using the Newton-Rapson
 2701 method [156] by that means obtaining the composition of the producer gas; iv) the
 2702 energy conservation equation is solved determining the new average gasification
 2703 temperature; v) the iterative method stops when convergence is achieved.

2704 **4.3.2 Results and discussion**

2705 The present study has been performed in cooperation with the research group from
 2706 University of Naples Parthenope, which developed a thermodynamic equilibrium
 2707 model using a commercial software, i.e., Aspen Plus®. In order to assess the strengths
 2708 and weaknesses of non-commercial and commercial codes, a comparison between the
 2709 results obtained from the two developed models has been carried out. In the following,
 2710 the commercial software-based model is referred as “Aspen model”, whereas the non-
 2711 commercial tool is referred as “Analytical model”.

2712 For the purpose of evaluating the accuracy of the simulation results on varying the
 2713 biomass properties and the gasification operating conditions, and evidencing
 2714 advantages and disadvantages of employing such models the developed
 2715 thermodynamic equilibrium models have been applied to different experimental data
 2716 on downdraft gasifiers available in scientific literature.

2717 In Table 4.1 and Table 4.2, the ultimate and proximate analysis of the different
 2718 types of biomasses used in the experimental campaigns and the operating conditions
 2719 of the simulated case studies have been summarized. Symbols C, H, O, N and S in
 2720 Table 4.2 represent, respectively, the carbon, hydrogen, oxygen, nitrogen and sulphur

Chapter 4 Analysis of reliability of thermodynamic equilibrium gasification models

2721 content in the biomass on dry basis (d.b.), while VM and FC are the volatile matter
2722 and fixed carbon, respectively.

2723 *Table 4.1 – Ultimate and proximate analysis of biomass used for the validation of models.*

Ref.	Biomass	Ultimate analysis [wt.% d.b.]					Proximate analysis [wt.% d.b.]		
		C	H	O	N	S	VM	FC	ASH
[158]	Hazelnut shell	46.76	5.76	45.83	0.22	0.67	71.62	27.50	0.88
[159]	Rubber wood	50.60	6.50	42.00	0.20	-	80.10	19.20	0.70
[160]	Wood Pellets	52.05	5.82	40.85	0.28	0.01	77.70	21.30	1.00
[148]	Corn cobs	46.59	5.97	44.81	0.51	-	80.06	17.82	2.12
[148]	Wood pellets	47.88	5.68	44.16	0.18	-	80.63	17.27	2.10
[148]	Rice husks	41.23	5.21	36.50	0.45	-	67.95	15.45	16.6
[148]	Vine pruning	49.51	5.67	41.35	0.86	-	80.84	16.54	2.62
[161]	Coniferous wood	49.79	5.55	43.91	0.10	0.38	-	-	0.28

2724 *Table 4.2 – Gasifier operating conditions of the experimental tests used for the validation of models.*

Ref.	Biomass	Case	Biomass feeding rate [kg/h]	Air flow rate [Nm ³ /h]	Equivalence ratio (ER)	Moisture content [wt.%]
[158]	Hazelnut shell	1	1.73	2.82	0.449	12.5
		2	2.15	3.53	0.451	12.5
		3	2.64	4.01	0.418	12.5
		4	3.19	4.40	0.380	12.5
[159]	Rubber wood	5	27.39	43.00	0.406	18.5
		6	25.27	43.00	0.425	16.0
		7	23.46	43.00	0.455	14.7
		8	28.37	43.00	0.380	16.0
[160]	Wood pellets	9	4.10	5.50	0.301	7.0
		10	6.20	9.60	0.347	7.0
		11	6.40	8.70	0.305	7.0
		12	5.90	6.70	0.254	7.0
[148]	Corn cobs	13	59.00	69.30	0.279	10.1
		14	73.00	85.00	0.277	10.1
		15	73.00	88.70	0.287	10.1
[148]	Wood pellets	16	58.00	62.30	0.272	9.5
[148]	Rice husks	17	50.00	73.90	0.382	12.5
		18	63.00	100.50	0.414	12.5

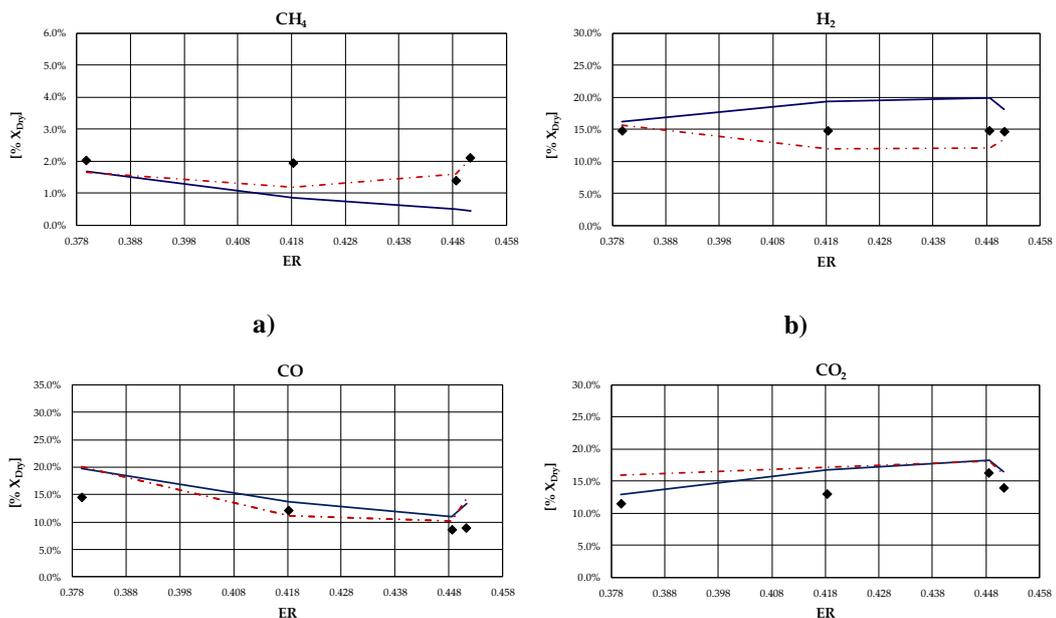
Chapter 4 Analysis of reliability of thermodynamic equilibrium gasification models

[148]	Vine pruning	19	44.00	51.10	0.257	17.6
[161]	Coniferous wood	20	0.54	0.81	0.301	7.1
		21	0.56	0.91	0.322	7.1
		22	0.60	1.06	0.337	7.1
		23	0.60	1.14	0.368	7.1

2725 The comparison between numerical and experimental results has been carried out
 2726 in terms of: i) molar composition of syngas (e.g., CH₄, H₂, CO, CO₂ and N₂) and ii)
 2727 average gasification bed temperature. The analytical model is able to predict the
 2728 average gasification bed temperature; therefore, the model has been validated also
 2729 comparing the temperature values obtained from simulation with the experimental
 2730 temperature values. The comparison has been carried out using the data reported in
 2731 [158] and [161] as these are the only works containing information on the gasification
 2732 temperature, among those considered in this paper.

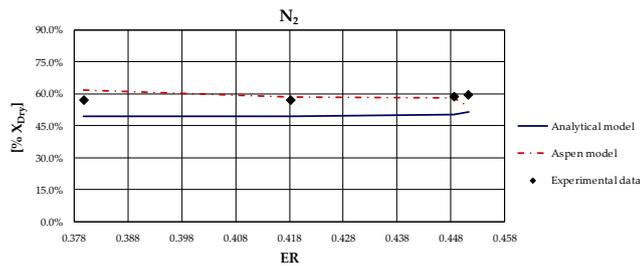
2733 Present results have been obtained calibrating both the analytical and the Aspen
 2734 models. In particular, the analytical model has been calibrated by calculating the K₁
 2735 and K₂ values that minimize the error between the experimental and the model results
 2736 in terms of gas composition and average combustion bed temperature. As regards the
 2737 Aspen model, the calibration is based on the Restricted Equilibrium Model.

2738 The results of such a comparison, as a function of the equivalence ratio (ER), for
 2739 the considered biomasses and case studies, are shown from Figure 4.1 to Figure 4.8
 2740 and in Table 4.3 and Table 4.4.



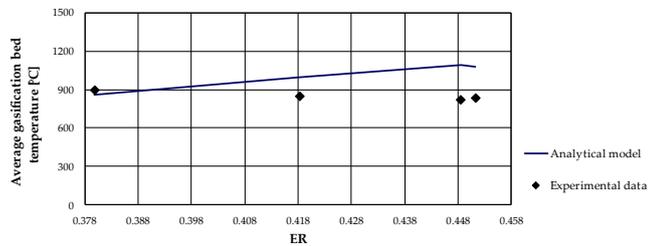
c)

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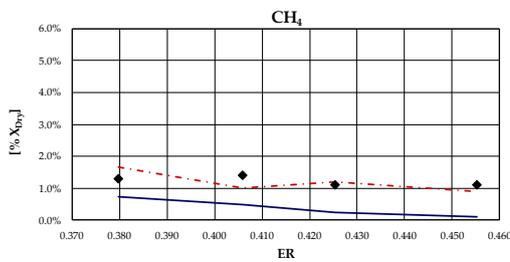
e)

2741 *Figure 4.1 – Comparison of syngas mole fraction composition between results of calibrated and not-*
 2742 *calibrated analytical and Aspen models and experimental data relating to hazelnut shell [158]: a)*
 2743 *CH₄ mole fraction; b) H₂ mole fraction; c) CO mole fraction; d) CO₂ mole fraction; e) N₂ mole*
 2744 *fraction.*

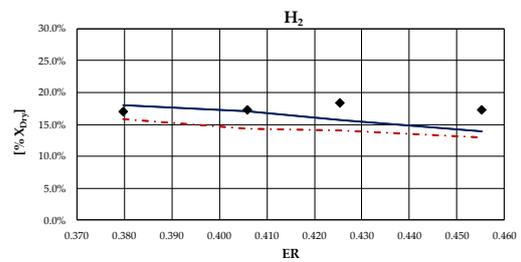


2745

2746 *Figure 4.2 – Comparison of average gasification bed temperature between the analytical model*
 2747 *results and experimental data relating to hazelnut shell [158].*

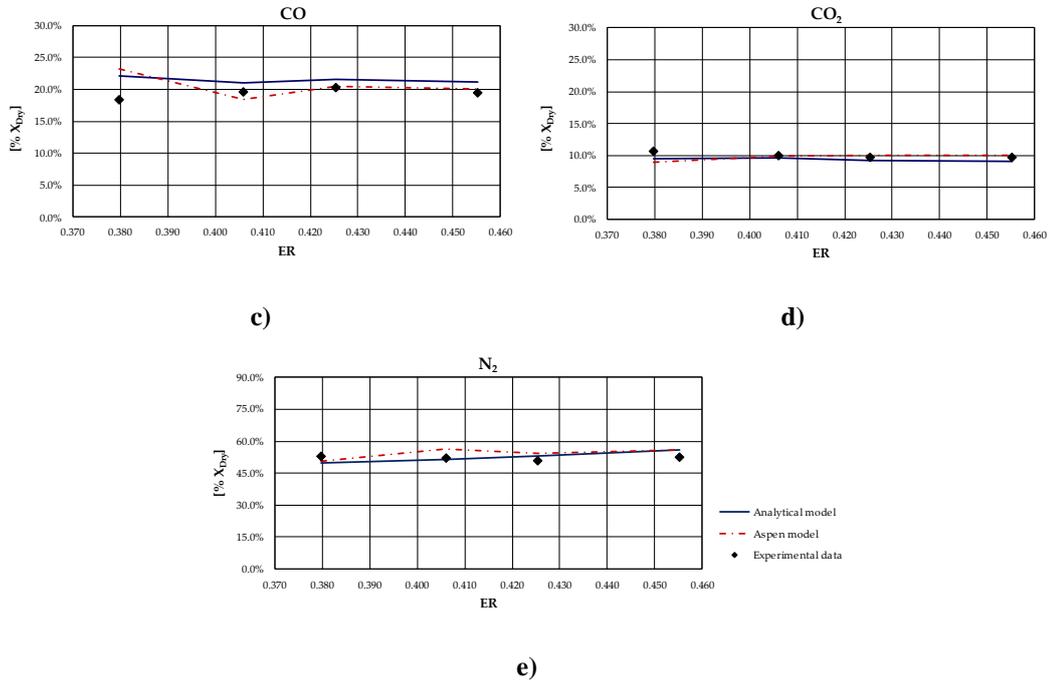


a)

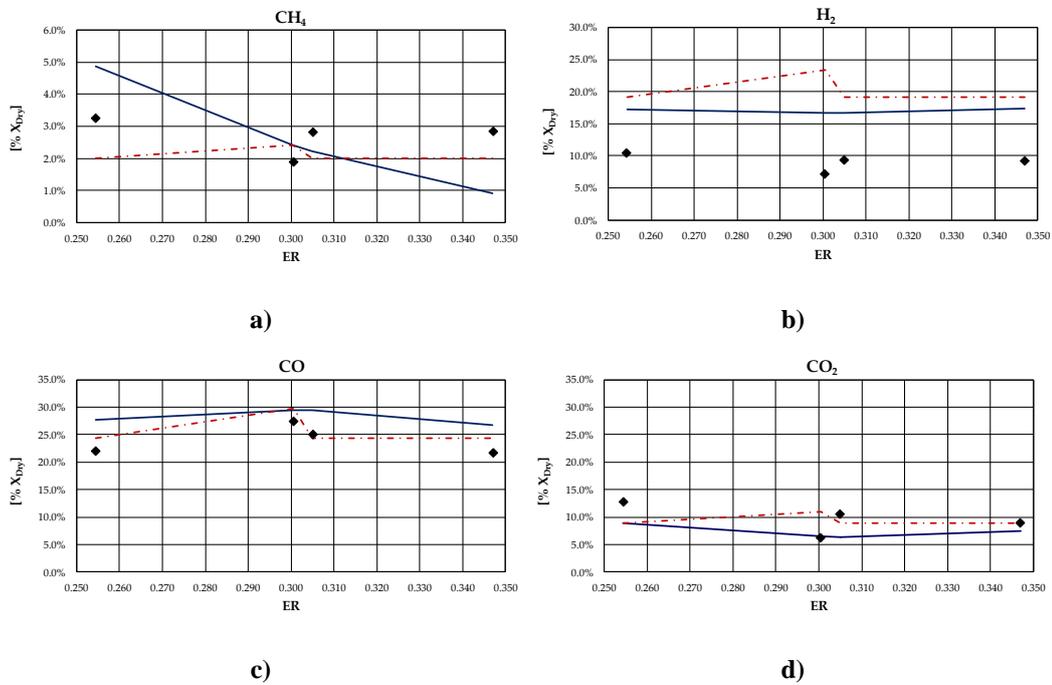


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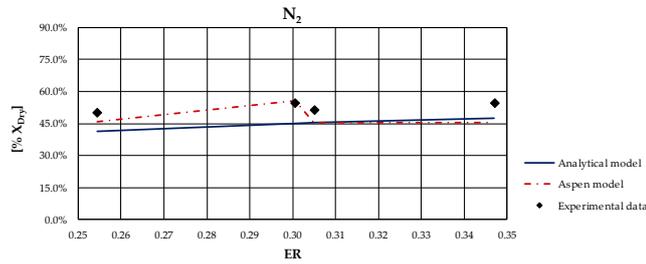
Chapter 4 Analysis of reliability of thermodynamic equilibrium gasification models



2748 *Figure 4.3 – Comparison of syngas mole fraction composition between results of calibrated and not-*
 2749 *calibrated analytical and Aspen models and experimental data relating to rubber wood [159]: a) CH₄*
 2750 *mole fraction; b) H₂ mole fraction; c) CO mole fraction; d) CO₂ mole fraction; e) N₂ mole fraction.*

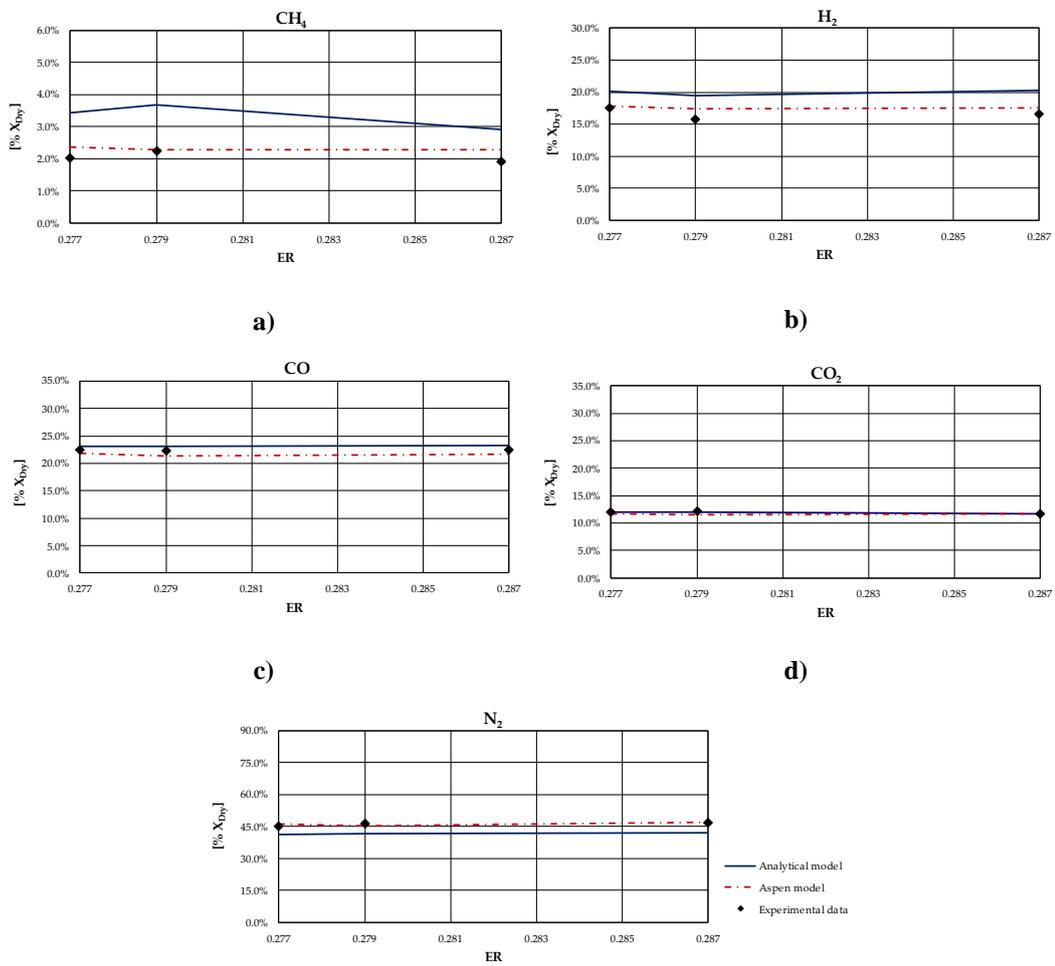


Chapter 4 Analysis of reliability of thermodynamic equilibrium gasification models



e)

2751 Figure 4.4 – Comparison of syngas mole fraction composition between results of calibrated and not-
 2752 calibrated analytical and Aspen models and experimental data relating to wood pellets [160]: a) CH₄
 2753 mole fraction; b) H₂ mole fraction; c) CO mole fraction; d) CO₂ mole fraction; e) N₂ mole fraction.



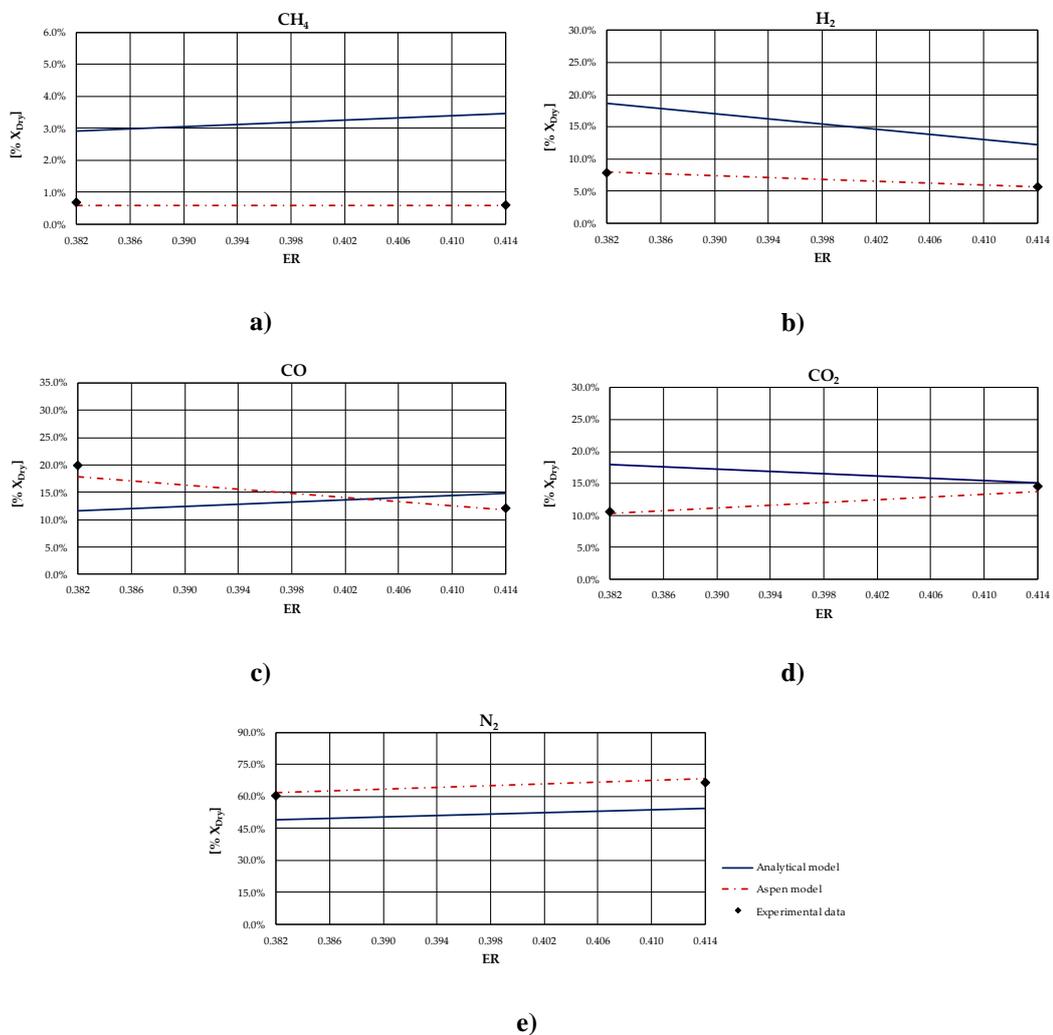
e)

2754 Figure 4.5 – Comparison of syngas mole fraction composition between results of analytical and Aspen
 2755 models and experimental data relating to corn cobs [148]: a) CH₄ mole fraction; b) H₂ mole fraction;
 2756 c) CO mole fraction; d) CO₂ mole fraction; e) N₂ mole fraction.

2757 Table 4.3 – Comparison of syngas mole fraction composition between results of analytical and Aspen
 2758 models and experimental data relating to wood pellets [148].

	Analytical model	Aspen model	Experimental data
ER	0.272		
CH ₄ [%]	3.6	2.3	2.3
H ₂ [%]	19.4	17.5	16.7
CO [%]	24.5	21.4	21.3
CO ₂ [%]	11.5	11.5	12.4
N ₂ [%]	41.0	45.4	47.3

2759

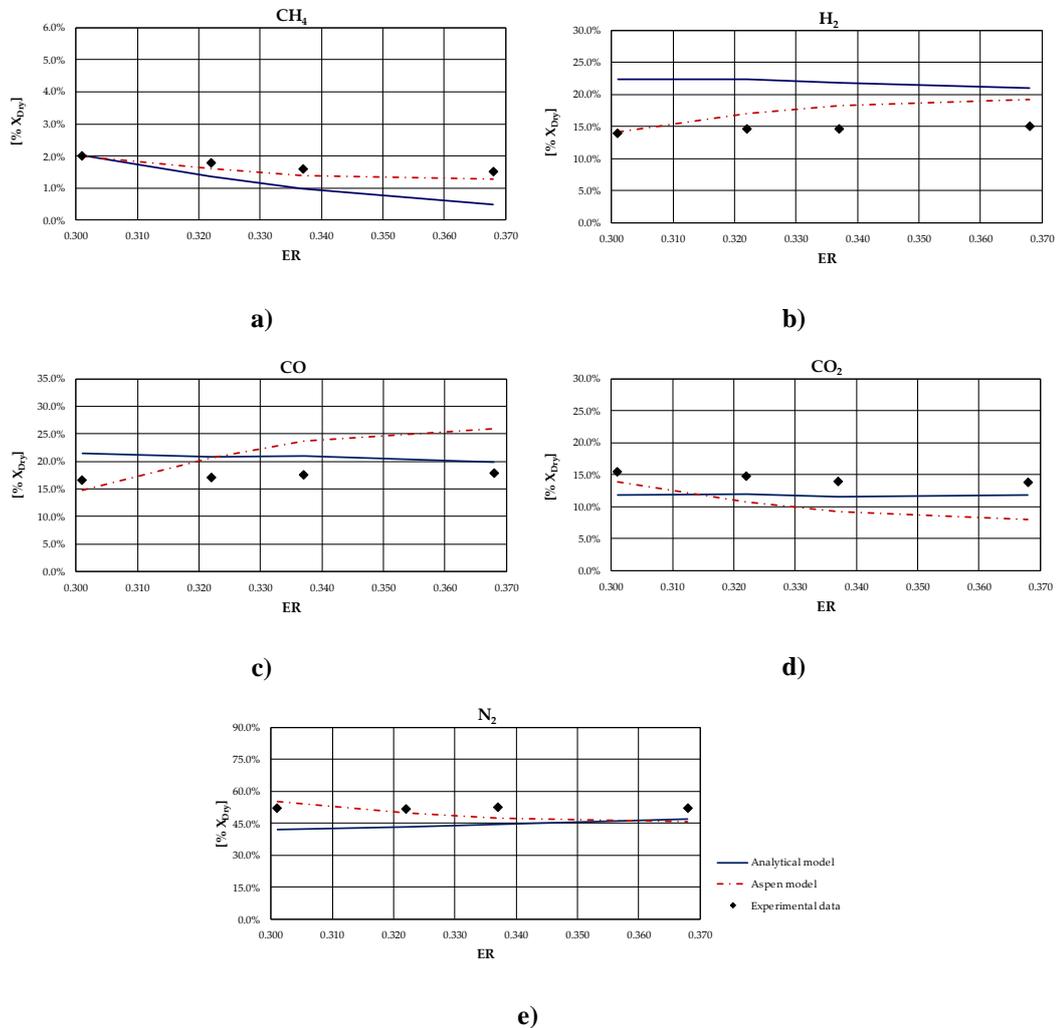


2760 Figure 4.6 – Comparison of syngas mole fraction composition between results of calibrated and not-
 2761 calibrated analytical and Aspen models and experimental data relating to rice husks [148]: a) CH₄
 2762 mole fraction; b) H₂ mole fraction; c) CO mole fraction; d) CO₂ mole fraction; e) N₂ mole fraction.

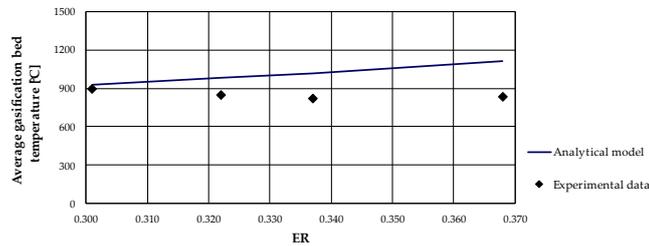
2763 Table 4.4 – Comparison of syngas mole fraction composition between results of analytical and Aspen
 2764 models and experimental data reported in [148].

	Analytical model	Aspen model	Experimental data
ER	0.257		
CH ₄ (%)	3.8	2.7	2.6
H ₂ (%)	21.8	18.3	17.1
CO (%)	23.1	23.4	21.7
CO ₂ (%)	12.0	10.8	13.3
N ₂ (%)	39.3	44.7	45.1

2765



2766 Figure 4.7 – Comparison of syngas mole fraction composition between results of analytical and Aspen
 2767 models and experimental data relating to coniferous wood [161]: a) CH₄ mole fraction; b) H₂ mole
 2768 fraction; c) CO mole fraction; d) CO₂ mole fraction; e) N₂ mole fraction.



2769

2770 *Figure 4.8 – Comparison of average gasification bed temperature between the analytical model*2771 *results and experimental data relating to coniferous wood [161].*

2772 From the analysis of figures and tables, it can be observed that even though the
 2773 proposed models are both based on the thermodynamic equilibrium hypothesis,
 2774 significant differences can be observed in obtained results. In particular, CH₄ and H₂
 2775 molar fractions evaluated using the analytical model are characterized by a higher
 2776 deviation from experimental values compared to the other synthesis gas components.
 2777 In detail, CH₄ molar concentration is underestimated for the case studies in which the
 2778 ER is higher. Conversely, it is overestimated for the case studies in which the ER
 2779 is lower. Furthermore, CH₄ content decreases as ER increases for all types of biomasses,
 2780 except for rice husks. The analytical model generally overestimates the H₂ molar
 2781 fraction regardless of the biomass feedstock and operating condition of gasification
 2782 process. A good agreement between the results of the analytical model and
 2783 experimental values can be observed as regards the CO and CO₂ molar concentrations
 2784 for all case studies analysed. The N₂ content tends to be underestimated by the
 2785 analytical model for each simulated case, probably because of the restrictive
 2786 hypothesis of complete carbon conversion on which the model is based. As regards
 2787 the Aspen model, a generally satisfactory agreement between the numerical results and
 2788 the experimental data can be observed, except for the campaign carried out in [160]
 2789 for which the trend of the fuel compounds, CH₄ and H₂, is not adequately captured.
 2790 Ignoring this last group of results, the highest average deviation between the
 2791 numerical and the experimental data is 23%, obtained for the first case of the campaign
 2792 carried out in [158]. More in detail, the concentration of CH₄, which slightly decreases
 2793 with the ER, is well predicted by the model. Such a result is promising since usually
 2794 the compounds characterized by the lowest concentrations, such as methane, are the
 2795 ones giving the highest deviations between numerical and experimental results. Also
 2796 the trend of H₂ concentration, which slightly varies with the ER, is fairly predicted by
 2797 the model, except for the campaign carried out in [159] and shown in Figure 4.3. This
 2798 could be related to the variation of the moisture content that influences the H₂

2799 concentration in the syngas. The concentrations of CO and CO₂ usually show an
2800 inverse trend due to the water shift reaction in which they are involved. Their trend, as
2801 well as that of N₂, is in excellent agreement with the experimental results. Obviously,
2802 the higher the ER the higher the dilution and thus the N₂ concentration.

2803 As expected, obtained results evidence that the major limitation of both models
2804 relies in the thermodynamic equilibrium assumption and in neglecting tar and char. In
2805 fact, the results show that syngas composition is better reproduced for the biomass
2806 types characterized by a low ash content and consequently, a relatively high heating
2807 value. Indeed, the major deviations from experimental data have been obtained for
2808 cases of gasification of rice husks, whose ash content is about 16.6 % on dry basis, as
2809 can be noted in Figure 4.6. This can be ascribed to the fact that the model does not
2810 simulate the tar and char formation that in this case are a considerable part of the
2811 products of the gasification process.

2812 Concerning the evaluation of average gasification bed temperature, as expected, as
2813 the ER increases the temperature increases, as can be observed from Figure 4.2 and
2814 Figure 4.8. Since the gasifier is assumed to be adiabatic, analytical model generally
2815 overestimates the gasification temperature and the deviation between numerical results
2816 and experiments increases as the ER increases.

2817 In general, it can be seen how using a more advanced model of Restricted
2818 Equilibrium allows to obtain results closer to the experimental ones, minimizing the
2819 error between simulated and experimental data, especially for the concentrations of
2820 CH₄ and H₂ which being very low values are difficult to model. Nevertheless, in the
2821 case of absence of necessary information for the calibration, the use of the non-
2822 calibrated Aspen model produces non reliable results. Moreover, it does not predict
2823 the gasification temperature. On the contrary, the analytical model is more general and
2824 the calibration process allows to find out the optimal multiplying coefficients of K₁
2825 and K₂ constants to be used as a function of the biomass type and ash content. Besides,
2826 the calculation of the gasification temperature is performed by solving the energy
2827 conservation equation.

2828 In summary, the limit of this approach occurs in critical conditions in which the
2829 thermodynamic equilibrium is not likely achieved and biomass is characterized by
2830 high humidity content, ashes, tar and char.

2831 In the future, the simulation of tar and char formation will be included in the model.
2832 In addition, the model will be implemented in a more complex modelling of
2833 polygenerative systems based on renewable sources with the aim of optimizing these
2834 systems from an energy and economic point of view.

2835 Finally, since it based on a non-commercial home-made code, future improvements
2836 in its ability to predict methane and hydrogen concentration can be easily
2837 implemented.
2838

CONCLUSIONS

In this work the complex issue of innovative use of the existing NG infrastructure, as well as the development of innovative technologies on natural gas transportation and distribution networks, has been address. The subject has been treated in a multidisciplinary manner, allowing the reader to have a broader view of the problem in its different facets. To this aim, numerous research experiences have been reported in the present thesis as case studies.

In *Chapter 1*, the subject of NG networks management in the event of fault occurrence is tackled employing the concept of resilience. Based on the literature review, one of challenges integral to assess the resilience of complex infrastructural systems, such as NG pipeline systems, is the development a holistic methodology able to take into account multiple factors, such as geometrical and topographic characteristic of the system, type of the fault, consumption scenario, recovery strategy etc., that impact on the state of service resulting from an unpredictable event.

To overcome such gap, the author proposed a novel resilience assessment methodology for acquiring, simulating, and processing the impact of failures on NG distribution networks (*case study #1*), with the aim to evaluate the behaviour and fragility of distribution networks, under different fault scenarios and network structures, as well as to evaluate the fault management strategies to minimize the disservice and to define possible structural improvement measures. In detail, the author proposed a resilience assessment methodology based on network simulation to improve the resilience of NG distribution networks through: i) the estimation of the impact of a generic fault in the network in terms of quality of the service and potential number of disrupted citizens; ii) the identification of possible critical areas in which users are more susceptible to disruptions; iii) the development of operational

management strategies; iv) the selective assessment of structural improvements in the network to reduce the risk of fault and improve the resilience of the network itself.

The proposed methodology has been experimented in a real case study of a natural gas distribution network in southern Italy showing the effectiveness of this approach capable of avoiding or minimizing the risk of service interruption, through:

- i. the identification of the type of fault and the operating conditions with greatest impact;
- ii. the analysis of hourly trend of the number of disrupted users/nodes and the appropriate programming of the restoration interventions;
- iii. the development of targeted users breakdown as applicable recovery strategy (e.g., aiming at the disruption of a limited number of users that can be more easily reactivated after the fault has been repaired).

This approach may in the future be increasingly effective thanks to the greater availability of smart meters. These tools will allow distribution companies to better profile user consumptions on the basis of actual data from smart meters and to build increasingly precise and reliable structures and operating models for the network.

In *Chapter 2*, the author investigated the technical feasibility of the PtG concept to store intermittent RESs through the analysis and discussion of two different case studies.

In *case study #2*, a PtSNG plant coupled with a 12 MW wind farm has been designed and investigated by evaluating nominal and annual efficiencies for different sizes of the electrolysis unit. The analysed plant consists of four sections: (i) the electrolysis unit, (ii) the CO₂ methanation unit, (iii) the hydrogen storage unit, and (iv) the SNG upgrading unit. The main design specifications of the PtSNG plant regard the SNG composition (CH₄>95%, H₂≤5%, and CO₂ balance), and the capacity of the storage unit (minimum size to assure about 6.5 h of full load for the methanation unit with a small number of shutdowns and ON/OFF cycles). In order to define the plant operating conditions and the plant sizing in terms of mass and energy flows, an integrated thermochemical and electrochemical model has been developed. This model has allowed to assess the energy and mass balances as well as the performances

of each plant section in steady state conditions (i.e., full or partial load). The dynamic operation of the PtSNG plant has been investigated by developing a dynamic model that assess the mass and energy fluxes between the plant sections by following the control strategy of the storage unit, conceived to reduce the shutdowns and the ON/OFF cycles of the methanation unit with the minimum storage capacity. The findings obtained by applying the integrated steady state model and the dynamic model are:

- i. the nominal overall efficiency, referred to the LHV and calculated by considering only the SNG as useful product, is equal to 44% for all the three plant sizes considered; in the full load operation, the SNG production is equal to 34.1, 102.4, and 204.9 kg/h, respectively;
- ii. the rectifier efficiency (the efficiency of the AC/DC transformer) and the electrolysis unit efficiency mainly impact on the overall plant efficiency; thus, using more efficient devices (i.e., by assuming 98% and 70%), the overall nominal efficiency is 58.1%;
- iii. in order to increase the plant performances, the heat generated during the methanation operation should be valorised using thermal storage or in power generation systems; to this end, further investigations and improvement of the developed models are required;
- iv. the operation time diminishes with the plant capacity, reaching the greatest value (6047 h) for the 1 MW size configuration that shows also the smallest number of shutdowns (212);
- v. the annual overall efficiency as well as the plant load factor achieves the greatest values (43.7%, LHV basis and 60%) for the 1 MW size configuration. The annual SNG production is equal to 179,000 Nm³/year.

In *case study #3*, two heat recovery systems for a 1 MW (the DC wind power to the electrolysis unit) PtSNG plant coupled with a 12 MW wind farm have been presented. The common approach is based on the valorisation of the waste heat generated during the plant operation to self-sustain, from an energetic point of view, the standby conditions (hot-standby and cold-standby). In the first one (thermal storage configuration), the recovered thermal energy is used to satisfy the thermal energy

demand of the reactors during the standby periods by means of a thermal carrier (hydrogen). In the second one (electrical storage configuration), the stored thermal energy is used to produce electricity by means of an Organic Rankine Cycle (ORC) where iso-butane is used as the working fluid. This electrical energy is then stored in batteries, and it is used to meet the energy requirements of the electrical trace heating system of the methanation unit during the standby periods. The analysis has been performed by means of thermochemical and electrochemical models developed in the Aspen Plus environment to define the mass and energy balances in steady-state conditions and integrated with a dynamic model built in the Matlab language to forecast the plant operation time and the annual performance. The main constraint for the thermal recovery systems sizing is the minimisation of the storage devices (oil tanks for the thermal storage configuration or batteries for the electrical storage configuration) achieved by identifying the optimum combination between the hot- and cold-standby hours. To this end, statistical analysis of the plant standby hours distribution in terms of the average duration and frequency has been carried out. For the specific energy input profile, the maximum duration of the hot standby has been found equal to 19 h. Results of the thermal storage configuration have shown that the hydrogen flow rate needed to maintain the methanation reactors at the minimum catalyst temperature (250 °C) is equal to 6.5 kg/h at 367 °C, while the hot oil flow rate required to heat the hydrogen is equal to 31.5 kg/h and is stored at 387 °C. Results of the electric storage configuration have shown that the electric power generated is equal to 24 kW and the ORC cycle efficiency is 16.6%. The battery is sized for a capacity of 80 kWh, guaranteeing 19 h of hot standby and a cold restart by means of electrical trace heating. When the methanation unit is in operation and the ORC produces electrical energy, part of this is used for plant auxiliaries (12.3 kWh at full-load operation for 1 h), so the electrical energy useful to charge the battery is equal to 11.8 kWh, resulting in a battery charge time of 6.8 h. The overall nominal efficiencies obtained for the PtSNG plant are equal to 43.9% and 48.8% on LHV and HHV bases, respectively. When heat recovery systems are coupled to the plant, the annual efficiencies on an LHV basis have been found to be 44.0% and 44.3% for the thermal storage and electrical storage configurations, respectively. The results showed that the

internal heat recovery system positively impacted the annual efficiency. It is expected that this could lead to a decrease in the plant operational costs with an associated relatively low initial capital expenditure. Furthermore, the amount of waste heat recoverable from the methanation unit resulted in being significantly higher than the energy needed for the self-sustainment of the plant. Therefore, to achieve a considerable increase in the efficiency of the plant, it is necessary to couple the plant with other thermal or electrical utilises (e.g., district heating, cogeneration of SNG and power, etc.). In the event that a decision is made to carry out cogeneration of electricity to sell it to the grid, this could also be produced with more efficient systems than the ORC (e.g., steam turbine), but careful cost assessments are required both on initial investments and on possible revenues.

It has to be underlined that the results presented both in *case study #1* and *case study #2* are valid only for the specific hypotheses and technologies adopted, as well as for the specific profile of the electrical input and the chosen plant management strategy.

In *Chapter 3*, an overview of the effect of hydrogen injection on NG networks has been presented. As *case study #4*, the analysis of impact of hydrogen injection on the reliability of measuring devices that currently run on the existing NG infrastructures has been proposed.

In particular, the changes in thermodynamic properties of the gas mixtures with different H₂ contents (i.e., up to 25% vol) have been evaluated and the effects on the accuracy of volume conversion at standard conditions have been investigated. Moreover, the author has been presented and discussed the effect of H₂ injection in gas networks on static ultrasonic domestic gas meters, both from a theoretical and an experimental point of view. Finally, the impacts of hydrogen injection on NG quality measurements have been discussed from a theoretical point of view.

The results obtained show that injection of H₂ into natural gas significantly impacts the relative density, specific heat and higher calorific value and speed of sound (which is a critical parameter for ultrasonic static gas meters), while the Wobbe Index is less affected. In particular:

CONCLUSIONS

- i. the relative density decreases as the H₂ content increases (e.g., for the investigated GAS 6 10%vol of H₂ is sufficient for the corresponding limit to be exceeded);
- ii. the higher calorific value significantly decreases as the H₂ content increases (e.g., for gases characterized by a low content of CH₄, the calorific value is reduced by 20% when the hydrogen content is equal to 25%, exceeding the lower limit indicated by the ISO 12213-2 standard;
- iii. the speed of sound increases up to 12.3% becoming generally higher than the accepted limit of 475 m·s⁻¹ indicated by ISO 14236 for ultrasonic gas meters already at x_{H2} = 5%.

The trend of the compressibility factor Z as a function of H₂ injection was also analysed using the available calculation algorithms of ISO 12213-2 (complete composition), ISO 12213-3, AGA NX 19 and AGA NX 19 Mod. The obtained results show that the compressibility factor at high pressures is more influenced by the presence of hydrogen than at low pressures (e.g., Z increases in the range 0.10–0.14% at 5 bar and in the range 2.1–9.1% at 70 bar). Moreover, the pressure influence on Z is more significantly impactful at low H₂ contents. A similar behaviour has been found for the volume conversion factor.

In the end, no influence on the deterioration of metrological properties has been found for domestic G4 ultrasonic gas meters tested with the use of various gas mixtures with hydrogen content up to 10%. In fact, both the errors of indications and the weighted mean errors remained within the permissible limits, with measured errors not exceeding $\pm 1\%$ and WME within $\pm 0.4\%$.

In *Chapter 4*, the author presents an overview of biomass gasification modeling approaches with the aim to evaluate their effectiveness as a modeling tool for the design and optimization of polygeneration plants based on biomass gasification. In fact, the necessity to build plant operating maps for efficiency optimization requires a significant number of simulations and thermodynamic equilibrium models may allow fast computations thanks to their relative simplicity. The main objective consists in the assessment of thermodynamic equilibrium models as a function of biomass type and

CONCLUSIONS

composition to better understand in which conditions of practical interest such models can be applied with acceptable reliability. To this aim, the author proposed an equilibrium model using a simulation tool implemented in a non-commercial script as *case study #5*. The present study has been performed in cooperation with the research group from University of Naples Parthenope, which developed a thermodynamic equilibrium model using a commercial software, i.e., Aspen Plus®. In order to assess the strengths and weaknesses of non-commercial and commercial codes, a comparison between the results obtained from the two developed models has been carried out.

The developed thermodynamic equilibrium models have been applied to different case studies on downdraft gasifiers available in the literature in order to evaluate the accuracy of the simulation results on varying the biomass type and composition and the operating conditions of the gasification process, also evidencing advances and disadvantages of both models. Obtained results have been compared with experiments in terms of: i) molar composition of syngas and ii) average gasification bed temperature.

In general, the simulation results are comparable with the experimental data for both models. Namely, the results show that the analytical model predicts the syngas composition with better accuracy for the biomass types characterized by a low ash content and consequently, a relatively high heating value. This can be ascribed to the fact that the model does not simulate the tar and char formation. As regards the Aspen model, it appears to fairly predict the syngas composition at different conditions of ER. However, if the properties of the treated biomass change, the accuracy of the model may be reduced. Even though CH₄ and H₂ syngas concentrations are typically very low and difficult to predict, values predicted by the Aspen model are in satisfactory agreement with experiments.

Finally, the developed models offer several advantages: i) not require details of system geometry or estimate the necessary time to reach the thermodynamic equilibrium; ii) employ a limited number of input data; iii) lack of need of employing correlation for the calculation of the biomass gasification temperature. In addition, the adoption of a general model also allows to obtain quickly results in the case it is necessary to simulate numerous operating conditions.

APPENDIX A

A.1 Methanation Unit Modelling

In order to validate the proposed TREMP model, shown in Figure A.1, a comparison of experimental data reported in [66] with the simulated results using the present model was performed. The experimental data refers to results obtained by experiments on high-temperature methanation carried out in the bench-scale unit ADAM I [162]. The methanation plant ADAM I consisted of three adiabatic fixed bed methanation reactors including recycle according to the TREMP process. As shown in Table A.1, a good agreement is observed between the experimental data and results predicted from the proposed model. The feeding gas flow rate is 535 Nm³/h (composition is 65.45% mol H₂, 9.84% mol CO, 8.96% mol CO₂, 11.3% mol CH₄), at 300 °C and 27.3 bar.

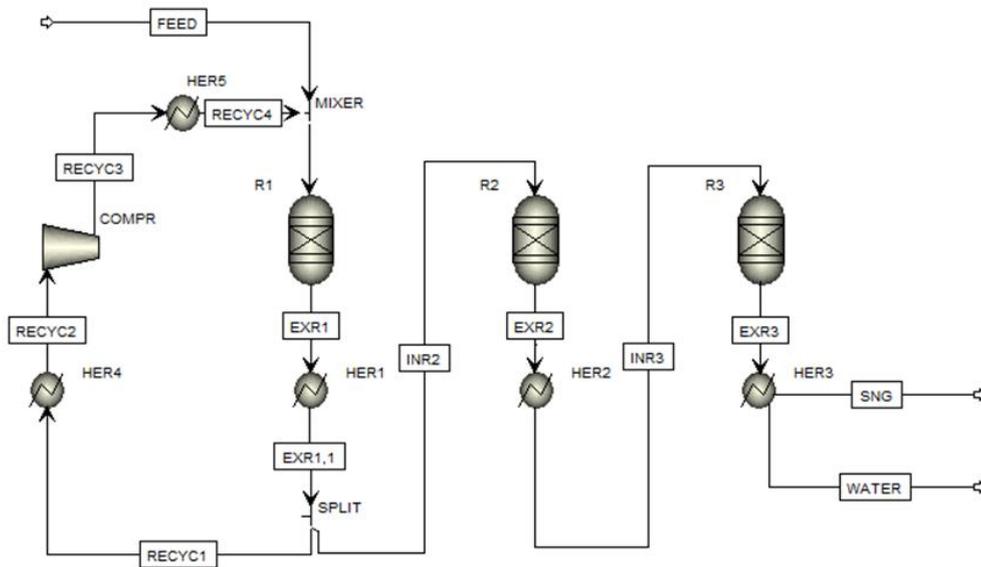


Figure A.1 – Flowsheet of the methanation process model.

Table A.1 – Simulation results compared with experimental data of ADAM I.

	R1 Inlet		R1 Exit		R2 Exit		R3 Exit		SNG	
	[30]	model	[30]	model	[30]	model	[30]	model	[30]	model

APPENDIX A

Temperature [°C]	300	299	604	595	451	451	303	302	23	23
Pressure [bar]	27.2	27.3	27.1	27.1	27.05	27.05	27	27	27	27
Flow rate [Nm ³ /h]	1416	1416.0	1255	1252.0	348	346.6	334	336.2	191	190.4
Gas composition [vol. %]										
H ₂	36.88	36.74	20.96	19.29	8.10	7.50	1.77	1.63	3.11	2.88
CO	4.28	4.30	1.17	0.94	0.00	0.05	0.00	0.00	0.00	0.00
CO ₂	6.13	5.93	4.46	4.09	2.07	1.81	0.95	0.37	1.67	0.66
CH ₄	28.12	28.16	37.44	38.40	44.36	44.64	47.28	47.55	82.95	83.96
H ₂ O	19.19	19.25	29.82	30.94	38.84	39.21	43.06	43.45	0.10	0.14
N ₂	5.41	5.61	6.15	6.35	6.64	6.80	6.93	7.00	12.16	12.37

A.1 The storage control strategy flow chart

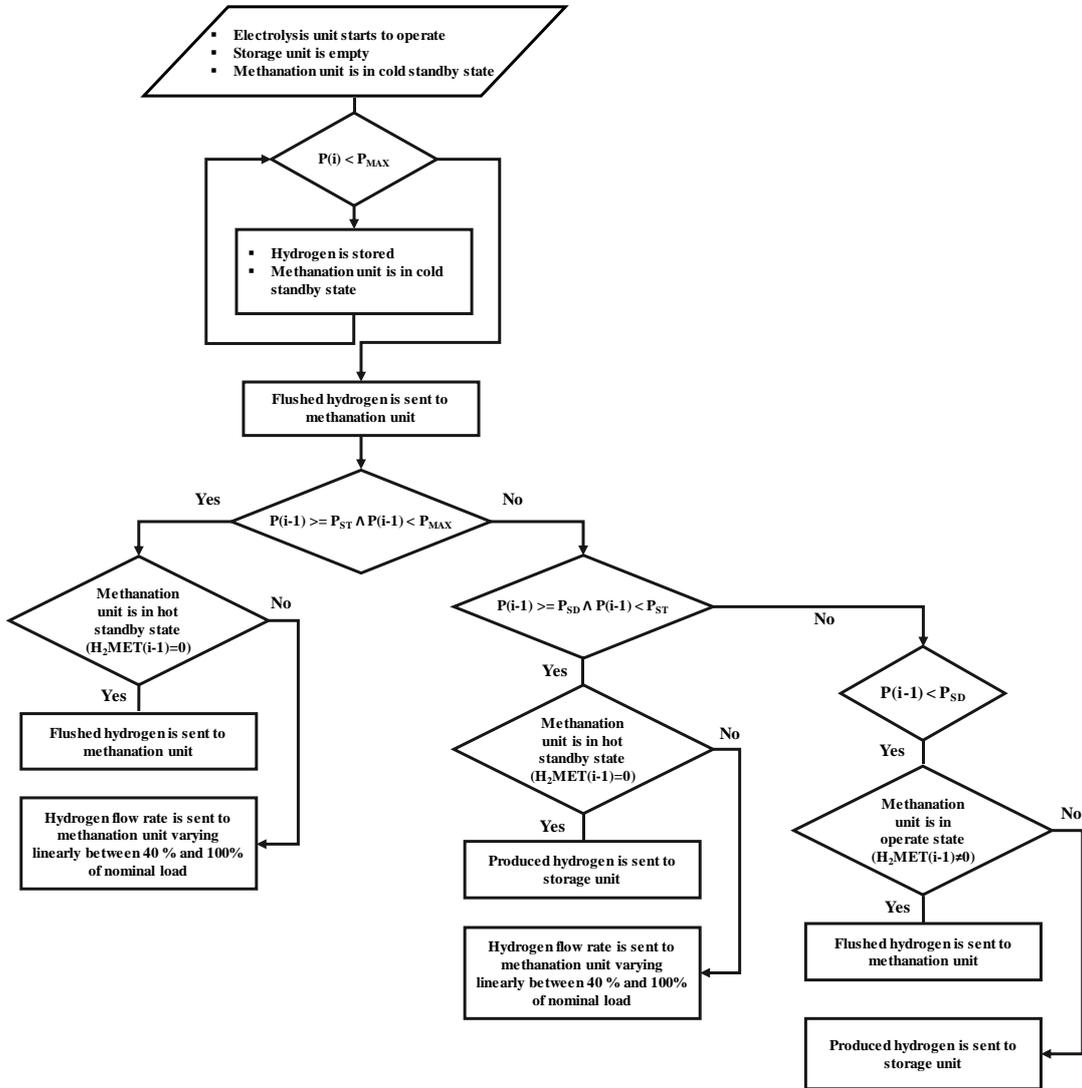


Figure A.2 – Flowchart of the storage control strategy.

APPENDIX B

Methanation Unit Modelling

The methanation unit has been modelled by using Aspen Plus software. The flowsheet of the model is shown in Figure B.0.1. Adiabatic Plug Flow Reactor (PFR) blocks have been used to simulate the methanation process. The thermodynamic and transport properties of the components are calculated using the RKSMHV2 property method. This property method, based on the Redlich–Kwong–Soave equation of state with modified Huron–Vidal mixing rules, is used for mixtures of non-polar and polar compounds, in combination with light gases.

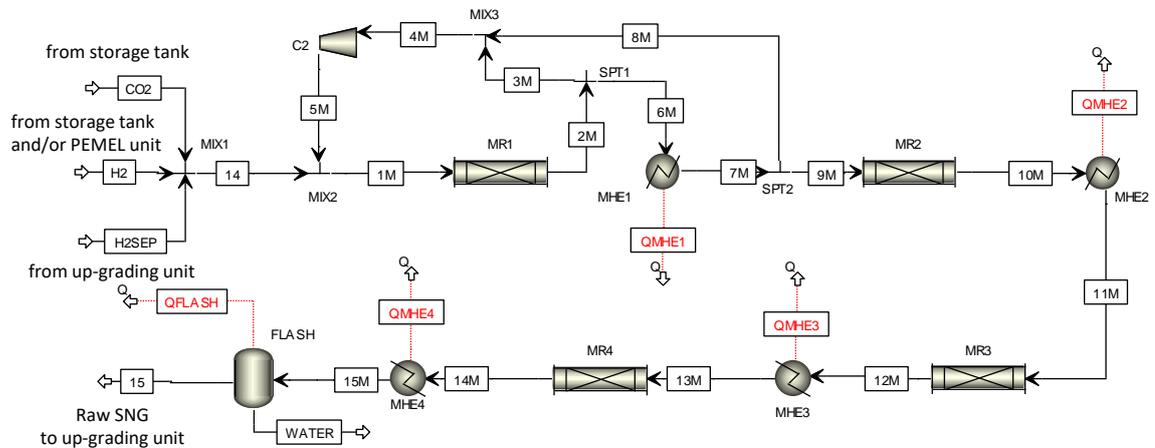
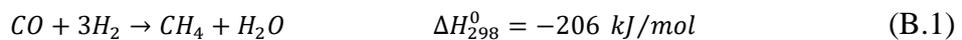


Figure B.0.1 – Flowsheet of CO₂ methanation unit.

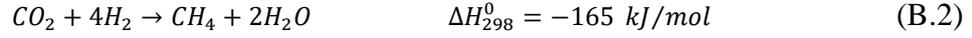
The kinetic model developed by Xu and Froment has been applied to calculate the catalytic reaction rates as in [72].

The following three reaction are considered:

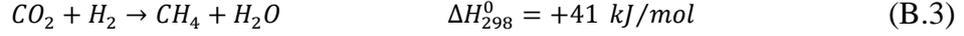
CO methanation:



CO₂ methanation:



Water gas shift:



The corresponding rate equations are given by the equations below, whose kinetic parameter values are summarized in Table A1.

CO methanation:

$$r_1 = \frac{k_1}{p_{H_2}^{2.5}} \cdot \frac{p_{CH_4} \cdot p_{H_2O} - \frac{p_{H_2}^3 \cdot p_{CO}}{K_1}}{DEN^2} \quad (\text{B.4})$$

CO₂ methanation:

$$r_2 = \frac{k_2}{p_{H_2}^{3.5}} \cdot \frac{p_{CO} \cdot p_{H_2O}^2 - \frac{p_{H_2}^4 \cdot p_{CO_2}}{K_2}}{DEN^2} \quad (\text{B.5})$$

Water gas shift:

$$r_3 = \frac{k_3}{p_{H_2}} \cdot \frac{p_{CO} \cdot p_{H_2O} - \frac{p_{H_2} \cdot p_{CO_2}}{K_3}}{DEN^2} \quad (\text{B.6})$$

where:

$$DEN = 1 + K_{CO}p_{CO} + K_{H_2}p_{H_2} + K_{CH_4}p_{CH_4} + \frac{K_{H_2O}p_{H_2O}}{p_{H_2}} \quad (\text{B.7})$$

The pre-exponential factors $A(k_i)$ and $A(K_j)$ are determined from $k_{i,T}$ and $K_{j,T}$, while E_i and ΔH_j values can be calculated by the Arrhenius and van't Hoff equations:

$$k_i = A(k_{i,T}) \cdot \exp\left(\frac{-E_i}{RT}\right) \quad (\text{B.8})$$

$$K_i = A(K_{i,T}) \cdot \exp\left(\frac{-\Delta H_i}{RT}\right) \quad (\text{B.9})$$

$$K_j = A(K_{j,T}) \cdot \exp\left(\frac{-\Delta H_j}{RT}\right) \quad (\text{B.10})$$

$$A(k_i) = k_{i,T} \cdot \exp\left(\frac{E_i}{RT}\right) \quad (\text{B.11})$$

$$A(K_j) = K_{j,T} \cdot \exp\left(\frac{\Delta H_j}{RT}\right) \quad (\text{B.12})$$

where:

- $i = 1, 2, 3$ and $j = \text{CH}_4, \text{H}_2\text{O}, \text{CO}, \text{H}_2$;
- k_1, k_2, k_3 are the rate coefficient of Eqs. (B.5) – (B.7), respectively;
- K_1, K_2, K_3 are the rate coefficient of Eqs. (B.5) – (B.7), respectively;
- $K_{\text{CH}_4}, K_{\text{CO}}, K_{\text{H}_2}$ are the adsorption constants of reaction for CH_4, CO and H_2 ;
- $K_{\text{H}_2\text{O}}$ is the desorption constant of H_2O ;
- $A(k_i)$ is pre-exponential factor of the rate coefficient, k_i ;
- $A(K_j)$ is pre-exponential factor of the rate coefficient, K_j ;
- E_i is the activation energy of reaction i -th;
- ΔH_j is the enthalpy change of reaction or adsorption;
- p is the partial pressure.

Table B.0.1 – Parameters of the reaction rate by Xu and Froment [163].

Parameter	Value
$A(k_1)$ [kmol bar ¹ kg _{cat} ⁻¹ h ⁻¹]	$4.225 \cdot 10^{15}$
$A(k_2)$ [kmol bar ^{1/2} kg _{cat} ⁻¹ h ⁻¹]	$1.020 \cdot 10^{15}$
$A(k_3)$ [kmol bar ^{1/2} kg _{cat} ⁻¹ h ⁻¹]	$1.955 \cdot 10^6$
$A(K_{\text{CH}_4})$ [bar ⁻¹]	$6.65 \cdot 10^{-4}$
$A(K_{\text{H}_2})$ [bar ⁻¹]	$6.12 \cdot 10^{-9}$
$A(K_{\text{CO}})$, [bar ⁻¹]	$8.23 \cdot 10^{-5}$
$A(K_{\text{H}_2\text{O}})$	$1.77 \cdot 10^5$
$A(K_1)$ [bar ²]	$5.375 \cdot 10^{10}$
$A(K_2)$ [bar ²]	$4.707 \cdot 10^{12}$
$A(K_3)$	$1.142 \cdot 10^{-2}$
E_1 [kJ mol ⁻¹]	240.1
E_2 [kJ mol ⁻¹]	243.9
E_3 [kJ mol ⁻¹]	67.13
ΔH_1 [kJ mol ⁻¹]	-206.1
ΔH_2 [kJ mol ⁻¹]	-165.0
ΔH_3 [kJ mol ⁻¹]	+41.15
ΔH_{CH_4} [kJ mol ⁻¹]	-38.28
ΔH_{H_2} [kJ mol ⁻¹]	-82.90
$\Delta H_{\text{H}_2\text{O}}$ [kJ mol ⁻¹]	88.68
ΔH_{CO} [kJ mol ⁻¹]	-70.65

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